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NORTHERN GRAPHITE CORPORATION



Technical Report

Title:	OKANJANDE GRAPHITE PROJECT Preliminary Economic Assessment Study Update Report
ISSUED to:	Northern Graphite
Location:	Namibia, Otjozondupa Region
DATE:	31/07/2023
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LEAD CONSULTANT

Name	Project Role	Company	Date	Signature
Etienne Roux	Lead Consultant	CREO Engineering Solutions	31/07/2023	

PROJECT OWNER

Name	Project Role	Company	Date	Signature
Kirsty Liddicoat	Client	Northern Graphite Corporation	31/07/2023	
Hugues Jacquemin	Project Sponsor	Northern Graphite Corporation	31/07/2023	

QUALIFIED PERSONS SIGNATURES

This report entitled “OKANJANDE GRAPHITE PROJECT - Preliminary Economic Assessment Study Update Report”, of February 1, 2023 and dated June 1, 2023 was prepared and signed by the following authors:

Competent Person	Company	Date	Signature
Etienne Roux	CES 	31/07/2023	
Robert Barnett	MSA 	31/07/2023	
Ipelo Gasela	MSA 	31/07/2023	
Mark Mohring	MSA 	31/07/2023	

CERTIFICATE OF QUALIFIED PERSON

Etienne Roux

I, Etienne Roux, Director at Creo Engineering Solutions and SME Registered Member, of Windhoek, Namibia, as an author of the technical report entitled “OKANJANDE GRAPHITE PROJECT: Preliminary Economic Assessment Study Update Report, Otjozondupa Region, Namibia” (the “PEA Report”) with an effective date of 30 June 2023 and a report date of 31 July, 2023 prepared for Northern Graphite Corporation (the “Issuer”), do hereby certify:

- 1) I am currently employed as a Director at Creo Engineering Solutions Pty Ltd, 1551 Sam Nujoma Drive, Tsumeb, Namibia.
- 2) This certificate applies to the PEA report titled “OKANJANDE GRAPHITE PROJECT: Preliminary Economic Assessment Study Update Report, Otjozondupa Region, Namibia”, NI 43-101 PEA Report, that has an effective date of 30 June 2023 and a report date of 31 July 2023 (the PEA Report).
- 3) I graduated with a Bachelor of Science degree in Chemical Engineering in 1999 from the University of Stellenbosch, South Africa.
- 4) I am a Registered Member (#04155055) of the Society of Mining, Metallurgy and Exploration (SME).
- 5) I have been employed as an engineer continuously for over 20 years. My experience includes mineral processing and extractive metallurgy, process operations and management, process and infrastructure design and project management at various Zinc, Copper, Gold and other operations and research facilities in Africa, South America, Europe and North America. I have worked continuously as a consultant to mining operations since 2017.
- 6) I have read the definition of “Qualified Person” set out in National Instrument 43-101 – *Standards for Disclosure for Mineral Projects* (“NI 43-101”) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfil the requirements to be a “Qualified Person” for the purposes of NI 43-101.
- 7) I made personal inspections of the Okanjande Project site during December 2021, January and February 2022 and May 2023.
- 8) I am responsible for Sections 1, 2, 3, 13, 17, 18, 19, 20, 21, 22, 24, 25, 26, 27 and 28 and co-responsible for Section 5 of the PEA Report.
- 9) I am independent of the Issuer as independence is described in Section 1.5 of NI 43-101.
- 10) Prior to being retained by the Issuer, I have not had prior involvement with the property that is the subject of the PEA Report.
- 11) I have read NI 43-101 and Form 43-101F1, and the portions of the PEA Report for which I am responsible have been prepared in compliance with NI 43-101.
- 12) As of the effective date of the PEA Report, to the best of my knowledge, information and belief, the portions of the PEA Report for which I am responsible contain all scientific and technical information that is required to be disclosed to make the portions of the PEA Report for which I am responsible not misleading.

Dated: July 31, 2023



(signed/sealed)

Etienne Roux, SME-RM

CERTIFICATE OF QUALIFIED PERSON

Robert Barnett

I, Robert Nicholas Barnett, Pr. Sci. Nat. do hereby certify that:

- 1) I am an Associate Consulting Geologist of: The MSA Group (Pty) Ltd Henley House Greenacres Office Park, Victory Rd, Victory Park, Randburg, 2195
- 2) This certificate applies to the technical report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia, NI 43-101 Technical Report, that has an effective date of 29 November 2021 and a report date of 15 December 2021 (the Technical Report).
- 3) I graduated with a degree in BSc Eng Mining Geology from the University of the Witwatersrand in 1972. In addition, I have obtained an MSc Industrial Mineralogy in 1979 at Hull University.
- 4) I am a Fellow of the Geological Society of South Africa, a Professional Natural Scientist (Pr. Sci. Nat) registered with the South African Council for Natural Scientific Professions (SACNASP), and a member of the Zimbabwe Geological Society.
- 5) I have worked as a geologist for a total of 49 years, during which time I have worked in a wide range of roles ranging from base metal prospecting, industrial mineral prospecting and mining, manager of an industrial mineral laboratory, industrial mineral product research and industrial mineral market studies. I have had extensive experience in the mineral fields of graphite, limestone, clays, vermiculite, silica, talc and abrasive minerals.
- 6) I have read the definition of “qualified person” set out in National Instrument 43-101 (NI 43-101) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfil the requirements to be a “qualified person” for the purposes of NI 43-101.
- 7) I visited the Okanjande Graphite Mine property on 29 September 2021 for one day. I again visited the Property from 19 to 21 October 2021.
- 8) I am responsible for Section 2, 3, 4, 5, 7, 8, 9, 10 and 19; and co-responsible for, the preparation of sections 1, 11, 12, 23, 24, 25, 26 and 27 of the Technical report.
- 9) I have not had prior involvement with the property that is the subject of the Technical Report.
- 10) I am not aware of any material fact or material change with respect to the subject matter of the Technical Report that is not reflected in the Technical Report, the omission to disclose which makes the Technical Report misleading.
- 11) I am independent of the issuer according to the definition of independence described in section 1.5 of National Instrument 43-101.
- 12) I have read National Instrument 43-101 and Form 43-101F1 and, as of the date of this certificate, to the best of my knowledge, information and belief, those portions of the Technical Report for which I am responsible have been prepared in compliance with that instrument and form.
- 13) I consent to the filing of the Technical Report with any stock exchange and other regulatory authority and any publication by them for regulatory purposes, including electronic publication in the public company files on their websites accessible by the public, of the Technical Report.

Dated: July 31, 2023



(signed/sealed)

Robert Barnett, M (Eng); NHDip; BSc Geology (Hons), Pr Sci Nat, FGSSA

CERTIFICATE OF QUALIFIED PERSON

Ipelo Gasela

I, Ipelo Golang Rebecca Gasela, Pr. Sci. Nat. do hereby certify that:

1. I am a Senior Mineral Resource Consultant of: The MSA Group (Pty) Ltd Henley House Greenacres Office Park, Victory Rd, Victory Park, Randburg, 2195
2. This certificate applies to the technical report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia”, NI 43-101 Technical Report, that has an effective date 29 November 2021 and a report date of 15 December 2021 (the Technical Report).
3. I graduated with a B.Sc. (Hons) degree in Geology from the University of the Witwatersrand in 2004. I also obtained a M.Sc. (Eng) from the University of the Witwatersrand in 2018.
4. I am a Professional Natural Scientist (Pr. Sci. Nat) registered with the South African Council for Natural Scientific Professions (SACNASP) and a Member of the Geological Society of South Africa (GSSA).
5. I have worked as a Geologist for a total of 17 years, during which time I have worked in a number of roles; as an Evaluation Management trainee and a Senior Evaluator at a mine and corporate office for a mining company and a Mineral Resource consultant for mining consultancies, where I have undertaken Mineral Resource estimates, due diligence reviews and audits for a variety of commodities including graphite.
6. I have read the definition of “qualified person” set out in National Instrument 43-101 (NI 43-101) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfil the requirements to be a “qualified person” for the purposes of NI 43-101.
7. I have not visited the Okanjande Graphite Mine property.
8. I am responsible for Section 14 and co-responsible for sections 11, 12, 23, 24, 25 ,26 and 27 of the Technical Report.
9. I have not had prior involvement with the property that is the subject of the Technical Report.
10. I am not aware of any material fact or material change with respect to the subject matter of the Technical Report that is not reflected in the Technical Report, the omission to disclose which makes the Technical Report misleading.
11. I am independent of the issuer according to the definition of independence described in section 1.5 of National Instrument 43-101.
12. I have read National Instrument 43-101 and Form 43-101F1 and, as of the date of this certificate, to the best of my knowledge, information and belief, those portions of the Technical Report for which I am responsible have been prepared in compliance with that instrument and form.
13. I consent to the filing of the Technical Report with any stock exchange and other regulatory authority and any publication by them for regulatory purposes, including electronic publication in the public company files on their websites accessible by the public, of the Technical Report.

Dated: July 31, 2023



(signed/sealed)

Ipelo Golang Rebecca Gasela, Pr. Sci. Nat, MGSSA

CERTIFICATE OF QUALIFIED PERSON

Mark Richard Mohring

I, Mark Richard Mohring, Consulting Associate Mining Engineer with MSA Group (Pty) Ltd, and Registered Professional Engineer, as a co-author of the technical report entitled “OKANJANDE GRAPHITE PROJECT: Preliminary Economic Assessment Study Update Report, Otjozondupa Region, Namibia” (the “PEA Report”) with an effective date of 30 June 2023 and a report date of 31 July, 2023 prepared for Northern Graphite Corporation (the “Issuer”), do hereby certify:

- 1) I am currently employed as a Consulting Associate Mining Engineer with MSA Group (Pty) Ltd, Henley House, Greenacres Office Park, Johannesburg
- 2) This certificate applies to the PEA report titled “OKANJANDE GRAPHITE PROJECT Preliminary Economic Assessment Study Report”, NI 43-101 PEA Report, that has an effective date of 30 July 2023 and a report date of 31 July 2023 (the PEA Report).
- 3) I graduated with a Bachelor of Science Mining Engineering in 1995 from the University of the Witwatersrand, South Africa.
- 4) I am a Registered Engineer (#20060170) in terms of the Engineering Profession Act, 2000 (Act No. 46 of 2000) with the Engineering Council of South Africa (ECSA).
- 5) I am a registered member of Southern African Institute of Mining and Metallurgy (SAIMM)
- 6) I have been employed as an engineer continuously for over 25 years. My experience includes mine management, design and planning at various coal and gold and other operations in Southern Africa. I have been a mining consultant to various mining operations in Africa, providing mine design and planning services, mining reserve determination and sign off, mine and project valuation, financial and operational analysis of operating mines and Greenfields projects
- 7) I have worked continuously as a consultant to mining operations since 2008.
- 8) I have read the definition of “Qualified Person” set out in National Instrument 43-101 – *Standards for Disclosure for Mineral Projects* (“NI 43-101”) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfil the requirements to be a “Qualified Person” for the purposes of NI 43-101.
- 9) I made personal inspections of the Okanjande Project site during May 2021.
- 10) I am responsible for Section 15 and 16 of the PEA Report.
- 11) I am an independent of the Issuer as independence is described in Section 1.5 of NI 43-101.
- 12) I have read NI 43-101 and Form 43-101F1, and the portions of the PEA Report for which I am responsible have been prepared in compliance with NI 43-101.
- 13) As of the effective date of the PEA Report, to the best of my knowledge, information and belief, the portions of the PEA Report for which I am responsible contain all scientific and technical information that is required to be disclosed to make the portions of the PEA Report for which I am responsible not misleading.

Dated: July 31, 2023



(signed/sealed)

Mark Mohring

1 EXECUTIVE SUMMARY

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

1.1 INTRODUCTION

1.1.1 Issuer

Northern Graphite Corporation (“Northern” or the “Company”), a Canadian company listed on the Toronto Venture Exchange (“TSX-V”), acquired 100% ownership of Imerys Gecko Holdings (Pty) Ltd, which in turn owned 100% interest in Imerys Gecko Graphite (Namibia) (Pty) Ltd and Imerys Gecko Okanjande Mining (Pty) Ltd. The companies were rebranded to Northern Graphite Holdings (Namibia) (Pty) Ltd (“Holdings”), Northern Graphite Okanjande Mining (Pty) Ltd (“NGOM”) and Northern Graphite Processing (Namibia) (Pty) Ltd. (“NGP”). Collectively, Holdings, NGOM and NGP are referred to as “Northern Graphite (NG) Namibia” as they operate as one under joint management and control.

NGOM holds Mining Licence (“ML”) 196 (the “Property”) which covers the Okanjande graphite deposit. It is located approximately 23km by road to the southwest of the town of Otjiwarongo, 230km north of Windhoek, the capital city of the Republic of Namibia (“Namibia”), and 388km east from the port of Walvis Bay. The previously producing Okanjande graphite Mine, active from 2017 to 2018, is located on the Property. The Okanjande graphite Mine is currently under care and maintenance, with no mining taking place. NGOM also holds Exclusive Prospecting Licence (“EPL”) 4717 which surrounds the mining licence.

NGP owns all the equipment necessary to process graphite-bearing material and produce a saleable concentrate. The equipment is currently located on land and in buildings leased from Okorusu Fluorspar (Pty) Ltd at the idle Okorusu Fluorspar mine site (“Okorusu”), all of which will be relocated to the Okanjande mine site. Okorusu is 60km, by road, northeast of Otjiwarongo and is located on Mining Licence ML 90, held by Okorusu Fluorspar (Pty) Ltd.

This Preliminary Economic Assessment Report (PEA) was prepared Creo Engineering Solutions (Pty) Ltd on behalf of the Company for the Okanjande Graphite Project (the Project). The Okanjande Mine and Processing plant is located approximately 60km southwest from the town of Otjiwarongo in the Otjozondjupa region of Namibia. Okanjande mine and the Okorusu process plant assets have both been included as part of the acquired JV. The project has been under care and maintenance since 2018 and is planned to be restarted by the client after executing a process plant relocation, several brown fields expansion projects, modifications and refurbishment activities.

1.1.2 Terms of Reference

Northern Graphite commissioned a team of consultants in mid-2021 to conduct a PEA study on the Okanjande Graphite Project. The study was updated in Quarter 1 of 2023 to include the relocation of processing operations to the Okanjande mine site. The Project intends to bring the in-situ (weathered and fresh mineralized material) resource into production. A production rate of 31,103kt/a graphite concentrate of greater than 95% TGC is projected. This PEA focusses on bring in only 6,1Mt of mineralized rock into production from a total resource of 24.2Mt.

The purpose of this PEA Report is to provide an economic analysis of the potential viability of the Okanjande mineral resource. This report was compiled in accordance with the requirements and specifications of the CIM NI 43-101 standard.

1.1.3 Sources of Information

Information contained in this report includes abstracts of professional sources for the different fields of expertise. The following table provides a list of the key sources of information incorporated into this document. See Table 1-1.

Table 1-1: Key sources of information

Source	Title
MSA Group	Technical Report: Northern Graphite Corporation Okanjande Mineral Resource Estimate Namibia, November 2021
MSA Group	Okanjande Project – Mining Report, December 2021
Knight Piésold Consulting	Environmental Scoping Study and Gap Analysis, January 2022
CREO Engineering Solutions	Okanjande Graphite Project: Preliminary Economic Assessment Study, July 2022
METPRO	Process Review and Optimization for the Okorusu Plant, October 2021

It should be noted that some of the historical reports used refer to the mineralization as “ore”. These reports were generated prior to the establishment of the NI-43-101 guidelines and should not be interpreted as being a proven, economically extractable, material.

1.1.4 Competent Person Site Inspection Report

E. Roux – Consulting Process Engineer, CREO Engineering Solutions

Mr Etienne Roux, Director at CREO Engineering Solutions Pty. Ltd., is the Competent Person for the Mineral Processing and Metallurgical testing, Recovery Methods, Capital and Operating Costs and Economic Analysis. I have visited the site on 4 occasions: 19 July 2021, 13 December 2021, 18 January 2022, 18 February 2022 and 8 May 2023 with relevance pertaining to inspection of existing processing equipment, infrastructure, tailings storage facility, previous mining operations and general lay of land. I have read the instrument, and the parts of the report that I am responsible for have been prepared in compliance with the instrument.

R.N. Barnett – Consulting Associate Industrial Minerals Geologist, MSA Group (Pty) Ltd

Mr Robert Nicholas Barnett, Associate Consulting Geologist with MSA Group (Pty) Ltd, is the Competent Person responsible for the Geological Setting and Mineralization, Deposit Types, Exploration and Drilling, and co-responsible for Sample Preparation, Analyses and Security, and Data Verification. I have visited the site on two occasions from 28 September to 1 October and 18 to 22 October 2021 with relevance to geology, drill core and graphite mineralized rock exposed in the mine pit. I have read the instrument and the parts of the report that I am responsible for and have been prepared in compliance with the instrument.

M Mohring – Consulting Associate Mining Engineer, MSA Group (Pty) Ltd

Mr Mark Richard Mohring, Associate Consulting Mining Engineer with MSA Group (Pty) Ltd, is the Competent Person responsible for the Mining Report. I have visited the site from 11 May 2021 to 14 May 2021, where I inspected the pit and associated infrastructure of the mining and processing operation. I have read the instrument and the parts of the report that I am responsible for have been prepared in compliance with the instrument.

1.2 RELIANCE ON OTHER EXPERTS**1.2.1 CREO Disclaimer**

This report was prepared for Northern Graphite Corporation (the Client), by Creo Engineering Solutions (the Consultant), as an independent Consultant, and is based in part upon information and historical study work furnished by the Client and third-party consultants (see Table 2-1).

Information on the graphite market (section 19 Market Studies and Contracts) was compiled from a report commissioned from Benchmark Mineral Intelligence (“BMI”), as well as other industry sources, by Gregory Bowes P.Geol. who is the Chairman of the Board of Northern and not an Independent Person

While it is believed that the information, conclusions, and recommendations will be reliable under the conditions and subject to the limitations set forth herein, the Consultant cannot guarantee their accuracy. The opinions expressed in this report subject to the project responsibilities fulfilled by Creo Engineering Solutions and sub-consultants, are Creo Engineering Solutions’ best estimate taken from the data available, combined with experience and knowledge of the work. The Client and any other Third Parties that might review or use the information in this report are responsible for their independent analysis based on the information herein and must acknowledge the risks associated with such an operation.

1.2.2 MSA Disclaimer

Neither MSA nor the authors of this Technical Report, are qualified to verify the legal status of ML 196 and EPL 4717. MSA has relied on a legal opinion by Ellis Shilengudwa Incorporated (“ESI”), dated 09 December 2021, that IGOM was the legal holder of 100% of the interest in ML 196 and EPL 4717, which are valid as of the date of the opinion. ESI is a corporate and commercial law firm based in Windhoek, is regulated by the Law Society of Namibia, and is a member of DLA Piper Africa, a Swiss Verein whose members are comprised of independent law firms in Africa working with DLA Piper. Furthermore, ESI have verified the Existing Lease Agreement in place and the form of the New Lease Agreement in which the agreement(s) provide Imerys-Gecko with the necessary rights, title and interest in and to the Leased Premises to allow for the conducting of agreed processing activities.

The present status of tenements listed in this report (Item 2 – Introduction preface, Item 4.2 – Mineral Tenure, Permitting, Rights and Agreements, and Item 4.3 – Environmental Liabilities) is based on information and copies of documents provided by Northern and ESI, and this Independent Technical Report has been prepared on the assumption that the tenements will prove lawfully accessible for evaluation. Comment on these legal agreements is for introduction only and should not be relied on by the reader. ESI has not provided comment on the status of ML90 which is not the focus of this Technical Report.

Similarly, neither MSA nor the authors of this report are qualified to provide comment on environmental issues associated with the Property.

1.3 PROPERTY DESCRIPTION AND LOCATION

Northern acquired 100 percent ownership of Imerys Gecko Holdings (Pty) Ltd, which in turn owned 100% interest in Imerys Gecko Graphite (Namibia) (Pty) Ltd and Imerys Gecko Okanjande Mining (Pty) Ltd. Imerys Gecko Okanjande Mining (Pty) Ltd held ML 196 which covers the Okanjande graphite deposit (“Okanjande”). Mining started at Okanjande in 2017 and the ROM product was transported to the idle Okorusu Fluorspar mine site (“Okorusu”) for processing over an 18-month period, before the Okanjande Mine was placed on care and maintenance in October 2018. Imerys Gecko Okanjande Mining (Pty) Ltd was renamed as Northern Graphite Okanjande Mining (Pty) Ltd (“NGOM”)

1.3.1 Location

The Property is located approximately 23km southwest of the town of Otjiwarongo, 230km north of Windhoek, the capital city of Namibia, and 388km from the port of Walvis Bay.

1.3.2 Mineral Tenure, Permitting, Rights and Agreements

The Okanjande Mine is located within ML 196. ML 196 is valid until February 9, 2042 and covers an area of 903.4ha.

NGOM also holds an Exclusive Prospecting License EPL 4717 which surrounds the mining license, covers 46,670ha, and is valid until March 2023. The EPL 4717 Renewal Application was submitted in December 2022.

An Environmental Compliance Certificate (“ECC”) covering mining and exploration activities for Okanjande was applied for, and received from the national authorities on November 11, 2021, valid for three years until November 11, 2024.

1.3.3 Environmental Liabilities

Historically, Imerys Gecko Okanjande Mining (Pty) Ltd carried out contract mining operations at the Okanjande graphite Mine during parts of 2017 and 2018 and mined approximately 300,000 tonnes of weathered mineralisation (most of which was transported to the Okorusu processing plant) and 400,000 tonnes of weathered waste rock. The waste rock and a stockpile of approximately 14,000 tonnes of graphite-bearing material remains on the Okanjande Property. Geochemical testing by Knight Piesold (Knight Piesold Consulting, 2022) of tailings and waste rock indicated that all weathered tailings and waste rock are potentially acid generating and that all transitional and fresh tailings and waste rock are acid generating. The waste rock has been deposited on a calc-silicate unit which has low permeability and provides natural buffering from any acidic seepage to occur. Mitigating measures are required to manage run-off from the historical waste dump. It is recommended that run-off be routed to the adjacent storm water pond.

1.4 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

Namibia is located in the south-western part of Africa. It is bound by the Atlantic Ocean to the west, and between South Africa and Angola to the south and north respectively. With an estimated population of 2.1 million and an area of 824,292km², Namibia is one of the most sparsely populated countries in the world. The capital city is Windhoek, with approximately 350,000 inhabitants.

1.4.1 Access and Infrastructure

1.4.1.1 Regional – Otjiwarongo / Otjozondupa

Otjiwarongo has a well-developed road network. It is situated at the junction of national road B1 that passes north-south through all of Namibia, the C38 to Outjo and further into the Kunene Region in Namibia's northwest, and the C33 to Karibib, which connects the coastal town of Swakopmund and the port city of Walvis Bay.

Otjiwarongo is connected by road and rail to the port at Walvis Bay, a natural deep-water harbour. Walvis Bay is the largest port on the country's coast and is an important logistical port for the southern African region, providing import and export facilities for Namibia, Zambia, the Democratic Republic of Congo, and Botswana. The port has the capacity to move one million containers a year.

1.4.1.1.1 Local Resources

Local resources required for labour, supplies and equipment will be insufficient from the nearby community to support the project. Some local resources can be utilised, but skilled mining personnel, consultants and suppliers will be available throughout Namibia to support the project and its operations.

During operations, Okanjande will have to source its key personnel outside of the immediate area, Otjiwarongo, the greater Namibia and possibly from the neighbouring country – South Africa.

Otjiwarongo's population sustains a diverse economy including cement manufacture, many service industries and institutions. The economics of the town mainly revolve around agriculture as the majority of businesses are related to this sector. Some 55,000 people are employed in the Otjozondjupa region, of whom about 31,000 are male. The employment participation rate is approximately 70%; this is above the national average of 55%. The largest occupational group is labourers and other unskilled trades, which constitute 40% of all those employed.

1.4.2 Climate, Operating Season

Most of Namibia has a subtropical desert climate characterized by hot summers and mild winters, great differences in day- and night-time temperatures, low rainfall and low humidity. Sunshine days are reported to average 300 days per year. Namibia experiences winter and summer at opposite times from Europe and North America. The winter (June to August) is generally dry. Although arid, the rainy season in the inland north-central areas and in the northeast is from November to March.

Rainfall in the Otjiwarongo area is fairly constant at 400 – 500mm per year; rain showers occur mainly during November to April, with the highest rainfall rates recorded in January and February. Rain showers are often of short duration and high intensity.

Yearly average mean temperatures are 21°C, with average monthly highs and lows of 29°C and 13°C respectively. However, much higher and lower daily temperatures often occur. The highest temperatures (in degrees Celsius) are experienced in January (an average of 22.7°C with highs of 37.2°C and lows of 12.1°C) and the lowest in June (an average of 15.2°C with highs of 28.5°C and lows of -1.9°C).

The Okanjande Mining Licence can be operated and/or are accessible all year.

1.4.3 Physiography

The Otjozondjupa region is characterized by a relatively flat plain with inselbergs and low ridges standing out above the general land surface. The general elevation is between 1,300m and 1,600m above mean sea level (“mamsl”). The Okanjande graphite deposit is situated on the lower west-facing slopes of one of the ridges. The proximity of other low ridges to the site means that the Okanjande deposit is effectively screened from public view.

The Project area forms part of the thornbush savanna with fairly thin soil and occasional patches of bedrock exposed at the surface. This habitat is widespread and homogeneous throughout central Namibia and contains no unique or singular features of high ecological importance. None of the species found on site are rare, threatened or endangered. The current land use in the area consists primarily of livestock farming and, to a lesser degree, hunting tourism operations. Because of the thin soils and low rainfall, the land has a very low agricultural capability rating. Grazing is only suitable on the lower slopes and valley areas as the soils on the upper parts of the ridges are too thin to support any specific land use.

1.5 HISTORY

In 1990, Rössing Uranium Limited (“RUL”), a subsidiary of Rio Tinto, noted the occurrence of flake graphite on a group of farms immediately south of Otjiwarongo. Subsequent reconnaissance geological programmes revealed substantial amounts of flake graphite particularly on the farms Okanjande and Highlands. Between 1991 and 1993, RUL undertook a number of detailed studies pertaining to the Project including geological, drilling, metallurgical and environmental studies and a feasibility study. Extensive exploration drilling and trial mining operations were carried out. A pilot plant was constructed, and metallurgical test work completed.

Gecko Namibia (Pty) Ltd (“Gecko”) subsequently acquired the rights to Okanjande. In 2016 Gecko entered into a Joint Venture (“JV”) with Imerys S.A. (“Imerys”) to form Imerys Gecko Holdings (Pty) Ltd (“Holdings”). The development scenario pursued by the JV consisted of mining weathered mineralisation at the Okanjande deposit and transporting it to the Okorusu fluorspar plant for processing. This plant was refurbished and modified to suit the requirements of graphite processing. The Okanjande/Okorusu operation started up in April 2017, however the processing plant encountered a number of difficulties with throughput, flake size and purity all being lower than expected. The operation was put on care and maintenance in November 2018.

1.6 GEOLOGICAL SETTING AND MINERALIZATION

1.6.1 Regional Geology

The Okanjande deposit is located in a metamorphic complex in the north-central (“nCZ”) part of the Neoproterozoic Damara Orogen. The Damara Orogen comprises three highly oblique mobile belts; namely the north-northwest trending Kaoko Belt in the northwest, the Gariep Belt in the southwest and the north-eastern trending Damara (inland) Belt.

The Project is located in the nCZ and is underlain by Nosib Group sediments and upper Swakop Group shelf carbonates that are intruded by a number of syn- and late-tectonic granites and pegmatites (Steven, 1993).

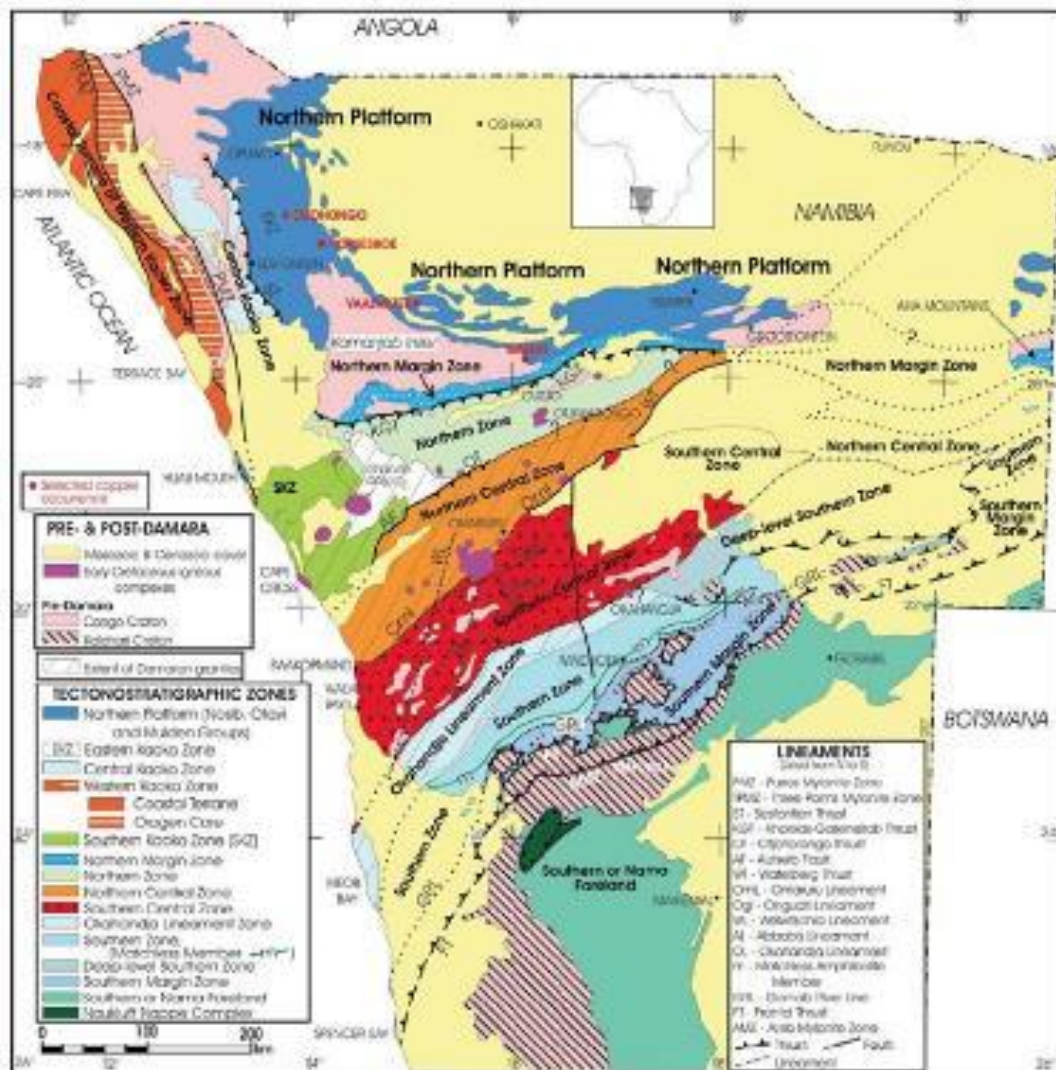


Figure 1-1: Regional Geology of the Damara Belt (Miller, 2013)

1.6.2 Local Geology

The Nosib and Swakop Groups in the Project area have been intruded by syn- and post-tectonic pegmatites and granites, with the latter typically comprising two types of leucogranites. These pegmatites and granites occur as

irregular and semi-disconcordant narrow bodies, which cross-cut the metasedimentary layering in places. Quartz veining also occurs in the area. The sedimentary rocks have been subjected to high grade regional metamorphism (RUL, 1993; RUL, 1994).

The Nosib Group comprises arkoses, feldspathic quartzites and ortho-quartzites that have been metamorphosed to feldspathic-quartzitic gneisses and graphitic quartzite, which host the graphite as an accessory mineral. The graphite mineralization is hosted locally in discrete units. Sulphide mineralization is prevalent in the Nosib Group.

The Swakop Group conformably overlies the Nosib Group, with its basal contact marked by a thin biotite schist. Cover rocks comprise Quaternary sediments, including alluvium, sand and gravel, which tend to accumulate in the valleys.

The area has been subjected to multiple phases of deformation, which have resulted in dome-and-basin structures with a WSW-ENE fold axis.

Major north-south and northeast-southwest orientated faults cut across the region, which are typical of the Damara region.

1.6.3 Property Geology

In the Project area, the Nosib and Swakop Groups are tightly folded along a west-southwest fold axis, resulting in quartz-rich ellipsoidal domes of the Nosib Group, while the valleys comprise Swakop Group. Graphite occurs locally as disseminated flakes in stratabound units.

The Okanjande deposit reaches its highest grade and is thickest in the centre, with graphitic carbon grades and mineralization thickness decreasing towards the peripheries. The contacts of the graphite units are difficult to define as they tend to be transitional.

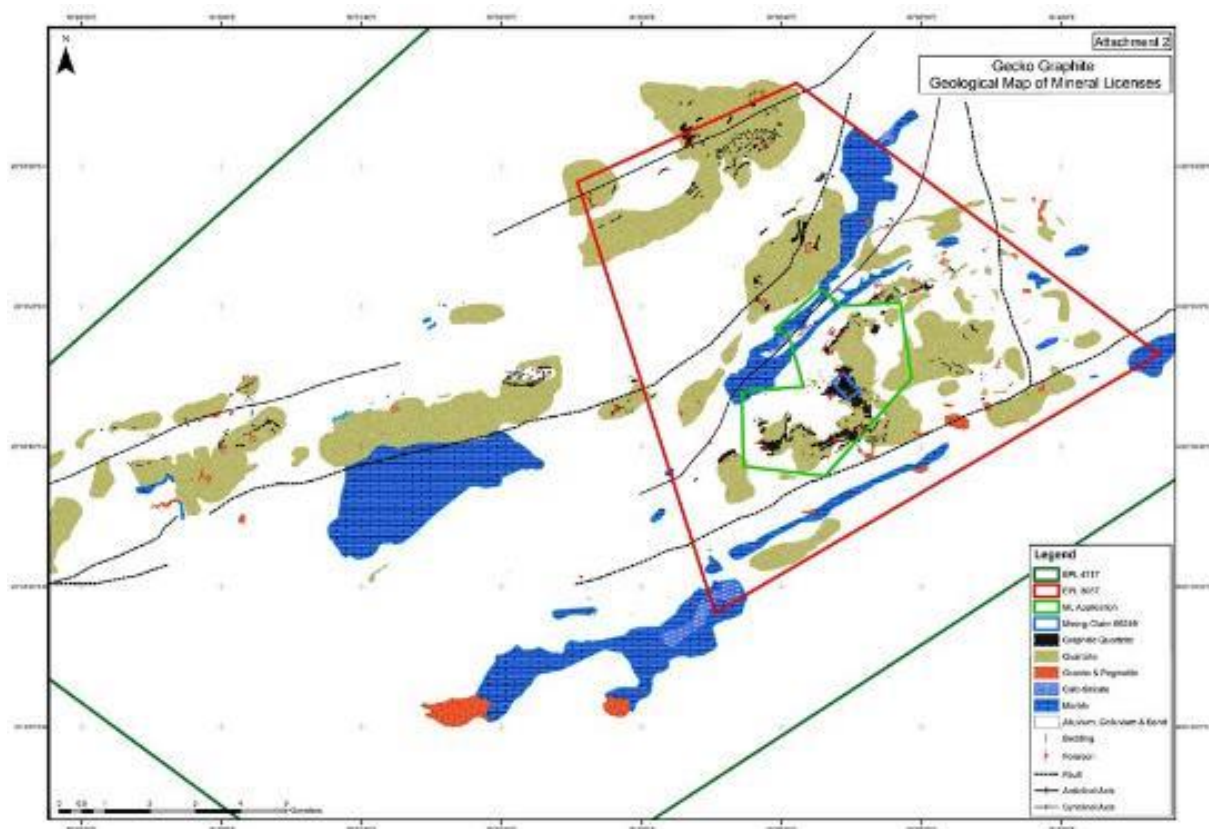


Figure 1-2: Okanjande Property Geology Map

The Okanjande deposit is a massive deposit with shallow dipping layering to the south-east. It extends on surface approximately 900m northeast to southwest and up to 820m from northwest to southeast. The mineralization extends from surface and has been defined by drilling up to a depth of 90m below surface. The mineralization is open at depth. The deposit is extensively weathered to depths of generally between 10m and 20m.

1.6.4 Mineralization

RUL completed mineralogical studies on selected Okanjande deposit core samples at an independent laboratory, Mintek, Johannesburg. The mineralogical studies were completed on weathered and fresh mineralized samples from 12 drillhole cores using transmitted and reflected light microscopy and point counting methods, scanning electron microscopy (“SEM”) and X-ray diffraction (“XRD”).

The mineralogical investigations revealed that the graphitic samples generally contain similar minerals, although they vary significantly in both proportion and texture. The minerals identified are shown in Table 1-2.

Table 1-2: Okanjande Mineralogy

Major Constituents (>10%)	Minor Constituents (between 10% and 1%)	Trace Constituents (<1%)
Quartz	Muscovite	Rutile
K-Feldspar	Plagioclase	Zircon
Jarosite (K-Fe-sulphate)	Graphite	Apatite
	**Phlogopite	Chlorite
	**Biotite	+Chalcopyrite
	Sillimanite	Ce-Phosphate
	+Pyrrhotite	Monazite
	+Pyrite	Ba-K Feldspar
	*Fe-Hydroxides (Goethite)	Baryte
	*Haematite	+Molybdenite
		Wolframite
		Cassiterite
		+Ni-Sulphide
		Cr-spinel
		Xenotime

Source: RUL, 1993

Note: * in weathered mineralized samples
+ in fresh mineralized samples
** present in some rocks only

1.7 DEPOSIT TYPES

Graphite mineralization occurs in three types of deposits worldwide: microcrystalline, vein graphite and crystalline flake graphite. The Okanjande deposit fits the description of a crystalline flake graphite deposit in a paragneiss. The original rocks at the Okanjande deposit were arkoses, feldspathic quartzites and ortho-quartzites that have been subjected to high grade metamorphism to become feldspathic-quartzitic gneisses and graphitic quartzites.

1.8 EXPLORATION

Exploration had been carried out by various companies since 1990. In addition to drilling work and analyses of the drillhole samples (documented in Items 10 and 11), the following exploration was undertaken:

- 1990 – 1993: RUL carried out topographic and aerial surveys.
- 2003 – 2010: Solvay – no work undertaken.

- 2010 – 2016: Gecko completed mapping, soil sampling, pitting, geophysical surveys and topographic surveys; and
- 2016 – to date: Imerys-Gecko completed a review of the exploration data and a topographic survey

1.9 DRILLING

Drilling at the Okanjande deposit was completed by RUL (early 1990s), Gecko (2014 to 2016) and Imerys-Gecko (2018).

1.9.1 Rossing Uranium Drilling

In 1990, RUL drilled 66 percussion drillholes and 25 diamond drillholes, with 11 of the diamond drillholes twinning the percussion drillholes at the Okanjande deposit (also known as Target 2).

The drillholes were either drilled at an approximate azimuth of 320° and a dip of 60° or vertical. The drillholes were completed along southeast to northwest lines at an average line spacing of 50m, with the southwestern part of the deposit drilled at 100m line spacing. The holes were drilled 50m apart on each line. Most of the drillholes reached a depth of 60m, with the minimum and maximum depths being 40m and 90m, respectively.

According to RUL reports (RUL, 1993; RUL, 1994), RUL originally identified four drilling targets. Target 2 is the Okanjande deposit discussed in the preceding paragraphs.

RUL also completed three percussion drillholes 600m to the northwest of the main Okanjande deposit for sterilization purposes for a tailings area site. The drillholes intersected mineralization, and the area (as labelled by Rossing) became known as the “Tailings Area Mineralized Zone” and is now referred to as the “Northern Mineralized Band”.

1.9.2 Gecko Drilling

From 2014 to 2015, Gecko drilled 91 diamond drillholes, which include infill drilling, twin drilling of the RUL drilling and exploration drilling in the fringes and outside the main Okanjande deposit. Eleven RUL percussion drillholes and two RUL diamond drillholes were twin drilled. The infill drilling resulted in approximately 25 m spacing between section lines towards the centre of the deposit. Exploration drilling was in the peripheries and in the Northern and Southern Extension of the main Okanjande deposit, the “Northern Mineralized Band”, and approximately 2km northeast of the main Okanjande deposit.

All Gecko holes were from diamond drilling. Holes were drilled at an azimuth of 320° and dip of 60°, or vertical in the central parts of the Okanjande deposit, while the rest of the holes were drilled at various orientations including vertically.

In 2014 a topographic survey was completed on the Namibian Bessel (Schwarzeck) system LO22/17 and was later converted to UTM WGS84 33S.

1.9.3 Imerys-Gecko Drilling

From June to September 2018, Imerys-Gecko completed a total of 44 diamond drillholes within the Okanjande deposit and surrounding prospects. Twelve of these were infill holes drilled in the southwestern portion of the

deposit, to reduce the drill line spacing from 100m to 50m. These holes were drilled at an approximate azimuth of 320° and a dip of 60° and at 50m spacing within the drill lines. These holes are within the current Mineral Resource area.

Imerys-Gecko identified additional drilling targets from the airborne survey and mapping previously completed by Gecko. Imerys-Gecko considered three of the targets to be of a higher priority for further drilling; the “Northern Mineralized Band” (farm Okanjande), “SW Target” (farm Highlands) and “Old Mine” (farm Welgelegen).

1.10 SAMPLE PREPARATION, ANALYSIS AND SECURITY

1.10.1 Rossing Uranium Campaign

Percussion drilling chips were collected at 1 m intervals. The sample material was collected directly from the cyclone underflow and was homogenised by passing it through the riffle splitter three times prior to collecting the sample. A sample of 2kg to 3kg was collected from the riffle splitter and bagged for assay. Diamond drillhole cores were cut longitudinally into half, then cut longitudinally in half again. The quarter core samples were collected at 2 m intervals continuously through the entire length of the drillhole core. RUL submitted samples to Scientific Services in Windhoek for sample preparation, entailing crushing, pulverising and homogenising. Sub-samples of pulverised material were dispatched to Scientific Services in Cape Town, an independent commercial laboratory, for graphitic carbon, sulphur and titanium oxide analyses. It was reported that a total of 94 field duplicate samples were collected as well as 25 field duplicates for umpire assays at a second laboratory. No further QAQC samples were mentioned in the RUL reports.

1.10.2 Gecko Campaign

The cores were logged and marked up for sampling by geologists. Sample positions were marked on the core using a permanent marker at 1 m intervals, while adhering to the geological contacts identified during logging. The cores were split longitudinally in half using a diamond saw and again into quarter core for sampling purposes. The cores were broken along sample markings, individual samples were placed into bags with uniquely numbered sample tickets and sample numbers were written on the outside of the bags. Visibly un-mineralized intervals were not sampled. All remaining cores were stored in galvanised core trays on site in a core shed. In 2018, a secure core storage facility was built next to the Okanjande mine offices and all the cores are currently stored there.

Samples were dispatched by Gecko to Bureau Veritas (BV) in Swakopmund, a SANAS accredited independent commercial laboratory, for sample preparation. Upon receipt of the pulverised material from BV at Gecko’s premises in Nonidas, outside of Swakopmund, Gecko homogenised the material and split it using a riffle splitter to prepare between 30g and 50g sub-samples for assaying. The samples were submitted to BV Swakopmund for dispatch to BV Rustenburg for assaying.

The samples were assayed for sulphur and total graphitic carbon (“TGC”) at BV in Rustenburg, which was a SANAS accredited independent commercial laboratory at the time (the laboratory is now closed). Gecko used five graphitic CRMs, a titanium oxide CRM and pulp duplicates for its QA/QC samples submitted to BV in Swakopmund. The CRMs showed acceptable accuracy for the TGC assays.

It is the QP's opinion that the Gecko TGC and S assays have been demonstrated by QA/QC processes to be of acceptable accuracy and precision to use in Mineral Resource estimation.

1.10.3 Gecko-Imerys Campaign

The samples from the Imerys-Gecko cores were taken at 1 m intervals continuously through the entire core length while adjusting for geologically logged boundaries. The samples were recorded on a sampling sheet. The cores were cut longitudinally in half using a diamond saw and one half was cut again longitudinally in half to generate quarter core. The cores were broken at sample beginning and end markings, and individual samples were placed in plastic bags together with manila sample tags labelled with the unique sample number. The bags were sealed with twine and another sample label was tied to the twine. Visibly un-mineralized intervals were not sampled. The remaining cores are stored in galvanised steel core trays in a secure core storage facility near the Okanjande mine office.

The Imerys-Gecko samples were transported by courier to BV Swakopmund. Upon receipt of samples, the laboratory signed a form to maintain chain of custody. Sample preparation was completed at BV Swakopmund, where the samples were crushed, split and pulverised. The pulverised sub-samples were dispatched to BV Centurion (South Africa), where the samples were analysed for S and TGC using method codes ACT-TPM-013 and ACT-TPM-028, respectively.

Imerys-Gecko inserted duplicates as quarter core samples at a rate of 5%, for a total of 58. The Imerys-Gecko field duplicates show good precision for TGC and S, where 90% of the data have a HARD value of less than 15% for TGC and less than 13% for S. The precision of the Imerys-Gecko assays is good. Accuracy was not assessed, as CRMs were not inserted, and umpire analyses were not performed.

The precision of the Imerys-Gecko assays is good. Accuracy was not assessed, as CRMs were not inserted and umpire analyses were not performed. The degree of contamination is not known, as blank samples were not used. The QP is therefore partly reliant on indirect QA/QC for the Imerys-Gecko data, i.e., by examining global bias between different exploration campaigns that had more acceptable QA/QC protocols.

1.11 DATA VERIFICATION

1.11.1 Qualified Person

The Geology QP, Mr Rob Barnett, established an audit trail of the electronic drilling data that is captured in several spreadsheets that were provided to MSA.

Mr Barnett also completed two site visits, one between 28th September and 1st October and the second between October 18-22, 2021.

The Mineral Resource QP, Ms Ipelo Gasela, completed the statistical analyses on the data to determine their appropriateness to use in a Mineral Resource estimation. Ms Gasela did not visit site.

1.11.2 Rossing Uranium Twin Drilling

MSA completed a quantile-quantile (“QQ”) plot comparing the grades of the samples obtained by RUL percussion drilling to the twin diamond drillholes completed by RUL. The QQ plot indicates that there is no bias between the sample assays of TGC from the two drilling methods. The QQ plot shows a significant positive bias of 16% towards the three percussion holes that were assayed for sulphur. The bias is higher at depth where grades are higher. The results of the statistical tests from the twin drilling reveal that samples from the RUL percussion and diamond drilling have TGC grades that are not significantly biased with respect to one another.

It is therefore the QP’s opinion that the two datasets can be used together in Mineral Resource estimation. The S grades show a high bias towards the percussion samples at greater depths. There are only two percussion holes at depths exceeding 24m, therefore the inclusion of percussion data in the estimation has limited risk on the S estimates at depth.

1.11.3 Gecko Verification Work

1.11.3.1 Gecko Check Sampling of RUL Drillhole Samples

Gecko completed a check sampling programme on a selection of the remaining core intersections drilled by RUL. Gecko completed the check sampling by collecting 1 m samples of quarter core from the remaining RUL core, and a total of 129 check samples were assayed. The samples were assayed for TGC and S at Bureau Veritas in Rustenburg (South Africa), which was an independent and accredited commercial laboratory at the time. The remaining cores were photographed. The Gecko check samples were not accompanied by QA/QC samples.

The TGC assays show a strong positive relationship between the original and check sample assays, though with some scatter, resulting in a correlation co-efficient of approximately 0.94. The two datasets show almost no bias to one another.

The check sample S assays show a strong positive relationship with the original assays, with some scatter, resulting in a correlation co-efficient of 0.95. The check sample assays show a positive bias against the original assays of approximately 20%.

It is the QP’s opinion that the TGC assays of the RUL samples compare within acceptable limits to the Gecko check samples. However, there is a significant bias between the RUL and Gecko check sample sulphur assays.

1.11.3.2 Gecko Twin Drilling of RUL Drillholes

Gecko completed twin drilling of 11 RUL percussion holes and two diamond drillholes. The Gecko twin drillhole samples included samples of three CRM’s. All but one of the CRM samples returned assays within the acceptance limit of three-standard deviations of the certified value.

The CRM sulphur assays showed poor results, with 42% of the GGC-09 CRM and 33% of the GGC-10 CRM sample assays outside the acceptance limit of three-standard deviations of the certified value. Most of the failures are by a small margin.

The QQ plot of the paired hole TGC assays demonstrate that there is minor overall bias between TGC sample assays of the RUL and Gecko drilling. The QQ plot shows insignificant bias in the sulphur grades of the two datasets as well.

It is the QP’s opinion that the two datasets are considered acceptable to use in a Mineral Resource estimation.

1.11.4 Imerys-Gecko Verification Work

1.11.4.1 Imerys-Gecko Check Sampling of Gecko Drillholes

In 2018, Imerys-Gecko completed check sampling of OKD039, a Gecko diamond drillhole. A good linear correlation was observed between the original Gecko sample assays and the Imerys-Gecko check sample assays for TGC and S, with correlation coefficients of 0.99 and 0.98, respectively. However, the original Gecko TGC assays are on average approximately 6% higher than the Imerys-Gecko check sample assays, with a tendency to be lower in the low-grade ranges and higher in the high-grade ranges. The sulphur assays are on average 8% higher than the Imerys-Gecko check sample assays with a tendency to be higher for assays greater than 2%.

In the QP's opinion, biases between the Gecko and Imerys-Gecko assays are not material.

1.11.4.2 Bias Test Between Imerys-Gecko and RUL Drilling

The QP completed a bias test between the TGC assays from the Imerys-Gecko drillhole samples and the RUL drillhole samples in the southwestern part of the deposit. The data were composited to 2 m intervals and a threshold of 2% TGC applied for both sets of data to account for un-mineralized intervals that were not sampled in the Imerys-Gecko drilling campaign.

The TGC QQ plot shows that the two datasets have similar TGC grade distributions, with the Imerys-Gecko average TGC grade being 2.5% higher. The QQ plot between the RUL and Imerys-Gecko sample sulphur grades show slight conditional bias and the Imerys-Gecko average sulphur grade is 4% lower.

It is the QP's opinion that the two datasets verify each other, and this allows for the two datasets to be used together in the Mineral Resource estimation process.

1.12 MINERAL PROCESSING AND METALLURGICAL TESTING

1.12.1 Early Process Test Work - MINTEK

Early investigations were carried out by Rossing Uranium Limited and Mintek in South Africa in the period 1989-1992. Their reports are summarized below:

1.12.1.1 Preliminary investigation into the upgrading of Graphite Ore, 1989

Batch floatation tests produced high grade graphite concentrate of 93% Carbon.

The recommended flowsheet from test campaign: Rod Milling, Rougher Flotation, three Pebble Mill re-grinding stages, three stages of flotation and magnetic separation.

1.12.1.2 Preliminary tests to recover flake graphite from the Okanjande deposit, 1990

A flow sheet was tested consisting of Rod Milling, Rougher Flotation, Pebble Mill regrind and 3 stages for cleaner flotation. Concentrates in the region of 93% C were produced from both areas.

A Flash Flotation stage was added, which increased the yield of >300um flakes, but did not affect the yield of >106um flakes.

1.12.1.3 Preliminary tests on drill core samples from the Okanjande deposit to recover flake graphite, 1991

Milling and flotation tests completed on 32 drill core samples with an average head grade of 6% fixed carbon as graphite and produced concentrate grades ranging from 83% to 96% at recoveries in excess of 80%. Optimal flow sheet developed from the test program was Rod Milling, Rougher Flotation, Pebble Mill regrinding of rougher concentrate and 3 stages of cleaner flotation.

1.12.1.4 Metallurgical Commissioning of the Okanjande Graphite Beneficiation Pilot Plant, 1991

Rossing Uranium ran a pilot campaign on the Okanjande deposit to recover flake graphite in 1991. Mintek assisted in the metallurgical commissioning of the pilot plant. The observations from the campaign were:

- Closed circuit rod milling with 2mm classification screen applied without sanding in rougher cells. High recovery obtained from rougher cells.
- Cleaning flotation noted free gangue, but investigation indicated graphite attachments leading to floating of gangue.
- Free and visible gangue was a constant problem in the final concentrate. Remaining gangue was mainly mica.

1.12.1.5 Autogenous Milling and Flotation of Okanjande Ore, 1992

Weathered and Fresh mineralized rock samples proved to be amenable to autogenous milling during laboratory testing. Closed circuit milling with 2mm screen classification was used and resulting in 40% circulating loads and a screen undersize P80 of 425um.

Autogenous milled samples produced similar flotation performance than rod milled samples, but fresh mineralized rock autogenous milled samples produced coarser graphite flake sizes than Rod milling.

Fresh material milling indices for AG and SAG milling were found to be: 2.5kWh/t and 3.1kWh/t respectively.

Weathered drop tests indicate high friability, but with enough competent rocks in between to allow for autogenous milling.

1.12.1.6 The Purification of Okanjande Graphite Concentrate: Exploratory Work, 1992

Exploratory test work was conducted in the pursuit of producing 99.5 and up to 99.9% concentrate purity.

Roasting of graphite concentrate with either sodium hydroxide or sodium carbonate, followed by successive leaching with water, sulfuric acid and hydrofluoric acid produced greater than 99.9% purity.

Roasting of graphite concentrate with sodium hydroxide followed by successive leaching with water and sulphuric acid produced graphite concentrate with greater than 99.5% purity.

1.12.1.7 Pneumatic Flotation of Graphite at Okanjande, 1992

Column and Jameson float cells were testing in the pilot plant at various functional locations. In general, it was found that the column cells produced slightly better recoveries at similar concentrate grades to that obtained from the Jameson cells but required additional frother addition.

1.12.2 Early Mineralogical Characterization – MINTEK

Mineralogical investigations were carried out by MINTEK over the period of 1991 to 1992.

1.12.2.1 A Mineralogical Examination of the Okanjande Graphite Ores, March 1991

Seventeen borehole core intersections were tested. The predominant minerals were Quartz, K-Feldspar, Graphite, Mica, Pyrite, Pyrrhotite and Sillimanite. Mean graphite flakes sizes were found to be in the region of 0.5-0.7mm while maximum flake size ranging from 1.0 to 2.0mm were observed.

Two distinct mineralized material textures were described:

1. Altered mineralized material with distorted graphite flakes which contained closely intergrown gangue minerals. Deformation appeared to be the result of shearing and deformation of the mineralized material. Full liberation of graphite expected to be difficult to achieve.
2. Un-Altered mineralized material containing undistorted and largely inclusion-free graphite.

The most common phase interlaminated with graphite flakes was Muscovite and Sericite. Sulphide minerals were also frequently found to be interlaminated with graphite.

1.12.2.2 A Mineralogical Investigation of the Sulphide minerals present in graphite concentrates from Okanjande, July 1991

Graphite concentrates prepared from Okanjande mineralized material were examined for the occurrence and nature of sulphide minerals.

The sulphide phases present were Pyrite, Pyrrhotite, Chalcopyrite and Molybdenite. The sulphide phases were present as free grains as well as occlusions within the graphite flakes. The fraction of occluded sulphides was the highest for the coarse graphite (>300um) fraction and the lowest in the fine graphite (38um).

Chalcopyrite and Molybdenum were found to preferentially occur in the fine fractions of graphite, perhaps due to their milling characteristics.

1.12.2.3 A Mineralogical Investigation of the Sulphate Minerals present in graphite Ore samples from Okanjande, July 1991

Fifteen samples of Okanjande graphite mineralization were examined. The main sulphate phases present were Jarosite, Barite, Gypsum, Siderotil and Melanterite. Siderotil and Melanterite are water soluble and mainly found in shear and fractures as well the altered mineralized material zones at depth.

The iron sulphate minerals are mainly present in the weathered mineralized material, but do occasionally occur in fresh mineralized material along fractures and joints. The presence of iron sulphates are the result of weathering.

1.12.3 Recent Process Test Work

1.12.3.1 Gecko MINTEK test program 2015

The test work program comprised comminution characterisation, mineralogy and flotation test work to develop a base case flowsheet. The flowsheet development test work was conducted on weathered material and the final flowsheet was tested on the fresh material.

1.12.3.1.1 Flotation Testing

The optimal processing flowsheet selected (Figure 1-3) consisted of rougher and cleaner flotation cells with scavengers on both tailings to maximise recoveries. The cleaner concentrate was milled in a pebble mill to liberate graphite flakes attached to gangue with minimum flake breakage. A double stage pebble mill was required for maximum graphitic carbon recovery. The base case flowsheet utilised DOW 200 as frother without the addition of a collector.

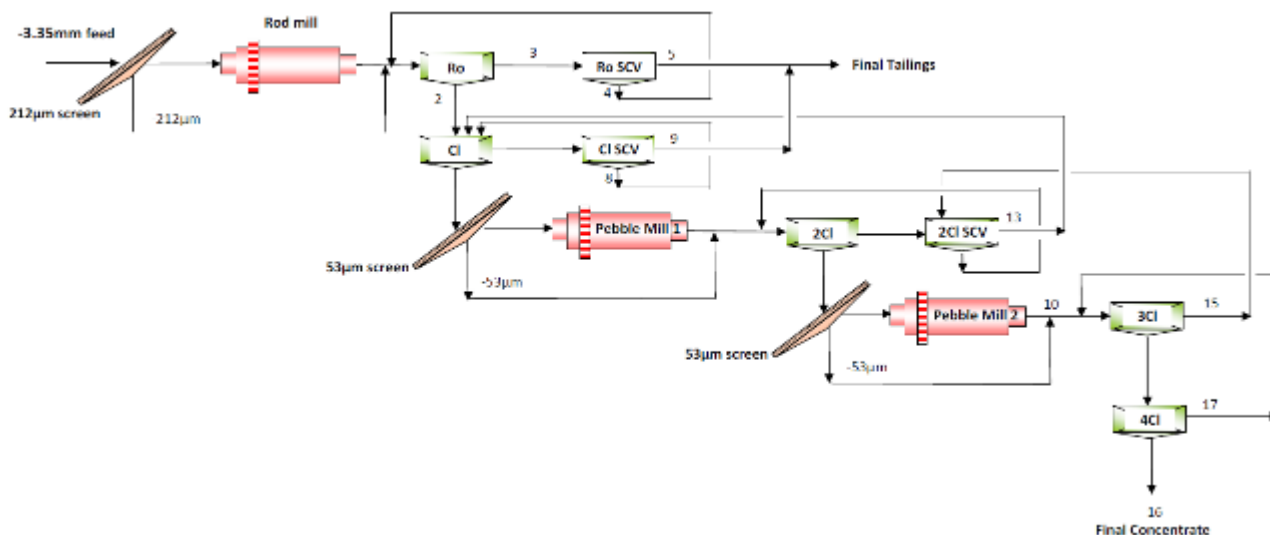


Figure 1-3: MINTEK Optimum Graphite Flotation Flowsheet

Test work on the weathered material showed that it was possible to obtain 95% Graphitic C at 85% recovery. The inclusion of the -53mm size fraction into the final product was critical to obtain the overall recovery targeted.

Table 1-3: Optimum Flowsheet Graphite Concentrate Size-by-size analysis (Weathered)

Size Fraction (mm)	% Mass	% Passing	%C as graphite	% S
>212	6.7	93.3	97.9	0.06
150 – 212	17.1	76.2	97.8	0.06
106 – 150	21.8	54.5	97.9	0.05
75 – 106	21.8	32.8	97.3	0.09
<75	32.8		94.8	0.16
Combined			96.7	0.10

1.12.3.1.2 Comminution Characterization

Samples of Weathered and Fresh material were used for various hardness tests conducted by JK Tech and MINTEK. The following tests were conducted for the hardness characterization and subsequent use in comminution circuit design criteria:

- JK Teck Dropweight tests
- SAG Milling Comminution tests (SMC)
- Uniaxial Compressive Strength tests (UCS)
- Bond Crushability tests (CWi)
- Bond Abrasion Index tests (Ai)
- Bond Rod Mill Work Index tests (BRWi)
- Bond Ball Mill Work Index tests (BBWi)

The results of the hardness test conducted in 2015 forms the basis of the process design criteria for the comminution circuit design and are presented in Table 1-4.

Table 1-4: Summarized Hardness Indices

Hardness Test	Samples	Result
JK Tech Dropweight	Weathered	Axb = 85.5 (Soft)
SAG Milling Comminution Tests (SMC)	Weathered and Fresh	Weathered Axb = 57.6 (Moderate) Fresh Axb = 57.2 (Moderate)
Uni-Axial Compressive Strength (UCS)	Weathered	UCS = 65.5MPa (Soft)
Bond Crushability Index (CWi)	Weathered	CWi = 12.0kWh/t (Soft)
Bond Abrasion Index (Ai)	Weathered and Fresh	Weathered Ai = 0.23 (Slightly Abrasive) Fresh Ai = 0.28 (Slightly Abrasive)
Bond Rod Mill Work Index (BRWi)	Weathered and Fresh	Weathered BRWi = 11.60kWh/t (Medium Hard) Fresh BRWi = 9.48kWh/t (Soft-Medium Hard)
Bond Ball Mill Work Index (BBWi)	Weathered and Fresh	Weathered BBWi = 12.06kWh/t (Medium Hard) Fresh BBWi = 10.51kWh/t (Soft-Medium Hard)

1.12.3.1.3 Mineralogy

A representative pulverized sub-sample was analysed using qualitative X-Ray diffraction (XRD) to obtain the bulk mineralogical assemblage. Polished sections of each size were prepared for examination by optical microscope with the aim of establishing textures and liberation characteristics of the graphite flakes and their association with gangue minerals.

Test work products were analysed using the following methods: Total C organic C, proximate analysis (high grade concentrates) and inductively coupled plasma optical emission spectrometry (ICP-OES) on feed samples.

The bulk mineralogical composition of the crushed weathered feed sample indicates the major gangue minerals to be quartz, mica and feldspar in varying amounts. Sulphur is contributed as a result of the presence of various sulphides including chalcopyrite, pyrite and pyrrhotite. In addition, sulphur is contributed by jarosite, which is an oxidation product of these sulphides. Rutile occurs in amounts less than the detection limit for XRD, but the presence of rutile was confirmed using scanning electron microscope (SEM) analysis as less than 1% by mass.

1.12.3.1.4 Liberation

Liberation characterization of the crushed samples indicated that graphite flake liberation is inversely related to flake size with ~50% liberation in the 500-850µm size and greater than 90% liberation in the minus 200µm flake size. Rod milling tests indicated that graphite flake liberation increases with milling time.

Gangue minerals observed in the concentrate include clay (kaolinite), quartz, plagioclase, K-feldspar, pyrite and goethite. Scanning electron microscopy shows that gangue occurs interlayered between the graphite flakes and delamination may not separate all the gangue minerals from graphite. Separation of this type of gangue might be possible by chemical leaching.

Table 1-5: Liberation Characteristics of Graphite Flakes in 1st Cleaner Concentrate (MINTEK Base Case Flowsheet)

Size Fraction (µm)	% Fully Liberated	% Liberated + Inclusions	% Attached	% Locked
>500	77.2	17.7	3.0	2.0
425 – 500	76.7	20.1	2.0	1.2
300 -425	75.8	21.9	1.4	0.9
212 – 300	80.8	17.8	1.2	0.2
150 – 212	81.2	18.1	0.7	0.0
106 – 150	77.2	22.2	0.3	0.2
53 – 106	84.3	15.7	0.0	0.0
45 – 53	80.2	19.3	0.6	0.0
38 – 45	86.4	13.0	0.6	0.0
25 – 38	89.9	10.1	0.0	0.0
<25	93.7	6.3	0.0	0.0

1.12.3.1.5 Conclusions and Recommendations of the Test program

- 200µm was selected as the optimum liberation size based on the mineralogy alone. However, to preserve coarse graphite flakes, 80%-425mm grind was selected as the optimum graphitic carbon liberation grind.
- A large proportion of graphite reported to the -53mm size fraction (down to -25mm), pointing out that it will not be possible to produce a discardable barren fines fraction.
- Sulphur grades in the weathered material was shown to not be of concern, since grades repeatedly reported well below the targeted 0.1% S.
- It is possible to produce a final concentrate product at 95% Graphitic Carbon content and with a recovery of 85% when processing (floating) the -53mm fraction.
- All flotation tests were performed using frother only (Dow200).
- Reagent optimisation test work, pH control test work and Mineralogy on the rougher tails of the fresh mineralized material is recommended to determine if recoveries could be improved.

1.12.3.2 Flowsheet Confirmation Test Work, METPRO/SGS 2021

Data reviewed by METPRO suggested that a more conventional graphite flotation flowsheet similar to the Bissett Creek process would be more suitable for the Okanjande mineralization. It was postulated that two polishing stages may be beneficial to maximize the +180µm yield. The suitability of the proposed revised flowsheet was evaluated in a series of seven cleaner flotation tests that were carried out at SGS Lakefield using one weathered

and one fresh sample, which were shipped from site. The weathered sample was a composite of four random samples taken from the mineralized material stockpile at the Okorusu plant. The fresh sample was a composite taken from six drill holes at depths between 55 and 71m.

Process variables that were investigated in the flotation program included:

- Elimination of ball mill and scavenger flotation
- Single and double polishing mill
- Classification of intermediate concentrate and SMM grinding
- Cleaning circuit without classification of intermediate concentrate

All tests employed a simplified reagent regime of diesel and MIBC without the use of a gangue dispersant. An aliphatic alcohol frother such as MIBC has the benefit of a less persistent froth, which reduces the risk of elevated gangue entrainment.

The SGS program matched and exceeded the results of the original Mintek work and confirmed that the current plant flowsheet is not suitable for treating the Okanjande mineralization.

Table 1-6: A Comparison of Graphite Flake Distribution between Plant and Lab (METPRO, 2021)

Flake Size (µm)	Okorusu Plant 2017-2018 (%)	MINTEK Lab tests (%)	MINTEK Pilot Plant (%)	SGS Weathered tests (%)	SGS Fresh Test (%)
>300	2	5	13	10	21
>180	16	35	35	41	46
>106	37	26	23	30	20
<106	45	34	29	19	13

1.12.3.3 Tailings Filtration Test Work

Relocation of the processing plant to the Okanjande mine site necessitated a new approach for tailings disposal. Due to the limited water supply available at the Okanjande site, dry tailings disposal was selected. A plate and frame filter press plant was selected for dewatering duty, due to the ability to achieve very low moisture in discharged filter cake.

The tailings samples were shown to be amenable to plate and frame filtration with no pre-thickening. A slurry of 22% solids content was filtered through the test unit in 11.17 minutes and achieved a final cake moisture of 16.1% solids after a cake blow of 5 minutes.

1.12.4 Sample Representativity

The Weathered samples were collected as grab samples from crushed mineralized material (-13mm) stockpiles in May 2021. The Fresh samples were collected in July 2021 and were prepared from a blend of 6 different drillholes that intersected the fresh mineralized material zone.

1.12.5 Mineral and Metallurgical Test Work Conclusions & Recommendations

The mineral processing test work program completed at SGS in 2021 indicated that the new flow sheet could improve on the existing flow sheet and corresponded well with early test work and pilot campaigns conducted in the 1990's.

Based on the latest test work conducted improved recovery and flake distribution was achieved when compared to the test work conducted by MINTEK in the 1990's and 2015.

Based on the completed SGS test work program the design criteria for the flotation plant modifications were developed.

Table 1-7: Expected Concentrate Grade and Flake Size Distribution from new process flow sheet

Product	Average		Range	
	%C(t)	% Distribution	%C(t)	% Distribution
Composite				
Composite	96.0	92.0	95.0 – 96.5	91.0 – 93.0
Concentrate Composition				
>300µm	96.0	9	95.0 – 97.0	7 – 11
300 – 180µm	96.0	40	95.0 – 97.0	38 – 42
180 – 106µm	96.0	30	95.0 – 97.0	28 – 32
<106µm	96.0	21	94.0 – 96.0	19 – 23

While the mineral processing test work leading up to this PEA Report supported the expectation for much better performance in terms of graphite flake purity and flake size distribution, it is recommended to conduct further test work to investigate:

- **Concentrate upgrading with gravity separation.** This would be beneficial to possible difficulties with graphite / mica separation by flotation as the only separation method, assuming that the mica contamination is not due to interstitial mica mineralization. If required, the newly redundant Okorusu gravity spirals can be brought back into operation with relative ease and low cost.
- **Concentrate upgrading by chemical means.** As per the early test work discussion a measure of success was achieved with various schemes of roasting followed by acidic leaching. This could potentially be a method to upgrade marginal concentrates or concentrates failing to achieve grade.

1.13 MINERAL RESOURCE ESTIMATES

1.13.1 Mineral Resource Data

MSA received all the drillhole data from Imerys-Gecko in Excel spreadsheets including data pertaining to:

- Collars;
- Sample assays;
- Hole inclination and direction (based on set up orientation);

- Geology logging; and
- Density.

The drilling data are from percussion and diamond drillholes completed at the Okanjande graphite deposit and the surrounding, satellite deposits (Table 1-8).

Table 1-8: Summary of the drillhole data (MSA, 2021)

Deposit	Drillhole Series	Campaign	Type	Number of drillholes
Okanjande Main	OAD03 – OAD13; OAD15 – OAD27; OAP10 – OAP75; OAP81; OAW01 - OAW14	RUL	Percussion and Diamond	105
Northern Mineralized Band	OAP76 – OAP78	RUL	Percussion	4
Okanjande Main (including North Mineralized Extension and Southern Mineralized Extension)	OAD19DT - OAD75D; OKD005 - OKD050; OPH001 - OPH009; SZH002 - SZH004; SZH006	Gecko	Diamond	72
Northern Mineralized Band	OKD051 – OKD063	Gecko	Diamond	13
North-east of Okanjande Main	OKD001 – OKD004	Gecko	Diamond	4
Okanjande Main (including North Mineralized Extension and Southern Mineralized Extension)	OKJ1 – OKJ12; OkaPBH_020 – OkaPBH_026	Imerys-Gecko	Diamond	19
Northern Mineralized Band	OkaPBH_027 – OkaPBH_034	Imerys-Gecko	Diamond	6
SW Target	OkaPBH_013 – OkaPBH_019	Imerys-Gecko	Diamond	7
Highlands Target	OkaPBH_009 – OkaPBH_012	Imerys-Gecko	Diamond	4
Old Mine	OkaPBH_007 – OkaPBH_008	Imerys-Gecko	Diamond	2
Leopardskloof Target	OkaPBH_001, OkaPBH_002, OkaPBH_005	Imerys-Gecko	Diamond	3
Rooibult Target	OkaPBH_035 – OkaPBH_037	Imerys-Gecko	Diamond	3

1.13.2 Exploratory Data Analysis

The data for the Okanjande Main deposit includes drilling data collected by RUL, Gecko and Imerys-Gecko. Generally, the drillholes are spaced 50m apart with drill lines at the central area of the deposit being 25m apart. The data comprises 81 percussion holes and 115 diamond holes. The TGC grade distribution is mixed due to the interbedded mineralised and non-mineralised layers in the graphitic gneiss/quartzite (Figure 1-4).

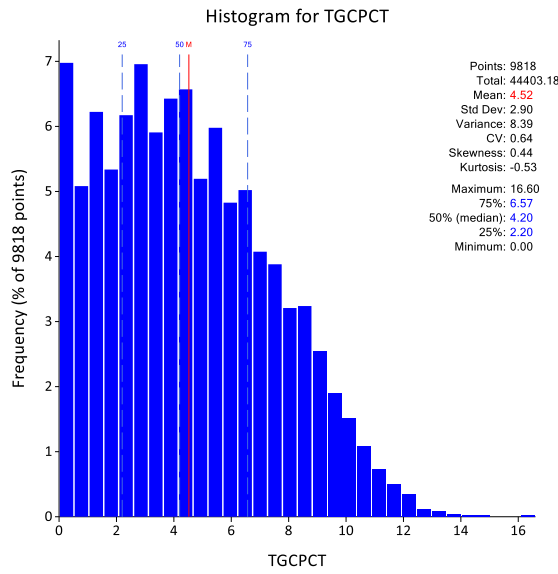


Figure 1-4: Histogram of un-composited lengths for TGC

1.13.3 Bivariate Analysis

The bivariate plot between TGC and S assays shows two groups of weakly positive correlations; one for weathered and the other for fresh rock (Figure 1-5).

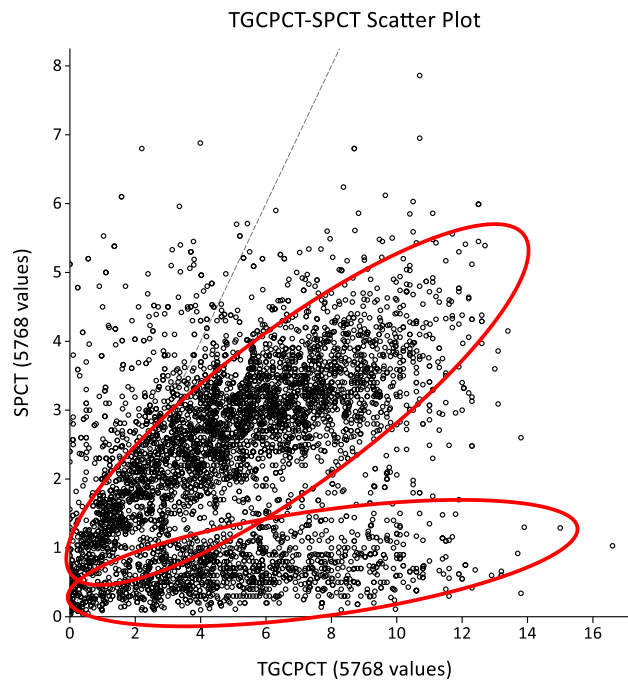


Figure 1-5: Bivariate plot between TGC and S (MSA, 2021)

1.13.4 Geological Modelling

The graphite at Okanjande Main is mainly contained within graphitic gneisses and quartzite while the un-mineralised rocks are pegmatite veins or interbedded un-mineralized quartzite. The deposit was modelled as a simple volume with the lateral extent constrained by un-mineralised drillholes. The lateral extent was modelled halfway between mineralised and un-mineralised drillholes or extrapolated up to a 100m from a mineralised, sampled and assayed drillhole (Figure 1-6).

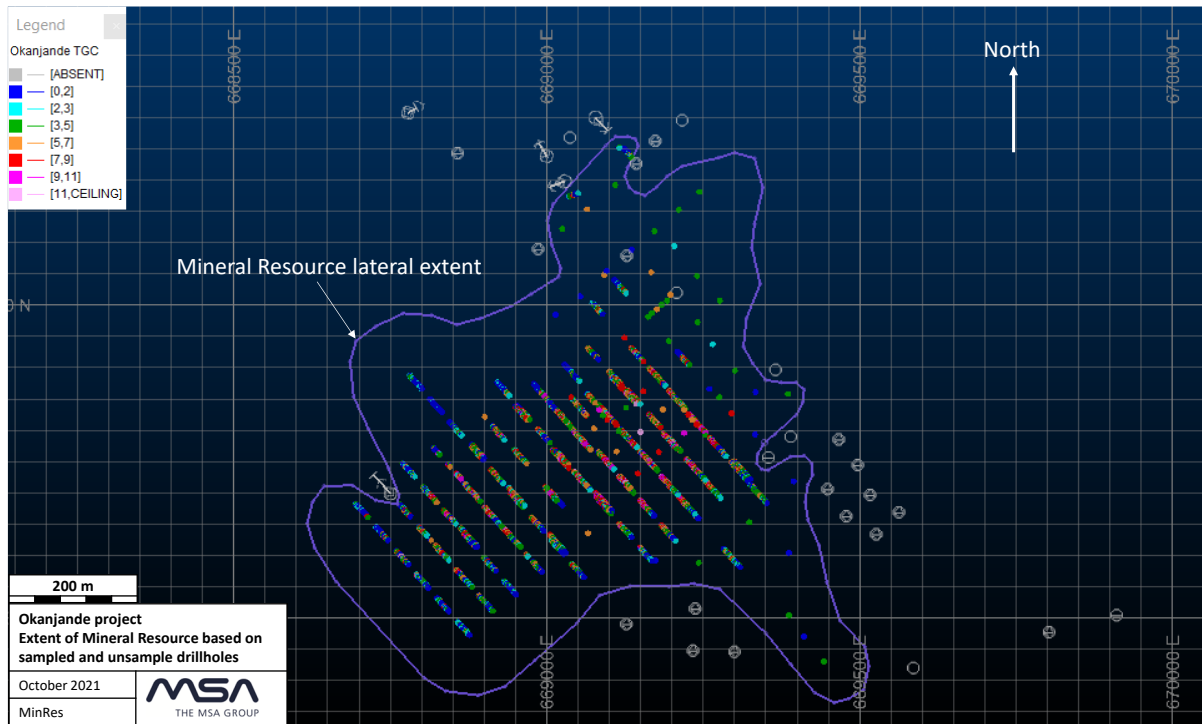


Figure 1-6: Plan view of Mineral Resource extent (MSA, 2021)

1.13.5 Estimation Domains

Weathering domains form the basis for sulphur grade estimation; therefore, three estimation domains were used for the sulphur grade estimation (Table 1-9). The TGC grades have a similar grade distribution for both the weathered and transitional domains. The higher average grade in the fresh domain is because of a high-grade zone at depth.

Table 1-9: Sulfur estimation domains (MSA, 2021)

Weathering Domain	Explanation
1	Weathered
2	Transitional
3	Fresh

1.13.6 Composite Data

The drillhole samples were composited to 1m because this is the most occurring sample length accounting for the majority of the data. The TGC sample grade population shows a mixed distribution, with a high frequency of composite samples with a grade of 0% TGC due to the zero values applied where samples were not taken (Figure 1-7: Composite TGC grade histogram (MSA, 2021)). The mixed distribution is due to the deposit being made up of alternating mineralised and un-mineralised layers, which could not be modelled as separate domains. The deposit also has a high-grade core, which transitions into lower grades towards the peripheries.

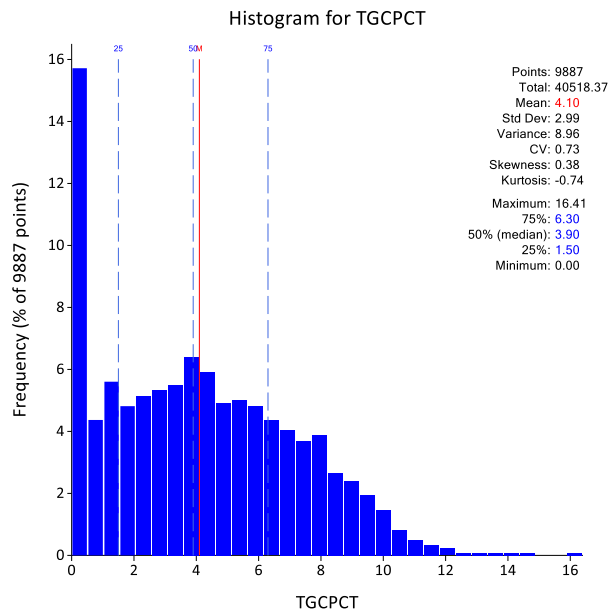


Figure 1-7: Composite TGC grade histogram (MSA, 2021)

1.13.7 Geostatistical Analysis

A single variogram was modelled for TGC and two variograms were modelled for S grade; one for weathered and the other for the fresh domain. A variogram for the transitional domain could not be modelled reliably and the variogram for the fresh domain was used in estimating the transitional domain.

The variograms demonstrate strong continuity in excess of the drillhole spacing.

Table 1-10: TGC and S variogram parameters (MSA, 2021)

Variable	Domain	*Rotation angles			Nugget Effect	Sill 1	Range 1 (m)			Sill 2	Range 2 (m)		
		X	Y	Z			X	Y	Z		X	Y	Z
TGC	-	140	20	180	0.11	0.56	40	40	5	0.33	230	195	25
Sulphur	Weathered	0	0	0	0.13	0.63	80	100	6	0.24	525	220	13
	Fresh	140	20	180	0.10	0.63	50	30	7	0.27	440	195	60

Note: *Rotation angles were applied to XZX.
Ranges are rounded off to the nearest 5m.

1.13.8 Block Modelling

A kriging neighbourhood analysis was completed using the TGC variogram to optimise the block size as well as other kriging parameters.

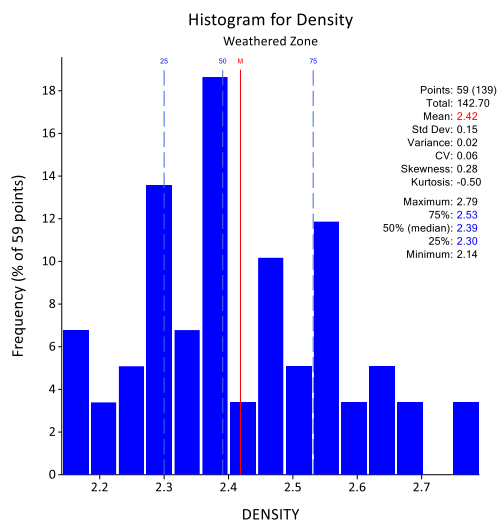
A block model was created with a block size of 20mX by 20mY by 5mZ, which is two fifths of the average drillhole spacing. The block model was sub-celled to the minimum dimensions of 5mX by 5mY by 2.5mZ. The block model was coded by weathering.

1.13.9 Estimation

The TGC and S grades were estimated into the block model using ordinary kriging into parent cells. TGC grades were estimated using a search ellipse of 100m by 80m by 20m. The transitional and fresh S grades were estimated using the same search ellipse and the weathered S grades were estimated using a search ellipse of 100 m by 50 m by 5 m in line with the variogram range proportions. The grades were estimated with a minimum number of 12 composites and a maximum number of 18 composites.

1.13.10 Density

A total of 144 density measurements were completed in October 2021 at the Okanjande deposit using the remaining core from previous drilling. The data were collected from 11 holes located in the central and southern parts of the deposit. Average density values were applied to the block model according to weathering.



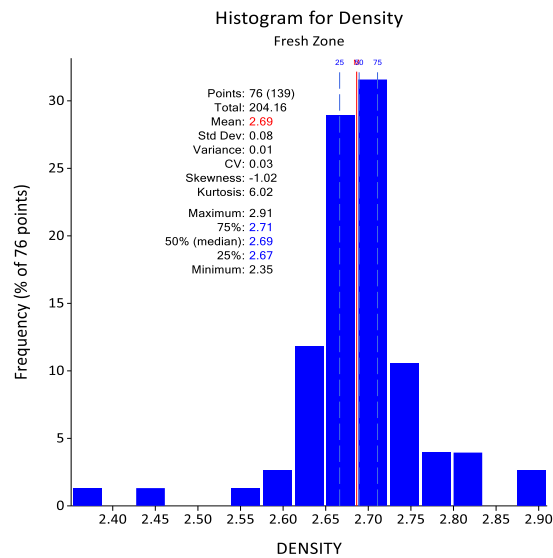


Figure 1-8: Density histograms per weathering domain (MSA, 2021)

1.13.11 Validation of the Estimates

The grade estimates were validated by:

- Visual inspection;
- Global mean comparison; and
- Swath plots.

Visual inspection of the block model shows that the estimated grades are representative of the input composite grades.

The global declustered mean grades of the TGC and sulphur composites were compared to the block model estimated average grades. The relative differences are less than 10% for all domains except sulphur estimates in the transitional zone.

Swath plots demonstrate that the TGC and S estimates follow the trend of the input data, with some level of smoothing that is expected through estimation.

1.13.12 Classification

The estimates were classified into Measured, Indicated and Inferred Mineral Resource categories as defined by the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014). The Mineral Resource is classified on the basis of TGC estimates as the main variable of economic interest.

- Modelled mineralisation covered by a 50m by 50m grid, that also includes the highest confidence Gecko drilling, was classified as Measured Mineral Resources. The Measured Mineral Resource was extrapolated by 25m laterally and by 5m from the base of drillholes at depth;
- Areas drilled at a grid of 50m to 100m spacing were classified as Indicated Mineral Resources. The Indicated Mineral Resource was extrapolated by 50m laterally and by 10m from the base of drillholes at depth; and

- Areas covered by drilling at wider than 100m spacing, or in the north-west where the area appears to be more geologically complex, were classified as Inferred Mineral Resources. Inferred Mineral Resources were extrapolated to the lateral extent of the Mineral Resource model and by 25m vertically from the base of drillholes at depth.

A plan of the classified Mineral Resource is shown in Figure 1-9.

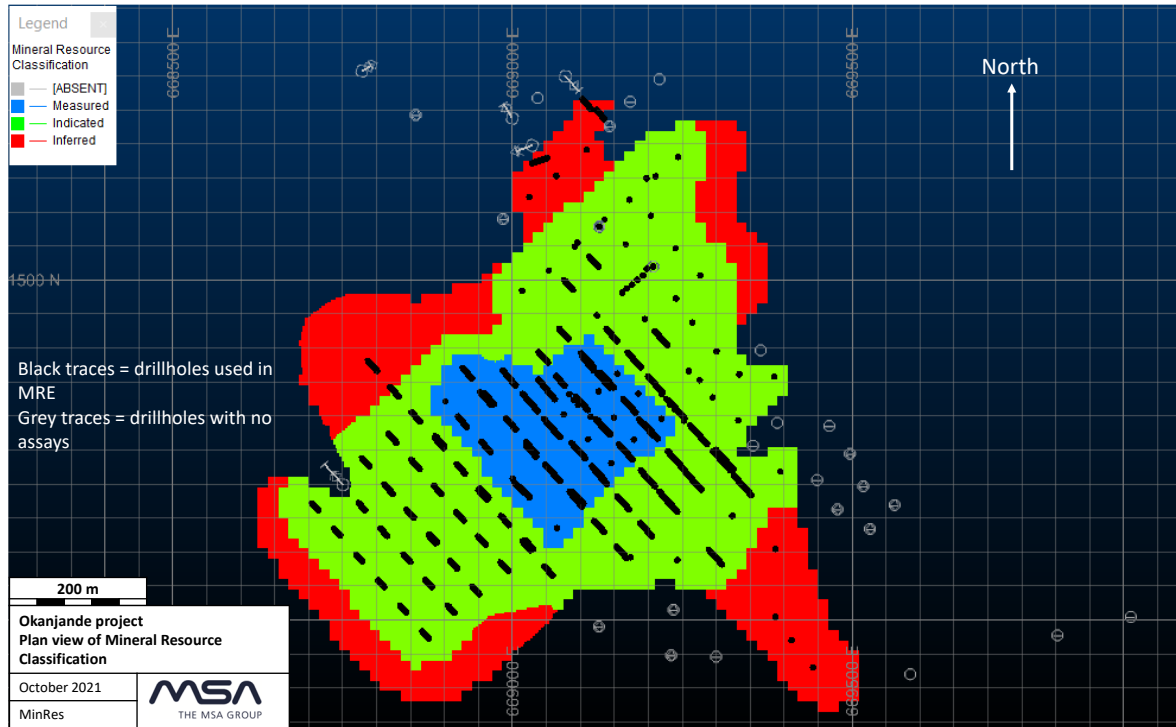


Figure 1-9: Mineral Resource Classification (MSA, 2021)

1.13.13 Reconciliation

Mining at Okanjande took place from early 2017 to October 2018. As part of grade control during mining, blast holes were drilled vertically at a spacing of 3m. MSA completed a reconciliation against the Mineral Resource model in the same areas that the blast holes were drilled (Figure 1-10). On average, the blast holes and Mineral Resource model TGC grades compare well, with a relative difference of -4%.

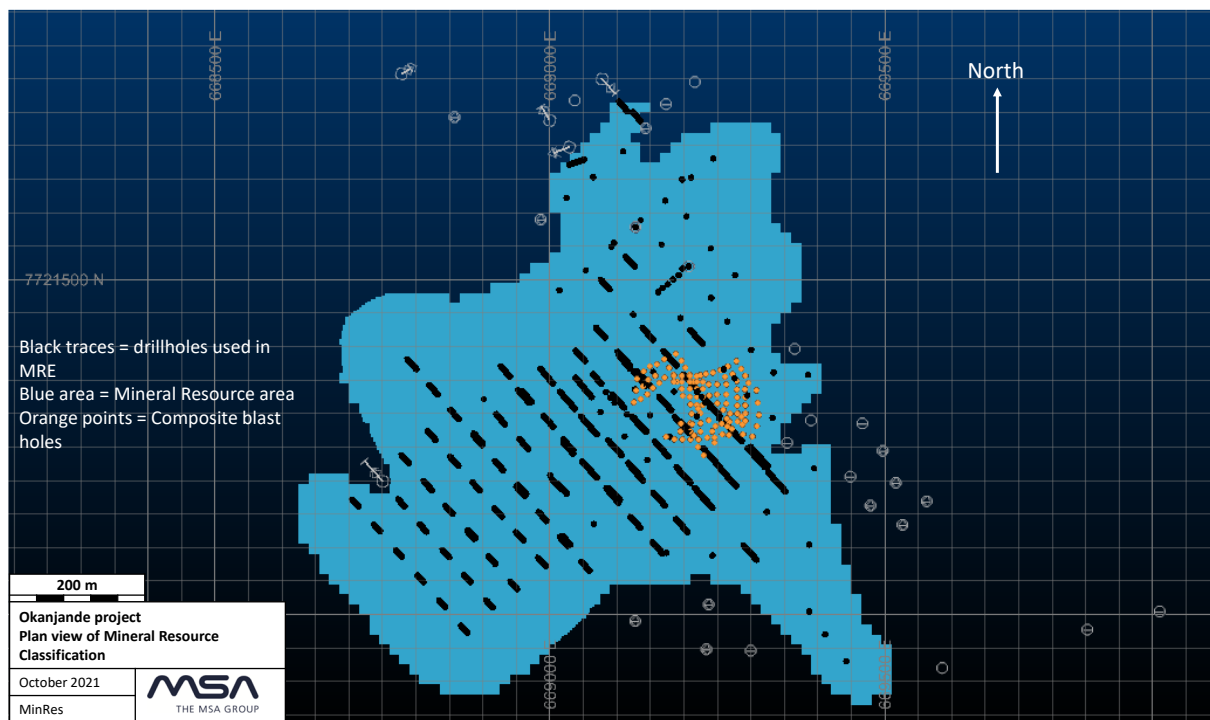


Figure 1-10: Plan view of the Mineral Resource reconciliation (MSA, 2021)

1.13.14 Reasonable Prospects for Eventual Economic Extraction (RPEEE)

MSA carried out a high-level economic analysis to determine “Reasonable Prospects for Eventual Economic Extraction” (“RPEEE”) as required by CIM Definition Standards for Mineral Resources and Mineral Reserves (2014).

1.13.15 Mineral Resource Statement

The Weathered Mineral Resource, as at November 29, 2021, is reported in accordance with the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014), as reported in Table 1-11.

Table 1-11: Weathered Mineral Resource at a cut-off grade of 2.6% TGC as at 29 November 2021 (MSA, 2021)

Material	Classification Category	Tonnes (Mt)	TGC (%)	Graphite content (kilo tonnes)
Weathered	Measured	1.7	4.66	80
	Indicated	4.2	4.03	168
	Subtotal	5.9	4.21	248
	Inferred	0.5	3.45	17
Transitional	Measured	0.2	5.42	13
	Indicated	1.0	4.08	40
	Subtotal	1.2	4.35	53
	Inferred	0.1	3.20	2

Notes:

1. All tabulated data have been rounded and as a result minor computational errors may occur.
2. Mineral Resources which are not Mineral Reserves have no demonstrated economic viability.
3. Inferred Mineral Resources are reported separately from other categories.
4. The Mineral Resources reported are the total Mineral Resources for the Project, regardless of ownership.
5. The Mineral Resource is reported for mineralization contained within pit shells above a cut-off grade of 2.6% TGC, which is based on a product price of USD 1,250/t TGC, mining costs of USD 3.75/t RoM, transport cost to plant of USD 6.5/t RoM, processing and treatment costs of 14.9 USD/t (RoM), G&A USD 0.8/t (RoM), transport cost to the market USD 175/t product, 2% royalty, concentrate recovery 92%.
6. MSA is not aware of any known environmental, permitting, legal, title-related, taxation, socio-political, marketing, or other relevant issue that could materially affect the Mineral Resource Estimate.
7. The Inferred Mineral Resource has a lower level of confidence than that applying to an Indicated Mineral Resource and must not be converted to a Mineral Reserve. It is reasonably expected that the majority of Inferred Mineral Resources could be upgraded to Indicated Mineral Resources with continued exploration.

The Fresh Mineral Resource, as at 29 November 2021, is reported in accordance with the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014), as reported in Table 1-12.

Table 1-12: Fresh Mineral Resource at a cut-off grade of 3.1% TGC as at 29 November 2021 (MSA, 2021)

Classification Category	Tonnes (Mt)	TGC (%)	Graphite content (kilo tonnes)
Measured	7.1	5.86	419
Indicated	17.0	5.10	868
Subtotal	24.2	5.33	1,287
Inferred	7.2	5.02	359

Notes:

1. All tabulated data have been rounded and as a result minor computational errors may occur.
2. Mineral Resources which are not Mineral Reserves have no demonstrated economic viability.
3. Inferred Mineral Resources are reported separately from other categories.
4. The Mineral Resources reported are the total Mineral Resources for the Project, regardless of ownership.
5. The Mineral Resource is reported for mineralization contained within a Whittle pit shell above a cut-off grade of 3.1% TGC, which is based on a product price of USD 1,250/t TGC, mining costs of USD 5.11/t RoM, transport cost to plant of USD 6.5/t RoM, processing and treatment costs of 17.88 USD/t (RoM), G&A USD 0.96/t (RoM), transport cost to the market USD 175/t product, 2% royalty, concentrate recovery 92%.
6. MSA is not aware of any known environmental, permitting, legal, title-related, taxation, socio-political, marketing, or other relevant issue that could materially affect the Mineral Resource Estimate.
7. The Inferred Mineral Resource has a lower level of confidence than that applying to an Indicated Mineral Resource and must not be converted to a Mineral Reserve. It is reasonably expected that the majority of Inferred Mineral Resources could be upgraded to Indicated Mineral Resources with continued exploration.

1.14 MINERAL RESERVE ESTIMATES

No mineral reserve estimate was determined or is available at the time of writing.

1.15 MINING METHODS

1.15.1 Background

The Okanjande graphite deposit lends itself to being mined by opencast mining methods. The intention is to mine at a rate of 631,450 tonnes of mineralised material per month which.

This report considers a 10-year mining plan, commencing with the mining of the weathered graphite-bearing material and ending with a short period of mining the transitional and fresh graphite-bearing material underlying the weathered material.

The intention is to mine the Okanjande deposit by making use of contractors who will supply their own mining equipment and to deliver the graphite-bearing material to a product stockpile and the balance to a waste rock dump.

1.15.2 Mine Scheduling

A pit shell was defined incorporating only the weathered graphite-bearing material and using geotechnical parameters as defined in the previous mine design contemplated by Imerys-Gecko. The pit shell was then reconfigured and expanded to include some of the transitional and fresh graphite-bearing material in order to achieve a 10-year life of mine at a production rate of approximately 631,450tpa graphite-bearing material to achieve approximately 31,103tonnes of product per annum.

1.15.2.1 Pit Design Parameters

The open pit design parameters considered are summarised below:

- Bench heights: 5m
- Bench widths: 10m to 70m
- Slope angles: 45° to 55°
- Cut-off Grade: 4% TGC
- Dilution to RoM tonnes: 2%
- Recovery: 92%.

1.15.2.2 Schedules

1.15.2.2.1 Base Case 4% TGC Cut-off

Using the above design parameters, a mining schedule was developed to mine at a rate of approximately 630,000tpa at an average Stripping Ratio ("S/R") of 0.66 (tonnes of waste to one tonne of graphite-bearing material) over the 10-year period (Table 1-12).

Table 1-13: 10-year mining schedule @3% TGC cut-off (MSA, 2021)

	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	TOTAL
Total Mining (t)	1 419 921	1 154 011	1 715 632	1 401 837	878 519	685 038	738 987	707 697	764 570	675 780	10 141 990
Waste (t)	1 009 479	522 561	1 082 452	770 387	247 069	53 588	105 807	76 247	133 120	44 330	4 045 040
ROM (t)	410 443	631 450	633 180	631 450	631 450	631 450	633 180	631 450	631 450	631 450	6 096 950
TGC %	5.58	4.95	4.81	5.05	4.90	5.06	5.46	5.65	5.47	5.86	5.27
S %	0.79	0.71	0.79	0.95	1.94	2.48	2.54	2.65	2.69	2.90	4.85
S/R	2.46	0.83	1.71	1.22	0.39	0.08	0.17	0.12	0.21	0.07	0.66
Product TGC (t)	21 070	28 756	28 019	29 337	28 453	29 395	31 806	32 823	31 777	34 043	295 480

1.15.2.2 Increasing and Decreasing TGC Cut-off

An exercise was carried out to look at the effect of increasing the cut-off grade to see the effect on the Graphite production, depletion of weathered material and total tonnes moved. Table 1-14 below shows the results for 3, 3.5, 4, 4.5 and 5% TGC cut-offs. The base case of 4%TGC is highlighted.

Table 1-14: 10-year Mining Schedule @ 3, 3.5, 4, 4.5 and 5% TGC cut-off (MSA, 2021)

	3.0%	3.5%	4.0%	4.5%	5.0%
Total Tonnes	8 464 034	9 193 913	10 141 990	11 086 863	11 938 342
Waste Tonnes	2 347 081	3 096 960	4 045 040	4 989 909	5 849 142
ROM Tonnes	6 096 953	6 096 953	6 096 950	6 096 953	6 089 199
%TGC	4.75	5.00	5.27	5.41	5.50
%S	1.40	1.61	1.88	2.11	2.48
S/R	0.38	0.51	0.66	0.82	0.96
Product Tonnes	266 338	280 259	295 480	303 441	308 119

1.15.3 Mining Recovery

Given the nature of the Okanjande deposit, 92% recovery with 2% dilution is anticipated, the dilution is based on barren rock within the Okanjande deposit which is randomly dispersed and differs in thickness and orientation throughout. The Strip ratio of the Okanjande deposit is extremely low for the selected 4% TGC cut-off and the relatively small amount of waste rock is stockpiled on a dedicated rock dump.

1.15.4 Mining Equipment

The following mining equipment is proposed to undertake the mining of the graphite-bearing material and waste at the Okanjande deposit (Table 1-15).

Table 1-15: Mining Equipment

Quantity	Description	Size / Capacity
2	Excavator 60 t	60 t
5	Articulated Dump Trucks ("ADTs")	30 t
1	Drill Rig	Top Hammer 102mm diameter
1	Track Dozer	40 t
1	Motor Grader	CAT140/Bell 770
1	Water Bowser	15,000 l
1	Lubrication Service Truck	
2	Front End Loader ("FEL")	3 -5 m ³
1	Diesel Bowser	15,000 l

1.15.4.1 Drilling and Blasting

To achieve this powder factor with 102mm holes and 5m high benches, each bench is drilled on a 3m by 3m pattern and charged with an emulsion explosive agent.

1.15.4.2 Mining Cost

Mining costs for the operation were obtained from three budget quotations provided by mining contractors in Namibia. A review of the costs by URE Consulting Inc concludes that costs provided by one of the contractors be used in the economic assessment of the Okanjande deposit and that closer to the start of mining, two of the three contractors be asked for full and final quotes.

Table 1-16: Mining and Transport Costs (Namibian Dollar)

Operating Costs	Units	Cost
Establishment and De-establishment	N\$, Once-off	3,064,250
P&Gs	N\$/a	12,862,591
Load, haul and dump topsoil	N\$/m ³	51.45
Drill & blast (waste and mineralized material)	N\$/m ³	43.84
Load, haul and dump waste	N\$/m ³	51.45
Load haul and stockpile mineralized material	N\$/t	56.47

1.15.5 Integrated Waste Dump Design (IWD)

Waste Rock from the Okanjande mining operations will be co-deposited with dry (filtered) tailings from the flotation process on a facility located to the northeast of the process plant. A dry tailings system was selected to minimize water consumption for the project. The IWD will be lined with a double HDPE liner with leak detection. Run-off water from the IWD is channelled to a lined collection pond, where run-off may be neutralized and recovered to the process. The location of the IWD was chosen over the outcrop of Swakop Group marble band which is located in north-western corner of the mining license. This footing will minimize penetration to ground

water and provide an additional measure of neutralization in the case that any potentially acid generating drainage or run-off should seep out of the IWD.

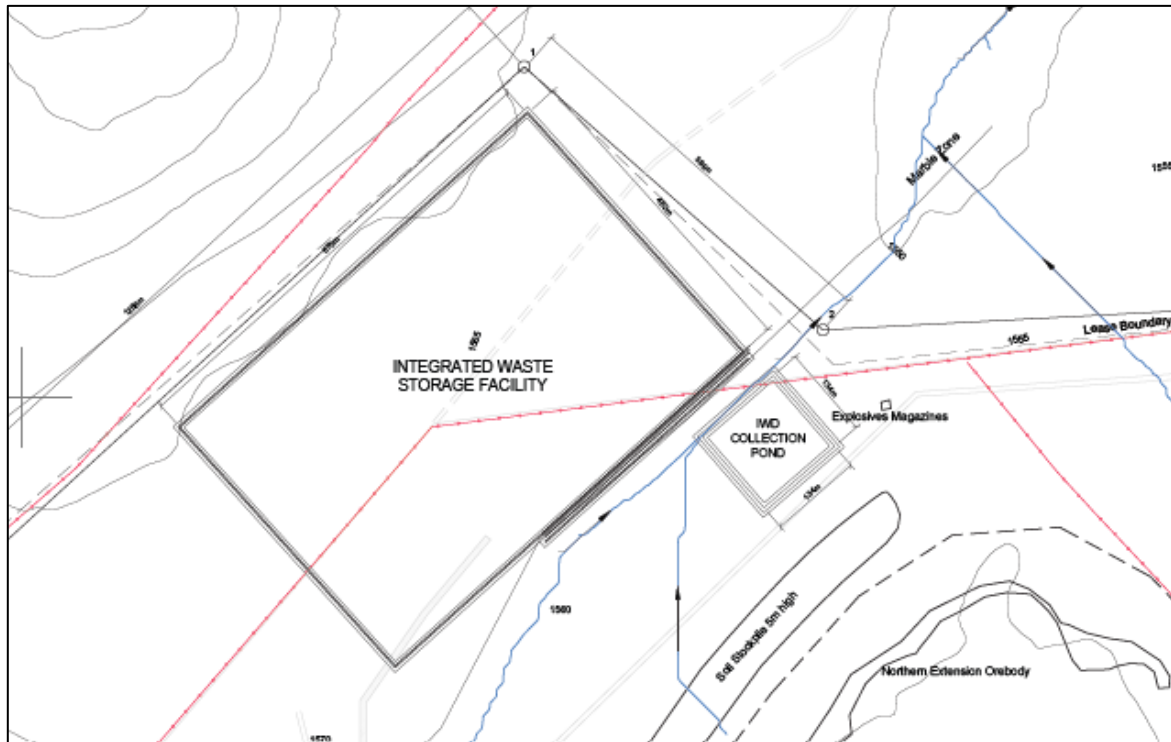


Figure 1-11: Okanjande WRD conceptual footprint

1.15.6 Conclusions

This 10-year mining plan (4%TGC cut-off) considers the mining of weathered as well as fresh graphite bearing material. The IWD design is suitable for all waste and ore rock types. All testing to date indicates that the same processing flowsheet can be used for both fresh and hard rock.

If the cut-off grade is increased, the above adjustments to process the fresh graphite bearing material will be required sooner.

1.16 RECOVERY METHODS

1.16.1 Previous Imerys-Gecko Process

The existing graphite processing plant at Okorusu, which was used to process Okanjande graphite mineralized material during 2017-2018 is based somewhat on the Imerys Lac de lles graphite operation in Quebec, Canada. This process relied heavily on the use of gravity spirals early on in the process. While spirals can be a suitable technology to separate gangue from graphite, it is paramount that the graphite is well liberated to produce an acceptable concentrate. This was not the case for the Okanjande graphite, and the plant struggled operationally to achieve nameplate production rates and concentrate grades.

1.16.2 Current Northern Graphite Process Design

1.16.2.1 Design Basis

The process plant will be relocated to the Okanjande Mine site. The relocation of the process plant will achieve 4 main benefits:

- Reduction of operating costs through elimination of ~90km of mineralized material hauling to the Okoruso site. Mineralized material hauling costs previously contributed to 15% of the overall C1 operational costs.
- Elimination of space constraints. The new process plant location at the Okanjande mine site will allow for unhindered expansions, which are planned for the future. The Okoruso site was space constrained and did not allow for any expansion.
- Tailings Storage. Tailings storage at the Okoruso site was limited and would only allow for storage of 1.2Mt of tailings. The Okanjande site has sufficient real estate for the construction of a tailings facility to accommodate life of mine operations.
- Reduction in Carbon footprint through the removal of the road haulage requirement to the Okoruso site and implementation of solar power generation.

The addition of new comminution and tailings dewatering circuits and the changes to process flow of the existing flotation circuit are the result of incorporation of findings from the outcomes of both early and recent test work programs and are designed to improve flake preservation and improve purity of graphite flake concentrate through the following changes:

- A split milling circuit that will prevent excessive recirculation and reduce the potential for flake degradation.
- The split milling circuit will make provision for potentially implementing a flash flotation stage after the Autogenous Mill (AG mill) to further improve large flake recovery.
- Generation of a flotation feed stream of the optimal solids content – 30%.
- A Flotation Feed Buffer Tank to reduce fluctuations in feed rate and solids content and improve steady operation of the flotation circuit.
- The removal of gravity spirals early in the process flow where incomplete liberation would be expected.
- The removal of excessive recirculating streams which hamper effective capacity of process equipment and complicates process control and plant water balance.
- The reduced flotation solids content will remove the need for dispersants.
- The addition of more suitable frother in the form of MIBC.
- Allowance for reagent addition at all flotation banks.
- Reduction of raw water consumption due to the disposal of dry tailings.

1.16.2.2 Process Description

The Comminution circuit will consist of a single crushing stage followed by an open circuit SAG mill and closed-circuit Rod Mill with hydro-cyclone classification.

The Flotation circuit consists of the existing flotation banks transferred to the new Okanjande processing site. Ground ROM mineralized material at 30% solids will be fed to conditioning tanks where Diesel (collector) and MIBC (frothing agent) will be dosed. Conditioning will be followed by a Rougher stage with rougher tails pumped to a rougher scavenger stage. Tails from all downstream cleaner stages will be treated through a bank of cleaner scavenger cells.

The combined concentrate is passed through the existing Imerys circuit for dewatering through a thickener, belt filter and rotary drier. The dry concentrate is classified through screens into 4 products (+300µm, +180µm, +106µm and -106µm) and bagged.

Rougher and Cleaner Scavenger tails will be pumped to the tailings dewatering section where the tails will be filtered through plate and frame filter presses before being discharged on a stockpile. Filtered tailings will be loaded, hauled and dumped by the contract mining fleet, together with waste rock from mining operations, on the dry co-deposition IWD facility. The Okanjande tailings facility will be located on the north-western edge of the mining license above the band of the Swakop marble group. The tailings facility will be of the dry deposition type. Filtered tailings will be co-deposited with waste rock on the facility.

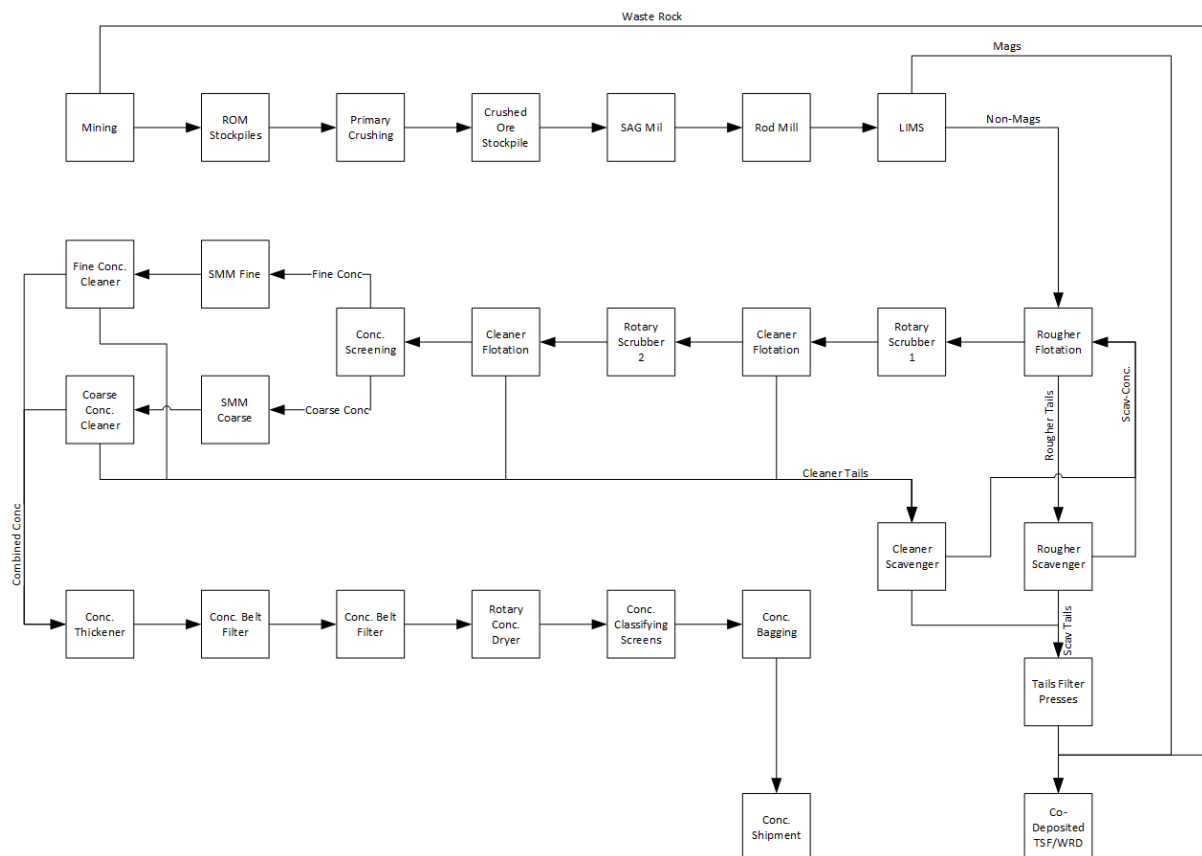


Figure 1-12: Planned Okanjande Process Block-flow Diagram

1.16.2.3 Mass Balance

Table 1-17: High Level Mass Balance

PROJECT KEY PERFORMOMANCE INDICATORS	UOM	Value
Throughput	t/h	87.1
Mean Head Grade	% TGC	4.75%
Total Graphite Concentrate produced	t/h	4.0
Concentrate Grade	%	95%
Total Contained Graphite in Concentrate	t/h	3.8
Total Tailings (dry) deposited	t/h	83.1
Recovery	%	92%

Table 1-18: Graphite Mass Balance (5% Head grade)

Product	Mass	Grade	Distribution
	%	% C(g)	% C(g)
+80 mesh 2nd Clnr Conc	2.22	95.1	42.2
+80 mesh 1st Clnr Tails	0.15	20.0	0.6
-80 mesh 3rd Clnr Conc	2.60	95.1	49.5
-80 mesh 1st Clnr Tails	0.25	30.0	1.5
4th Clnr Tails	0.10	20.0	0.4
3rd Clnr Tails	0.40	8.00	0.6
2nd Clnr Tails	0.75	3.00	0.4
1st Clnr Tails	7.53	0.86	1.3
Rougher Tails	86.0	0.20	3.4
Feed	100.00	5.00	100.0
Combined Products	Mass	Grade	Distribution
	%	% C(g)	% C(g)
Combined Concentrate	4.82	95.1	91.7
+80 mesh 2nd Clnr Conc	2.22	95.1	42.2
-80 mesh 3rd Clnr Conc	2.60	95.1	49.5
4th Clnr Conc	5.22	89.8	93.8
3rd Clnr Conc	5.32	88.5	94.2
2nd Clnr Conc	5.72	82.9	94.8
1st Clnr Conc	6.47	73.6	95.3
Rougher Conc	14.0	34.5	96.6
Feed	100	5.00	100

Combined Products	Mass	Grade	Distribution
	%	% C(g)	% C(g)
Combined Concentrate	4.82	95.1	91.7
+80 mesh 1st Clnr Tails	0.15	20.0	0.6
-80 mesh 1st Clnr Tails	0.25	30.0	1.5
2nd Clnr Tails	1.25	5.95	1.5
1st Clnr Tails	7.53	0.86	1.3
Rougher Tails	86.0	0.20	3.4
Feed	100.00	5.00	100

1.16.3 Production Plan

Run-of-Mine (ROM) mineralized material will be stockpiled on the ROM pad the prior to the crusher. Material will be segregated by grade with blending into the crusher .

Production from the Okanjande mine will ramp up rapidly due to being a brown fields operation with all initial required overburden stripping completed during the previous operational period 2017-2018.

Plant commissioning and ramp-up can be fast tracked during construction through the availability of existing graphite mineralized material stockpiles at the Okorusu site. Plant production rates match mining production rates and stockpile inventory is kept to a minimum.

The run of mine head grade increases with depth with the average grade exceeding 5% from year 7. A flat concentrate graphite grade was assigned to the production plan. However, the increased grade in the last 3 years of operation could likely lead to improved achieved concentrate grade and/or overall graphite recovery.

Table 1-19: Process Production Plan

Production Plan	Units	Totals	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035
Mining															
Mineralized Material Mined (dry)	t/a	6,096,950	0	205,221	520,946	632,315	632,315	631,450	631,450	632,315	632,315	631,450	631,450	315,725	
Grade Mined	%	5.27%	0.00%	5.58%	5.20%	4.88%	4.93%	4.90%	5.06%	5.26%	5.55%	5.56%	5.67%	5.86%	
Waste Mined (dry)	t/a	4,045,040	0	504,740	766,020	802,507	926,420	508,728	150,329	79,698	91,027	104,684	88,725	22,165	
Total Mined (dry)	t/a	10,141,990	0	709,961	1,286,966	1,434,821	1,558,734	1,140,178	781,778	712,013	723,342	736,134	720,175	337,890	
Ore Stockpiles															
Opening															
Mineralized Material Tons	t		0	0	0	0	865	1,729	1,729	1,728	2,593	3,458	3,458	3,458	0
Incoming															
Mineralized Material Tons	t	6,096,950	0	205,221	520,946	632,315	632,315	631,450	631,450	632,315	632,315	631,450	631,450	315,725	0
Outgoing															
Mineralized Material Tons	t	6,096,950	0	205,221	520,946	631,450	631,450	631,450	631,450	631,450	631,450	631,450	631,450	319,183	0
Closing															
Mineralized Material Tons	t		0	0	0	865	1,729	1,729	1,728	2,593	3,458	3,458	3,458	0	0
Processing															
Mineralized Material Processed	t	6,096,950	0	205,221	520,946	631,450	631,450	631,450	631,450	631,450	631,450	631,450	631,450	319,183	0
Grade Processed	%	5.3%	0.0%	5.6%	5.2%	4.9%	4.9%	4.9%	5.1%	5.3%	5.6%	5.6%	5.7%	5.9%	0.0%
Graphite Processed	t	321,174	0	11,451	27,080	30,814	31,129	30,927	31,917	33,212	35,069	35,108	35,768	18,697	0
Concentrate Produced	t	311,031	0	11,090	26,225	29,841	30,146	29,951	30,909	32,164	33,961	34,000	34,639	18,107	0
Tailings															
IWD Deposition (Moist)	t	10,837,335	0	748,021	1,364,293	1,525,583	1,654,312	1,218,496	844,075	770,020	779,779	793,032	775,681	364,043	0

1.16.4 Major Equipment List

The major processing plant equipment is listed in Table 1-20.

Table 1-20: Okanjande Process Plant Major Equipment List

Equipment	Description	Dimensions	Installed Power	Condition
Comminution				
Jaw Crusher	C130 equivalent	1.3x1.0m	160	New
AG Mill		5.0x2.5m	500	New
Rod Mill		3.0x4.5m	600	New
Stack-Sizer	Derrick Screen	5 Deck	75	Refurbished
Buffer Tank		50m ³	20	New
Flotation				
Rougher	Rougher #1&2	12x2.8m ³	266	Refurbished
Polishing Scrubber #1	Scrubber	2.6x7.5m	110	New
1st Cleaner	Cleaner #1	6x1.4m ³	45.5	Refurbished
Polishing Scrubber #2	Mill	1.9x2.7m	228	Refurbished
2nd Cleaner	Cleaner #6	6x1.4m ³	45.5	Refurbished
3rd Cleaner	Cleaner #2	4x1.4m ³	30.5	Refurbished
180mm Classification Screen	SWECO 31, 32, 33		2.4	Refurbished
Dewatering Flotation	Scavenger #3	3x1.4m ³	22.5	Refurbished
Coarse Polishing Mill	Stirred Media Mill		55	New
1st Coarse Cleaner	Cleaner #3	2x1.4m ³	15	Refurbished
2nd Coarse Cleaner	Cleaner #4	2x1.4m ³	15	Refurbished
Fine polishing Mill	Stirred Media Mill		55	New
1st Fine Cleaner	Cleaner #7	4x1.4m ³	30.5	Refurbished
2nd Fine Cleaner	Cleaner #5	2x1.4m ³	15	Refurbished
3rd Fine Cleaner	Cleaner #8	2x1.4m ³	15	Refurbished
Dewatering, Classification and Bagging				
Thickener			2.2	Existing
Belt Filter	Vacuum		7.5	Existing
Dryer	Direct Fired		70	Existing
Vacuum Transfer			18.5	Existing
Filter Extraction Fan			75	Existing
Sizing Screens			11	Existing
Product Silos	4 classifications			Existing
Screens Extraction Fan			37	Existing
Bulk Bagging Stations	4 classifications		48	Existing

1.16.5 Process Plant Layout

The existing Okoruso processing plant including flotation, concentrate dewatering, drying and bagging will be relocated to the Okanjande site. The original layout will be maintained at the new Okanjande site as far as reasonably possible. The overall process plant is aligned in a south to north orientation. See Figure 1-13 below.

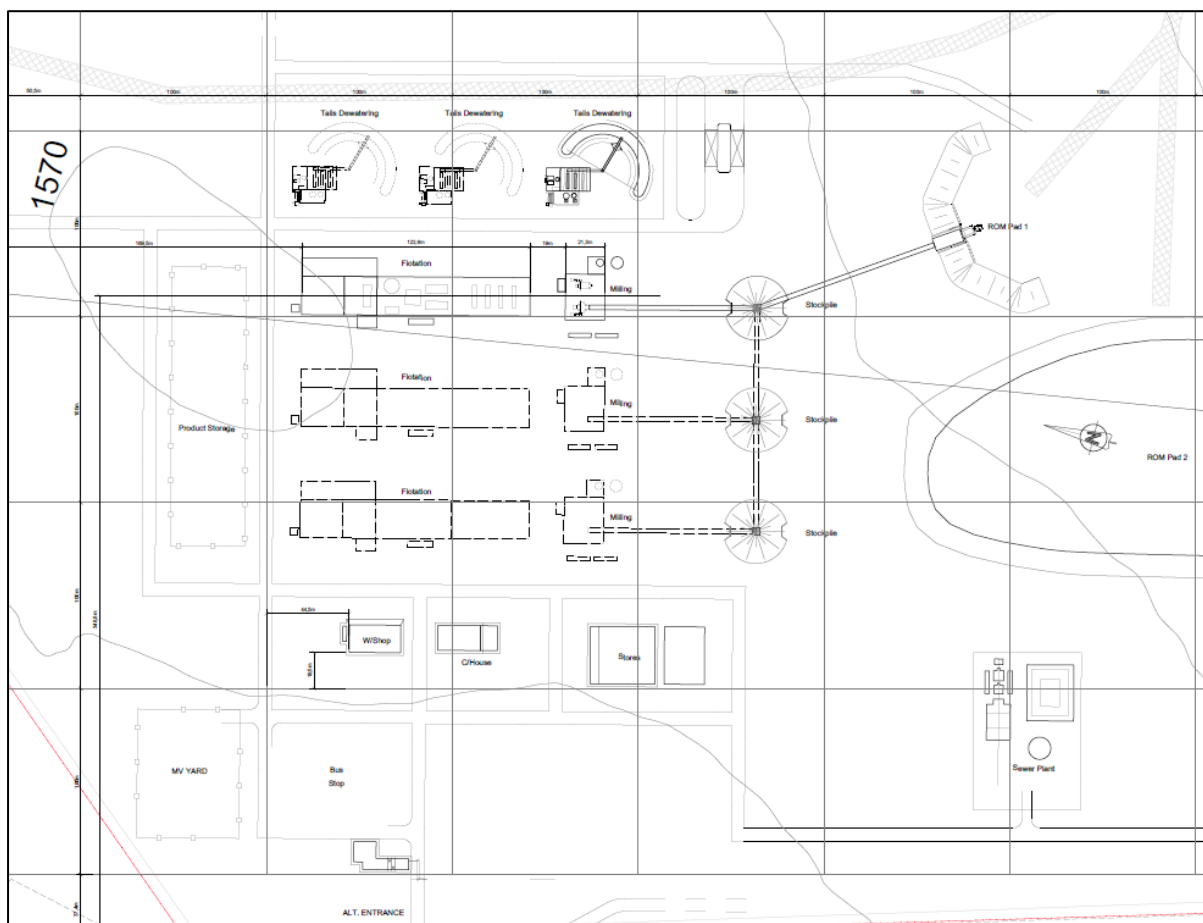


Figure 1-13: Okanjande Process Plant Layout

1.17 PROJECT INFRASTRUCTURE

1.17.1 Access Roads and Haul Roads

The Okanjande deposit is accessed from the B1, 8km south from Otjiwarongo, unto the D2512 (a well-maintained gravel road) for some 24km to the entrance of the property.

Previous mining activities at the Okanjande deposit is evident and the established haul roads can be utilised as soon as operations continue.

1.17.2 Buildings

The mining operations infrastructure located at the Okanjande deposit has adequate and well-maintained operating facilities required for supporting operational restart.

The following new building structures will be erected at the Okanjande site:

- Administration offices
- Operational office
- Warehouse
- Change house
- Site ablution facilities
- Maintenance Workshop
- Mobile equipment workshop

1.17.3 Surface Water and Drainage Collection

The Okanjande project is located between two hills on a relatively flat plain. The process plant is located on a local saddle structure with natural drainage away from the process plant via 2 streams draining towards the north-west and north-east. See Figure 18-2. Where possible all surface run-off will be directed to the two natural drainage systems with berms and channels. The dimensions for each channel will vary based on the peak flow estimates. Where required geo-textile and stone pitching / riprap and gabions will be used to minimize soil erosion.

1.17.4 Sewerage Disposal – Okanjande

A modular small trickling filter sewerage treatment plant (18.75m³/d) will be installed to accommodate for 50 persons and assuming a peak factor of 2.5.

1.17.5 Security

The Okanjande property is adequately fenced off. The existing access gate will be upgraded to accommodate increased traffic and to implement electronic access control.

1.17.6 Water Supply

The water supply will be obtained from two boreholes on Farm Doornlaagte. The boreholes will be installed with two new submersible borehole pumps which will pump water to a new 150m³ reservoir. From the reservoir a base pump station (two centrifugal pumps in a duty plus one standby configuration) will pump the water to Okanjande via a new underground OD 110 mm O-PVC pipeline. The water will be stored in 2 newly installed reservoirs.

1.17.7 Communications

The communications infrastructure will consist of a communications tower connecting the local network to the site as well as housing repeater infrastructure for radio communications on site. The incoming communications network will be distributed internally to the different users for an established office network providing internet connectivity and telephone communication. The internal distribution will contain a combination of fibre optic and copper networks.

1.17.8 Power Supply

No utility power is currently available at the Okanjande property. The power infrastructure will consist of relocating the existing 22kV distribution network in service at the Okoruso processing facility. This includes transformers and medium voltage switchgear.

Power supply will be supplied in two phases, initially from an independent power producer combining renewable (Solar and Battery) and thermal power; followed by a combination of grid power and solar power after 4 years when the grid connection is completed. The plant will be located on the Okanjande site within the existing mining licence boundary.

Grid power from the local supply authority, Nampower, is anticipated to be connected approximately 3 - 4 years thereafter.

1.17.9 Fuel

There is limited fuel storage facilities available on the Okanjande mine site. Diesel storage is available for the site generator powering the administration buildings and workshop only. Mining fleet fuel infrastructure will be provided by the Mining Contractor.

1.17.10 Camp and Canteen

No room or board facilities will be built for this project and no site located facilities are available to be utilised for operational personnel or construction crew. It is recommended to accommodate staff in Otjiwarongo and set up transport to and from site for the Okanjande site.

1.17.11 Fire Protection

The Okanjande fire protection system consists of the following elements:

- Fire Water Header Tank
- Diesel Driven Booster Pump Station
- Firefighting Water Reticulation
- Firefighting Equipment (Fire Hydrants, Hose Reels, Fire Extinguishers)
- Smoke Ventilation (Flotation Plant, Workshops, Warehouse)

1.18 MARKET STUDIES AND CONTRACTS

Because of supply concerns relating to China being the world's dominant producer, and to potential demand growth from new applications such as lithium-ion batteries ("LiBs"), the European Union considers graphite to be one of 14 "critical mineral raw materials" with high supply risk. The United States has also included graphite on a list of mineral resources whose loss could critically impact the public health, economic security and/or national and homeland security of the United States.

1.18.1 Graphite Supply

Benchmark Mineral Intelligence ("BMI") estimates that worldwide production of natural flake graphite in 2021 will be marginally over 1.0 Mt of which China will produce approximately 75%. Brazil (8%) and Mozambique (6%) make up most of the balance, with the rest largely scattered among Madagascar, Tanzania and Europe.

Chinese production of the larger flake sizes is declining. Heilongjiang has substantial small flake resources and excess production capacity, most of which is targeting the LiB market. Current sources indicate that Chinese production may be under-estimated by leading analysts.

A large new graphite mine was built in Mozambique in 2018 and it substantially increased world natural flake graphite supply. Most of its production has been sold into the Chinese steel market as the Chinese do not find its production suitable for the battery market.

Because of potential demand from the rapidly growing electric vehicle ("EV")/ LiB markets, there are over 20 advanced stage graphite projects outside of China that have completed feasibility studies. These projects are expected to take at least two years to bring into production after they obtain financing. Not all these projects have battery grade material and/or carry substantial country risk.

1.18.2 Graphite Demand

Graphite is the anode material in lithium-ion batteries and is their largest single component. There are no substitutes in this application although the anode material can be either synthetic or natural graphite. Natural graphite is less expensive and has a higher capacity whereas synthetic graphite charges faster and has a longer cycle life and in general a much lower CO² footprint. In a relatively short period of time, LiB's have grown from a small market to the second largest market by volume for flake graphite.

Hybrid vehicles, EVs and grid storage are potentially large markets for continued growth in LiB production and indirectly flake graphite. BMI estimates that flake graphite production may need to more than double by 2025 to meet this demand.

There is very little recycling of, or substitution for graphite.

1.18.3 Graphite Price

There is no posted spot price or futures market for graphite. Sales are negotiated between producers and consumers. Prices increase with flake size and carbon content. The premiums are relatively small going from +150 mesh (small) flake to +100 mesh (medium) and +80 mesh large flake. Prices are much higher for +50 mesh (XL) and +32 mesh (XXL) flake sizes as they are rarer and more valuable. Purity is also a factor, but not considered

important when concentrate grade is above 94% carbon. Most mine concentrates, produced by flotation alone, range between 94 and 98% carbon.

Graphite prices peaked in the \$1,300/t range for the premium grade (large flake +80 mesh, 94-97%C) in the late 1980s and then declined sharply as Chinese producers entered the market. This caused western mines to close and development projects to be shelved. Prices did not begin to recover until 2005 and peaked in a range of US\$2,500 to \$3,000/t in 2012 when some shortages were reported. This was due to the growth in China and the commodity super cycle and the resultant demand from the steel market. The subsequent slowdown in the Chinese economy combined with a lack of growth in the US/Japan/Europe caused prices to fall back approximately 50% from 2012 levels. In the second half of 2017 prices rebounded 30% to 40% due to a recovering steel industry, continued strong growth in LiB demand and environmentally related capacity shutdowns in China. Current CIF Europe prices are approximately US\$1,775/t for 94-95%C +50 mesh flake, US\$1,500/t for 94%C +80 mesh flake, US\$1,415/t for 94%C +100 mesh medium flake and US\$1,135/t for -100 mesh small flake. These prices consist of Benchmark's FOB China prices with approximately US\$300/t added for freight and handling. Due to the current state of container markets, these prices are likely understated.

Northern's sales from the second and third quarters of 2022 (See Table 19-1) from their Lac des Isles plant achieved ex-gate pricing of \$1,643 and \$1,572 respectively (Corporation, MANAGEMENT'S DISCUSSION AND ANALYSIS For the Three and Nine Month Periods Ended September 30, 2022 , 2022), (Corporation, MANAGEMENT'S DISCUSSION AND ANALYSIS For the Three and Six Month Periods Ended June 30, 2022, 2022). The concentrate produced from the Okanjande project will have a superior flake size distribution than that of the Lac Des Isles process and it would be reasonable to expect similar, if not higher, realised prices for this concentrate. Accordingly, a moderately conservative average price of US\$1,550/t has been used for the purposes of this PEA.

1.18.4 Potential Products and Markets

Graphite is sold based on 80 percent meeting the required size specification. A large flake concentrate for example is only 80 percent large flake with the balance being smaller sizes. Accordingly, the size fractions produced by metallurgical testing have been grossed up by 20 percent. The expected product split figures provided by Northern are 11% +50 mesh XL flake; 48% +80 mesh large flake; 24% +100 mesh medium flake; and 17% -100 mesh small flake.

1.18.5 Graphite Market Outlook

In the near- to medium-term, graphite prices are expected to increase as LiB demand is growing rapidly, supply from China is getting tighter, and political, environmental and social governance ("ESG") concerns are increasing with respect to all operations.

1.18.6 Material Contracts

The Company has not yet entered into any material contracts. However, the Company has opened a sales office in Italy to initiate market development in Europe and is also planning on developing a battery anodes material plant in Canada. Any excess product from Namibia would be delivered to this plant for further processing and "value add".

1.19 ENVIRONMENTAL STUDIES, PERMITTING AND SOCIAL OR COMMUNITY IMPACT

1.19.1 Project Environmental Permitting Requirements

A review of available documentation was undertaken to determine the authorisation status Okanjande sites. The status of environmental permitting is listed in Table 1-21.

Table 1-21: Environmental Authorisation Status Okanjande

Year	Consultant	Type of application	Status	Comments
2014	Enviro Dynamics	EIA		
2020	Imerys	Renewal application (with EAR)	EAR Submitted, ECC renewed and Valid until 2024-11-11	
2021	Geokey	Environmental Audit Report	Bi-Annual reporting up to date. EMP up to date	Requirement for Bi-annual internal audit and report

The Okanjande ECC for graphite mining activities was first issued on 23 April 2015.

1.19.2 Regulatory Requirements

1.19.2.1 Environmental clearance certificate (ECC)

Although Okanjande currently has an ECC, it is recommended that any potential changes to the process be reconciled with the current ECC to determine whether amendments are required.

1.19.2.2 Water Abstraction Permits

1.19.2.2.1 Okanjande

The water abstraction permit nr 11163 for the remaining portion of Farm Okanjande was issued 5 July 2021 for the abstraction of 25,000m³/a for the use in mining purposes. The water abstraction permit nr 10946 from the Farm Doornlaagte was issued 13 July 2021 for the abstraction of 90,000m³/a for the use in mining purposes. A new application for abstraction from farm Doornlaagte was submitted 10 February 2023.

1.19.3 Social/Community Requirements

1.19.3.1 Public Participation

Community consultation and stakeholder engagement was undertaken as parts of an ongoing EA process. Detailed in the Community Consultation (2013) and EIA reports (2014, 2022). In March 2023, the company Environam Consultants Trading (ECT) as part of an amendment study, submitted the latest draft of the revised Environmental Management Plan (EMP) for review in the public domain. The draft was also distributed to the Interested and Affected Parties as stakeholders in the process.

A stakeholder database was developed as part of the 2020 amendment (Hook, 2020) and includes all interested and affected parties, Government Ministries, local and regional government councillors and key officers, Government Parastatals, affected landowners, etc. It is updated on a regular basis. The latest draft EIA and EMP (ECT, 2023) was made available to these interested and affected parties on 2 February 2023 and was available for comment until 16 March 2023.

1.19.3.2 Analysis of Social Issues

No social issues of concern have been identified at the time of writing of this report.

1.19.4 Mine Closure

Mine closure plans need to be developed for the Okanjande Graphite project for the closure of Open Pits, Waste Dumps, Tailings Storage Facility, Process Plant Buildings and Infrastructure, as well as general surface rehabilitation.

The open pit mining at Okanjande lends itself to a number of risks which need to be considered during the development of closure plans. These risks were identified in 2014 Environmental Impact Assessment conducted by Namib Hydrosearch.

1.20 CAPITAL AND OPERATING COST ESTIMATES

The Okanjande overall capital and operating cost estimate was developed by Creo Engineering Solutions Pty Ltd and MSA Group, with input from Northern Graphite Corporation.

The cost estimates have been prepared in accordance with the recommended practices of the American Association of Cost Engineers (AACE) for which the minimum accuracy for this level of study is considered to have an accuracy of +40 -30% overall.

The currency for the cost estimate is expressed in 2nd Quarter 2023 US dollars. The base date of the estimate is 30 June 2023. No provision is included for potential future cost escalation. The exchange rate applied is NAD 18.50 : USD 1.00

The scope of facilities addressed in the cost estimate includes the following major elements:

- Mining (OPEX Only)
- Comminution (CAPEX and OPEX)
- Flotation (CAPEX and OPEX)
- Dewatering and Bagging (CAPEX and OPEX)
- Tailings Dewatering and Storage (CAPEX and OPEX)

1.20.1 Capital Cost Estimate

The initial project capital expenditure for the Okanjande Process Plant and ancillaries is given in Table 1-22. The majority portions of the initial capital are required for the construction of the new Grinding circuit, new Tailing Dewatering circuit and Integrated Waste Dump, Sitewide Services and Infrastructure for the new site and the relocation of the existing plant equipment from the Okorusu site to the new Okanjande site.

The Okanjande mine infrastructure has been established and no further capital is required. Contract mining will be applied and as such no capital is required from Northern Graphite for a mining fleet. In the current study no allowance has been made for mining capital cost.

Table 1-22: Okanjande Initial Plant CAPEX Summary

Year 0 CAPEX		US\$	% of DFC	% of TIC
Direct Field Costs	Site Wide Services and Infrastructure	5,694,823	22.8%	16.4%
	Mining and Stockpiles	0	0.0%	0.0%
	Crushing	0	0.0%	0.0%
	Milling	7,996,303	32.0%	23.1%
	Flotation	3,104,756	12.4%	9.0%
	Concentrate Dewatering	1,100,189	4.4%	3.2%
	Screening & Bagging	287,080	1.1%	0.8%
	Tailings Disposal	5,874,712	23.5%	17.0%
	Reagent Handling	8,108	0.0%	0.0%
	Utilities	335,135	1.3%	1.0%
	General Facilities & Buildings	574,045	2.3%	1.7%
TOTAL : DIRECT FIELD COST		24,975,151	100.0%	72.1%
Indirect Field Costs	Contingency	4,995,030	20.0%	14.4%
	EPCM	2,758,723	11.0%	8.0%
	Indirect Construction Costs	1,467,417	5.9%	4.2%
	Owners Team Cost	424,578	1.7%	1.2%
TOTAL INSTALLED COST		34,620,899	122.3%	100.0%

Over the life of mine a further US\$9,229,279 will be spent on sustaining and expansion capital, the breakdown of which is given in Table 1-23 below.

Table 1-23: Okanjande Life of Mine Capital costs

Capital Class	Description	Cost
Initial Capital	Process Plant, 2 Years of IWD	34,620,899
Expansion Capital	NamPower Grid Connection	4,608,447
Sustaining Capital	1% of Initial Direct Field Costs Annually	1,404,501
Close-down costs	IWD Reclamation	198,443
Working Capital	3 Months of year 1 OPEX	3,017,888
Total LOM Capital		43,850,178

1.20.1.1 Basis of Capital Cost Estimate

The estimate is developed based on a blend of material take-offs and factored quantities and costs, semi-detailed unit costs and firm quotes for certain work packages and major equipment supply.

The structure of the estimate is a build-up of the direct and indirect cost of the current quantities; this includes the installation/modification/rectification/construction costs, contractor construction distributable costs, bulk

and miscellaneous material and equipment costs, any subcontractor costs and freight. Source data used for developing the estimate included:

- Process design basis and criteria
- Process flow diagrams
- Drawings and Sketches
- Major Equipment list for new equipment
- Minor equipment list from the existing operations requiring refurbishments, modifications or removal.
- Budgetary quotes for major equipment
- In-house historical data from similar recent in country projects
- Multiple site visits and equipment inspections on existing equipment together with historical information reviews and reports.

1.20.1.1.1 Direct Field Costs

The estimating approach adopted was to identify and quantify the major mechanical equipment as well as most of the minor equipment, including the activities required for the relocation process. The supply and installation of these identified items were then priced through an enquiry process to selected established equipment / technology suppliers, to obtain a combination of budget and firm pricing. Supply and/or relocation costs for mechanical equipment that were not obtained through an enquiry process are based on developed preliminary material take-offs together with recent local contractor rates.

1.20.1.1.2 Indirect Field Costs

Indirect Construction Costs

The indirect construction related costs with respect to the Civils and Earthworks (Factorised at 15% of the discipline specific direct field costs) has been estimated as a percentage of the direct field costs. The related SMPPEI costs were estimated at 3% of direct field costs.

EPCM Costs

These costs cover the project management, engineering, and procurement, and construction management (EPCM) costs directly associated with the implementation of the relocated activities including the planned new additions to the process. The estimate provision ranged between 8% to 14% for the different functional areas. The resultant provision for EPCM services amounted to 11% of direct field costs.

Owners Team Costs

This estimate includes a limited amount (1.7% of total project direct field costs) for provision of owner's cost at the request of the client.

Contingency

Contingency is a sum of money included in an estimate to allow for uncertainty related to estimating "accuracy" and the risk of undertaking the project. A contingency amount of 20% was applied across the board. Whilst relatively low for a PEA level study, the lower contingency rate is supported by a significant fraction of equipment costs already sunk.

1.20.1.2 Mining Capital Cost Estimate

The Okanjande mine infrastructure has been established and no further capital is required. Contract mining will be applied and as such no capital is required from Northern Graphite for a mining fleet. In the current study no allowance has been made for mining capital cost.

1.20.1.3 Process Plant Capital Cost Estimate

The following Direct Field Costs were considered:

- **Site Wide Services and Infrastructure:** The Okoruso processing plant will be relocated to the Okanjande process site. Site wide services and infrastructure estimates are based on the new site requirements and project site layout.
- **Crushing and screening:** Primary crushing at Okanjande will be contracted together with mining and as such no capital cost allowance was made for crushing in this study.
- **Milling:** The costs associated with the milling circuit major mechanical equipment were obtained through formal enquiry processes where several suppliers were engaged for the supply of the required equipment. The site installation associated costs were obtained through completing the basic engineering, developing the supported bill of quantities and going out to market for tenders pricing.
- **Flotation:** The largest contributor to flotation circuit mechanical costs are due to a new polishing scrubber #1 and 2 new polishing vertical stirred media mills, for which firm costs were obtained. The associated installation costs were obtained through an enquiry process going out to market for pricing. The relocation costs (Civil and Structural Steel) were obtained through a budget estimate enquiry process for the relocation of the planned plant infrastructure.
- **Concentrate Dewatering:** The associated mechanical, electrical and instrumentation costs were included under the flotation section estimate. The relocation costs (Civil and Structural Steel) were obtained through a budget estimate enquiry process for the relocation of the planned plant infrastructure.
- **Screening and Bagging:** The associated civil, mechanical and structural steel relocation costs were included under the concentrate dewatering and flotation section estimate respectively. The heat shrinker was obtained from a formal quote from a preferred supplier.
- **Tailings Dewatering and Disposal:** Firm quotes for a turn-key plate and frame filter press plant was used as cost basis for tailings dewatering. A basic design for the IWD facility was completed and the resultant bill of quantity, coupled with local construction rates, was used as the basis of estimate.
- **Reagent Handling:** The mechanical relocation estimate was estimated based on developed high level material take-offs and applying recent contractor rates by in-country contractors. The associated structural steel and E&I costs were included under the flotation discipline estimate.
- **Utilities:** Provision was made for relocation of equipment based on a budgetary enquiry process, with the addition of fire protection and potable water systems.
- **General Facilities and Buildings:** Provision has been included in the estimate to construct new facilities and buildings as deemed necessary. The estimate was based on preliminary designs, developed material off takes and using recent local contractor rates.

1.20.2 Operating Cost Estimate

Total operating cost estimates for the Okanjande Project are presented in Table 1-24. The unit operating costs are based on total mined material of 10.1Mt, of which 6.1Mt is mineralized material and 4.0Mt is waste. The estimated mine life is ten years plus one year of pre-mining/construction. Processing will continue into Year 11. G&A costs, as per Table 1-24, include the following elements: Security Services, External Services, standard site related General and Administration costs and Royalties.

Table 1-24: Summarized Operating Costs

Operating Costs (C1)	(US\$ 10 ⁶)	\$/t Ore Processed	\$/t Concentrate Produced
Mining	60.80	9.97	195.47
Processing	131.97	21.64	424.28
G&A	19.47	3.19	62.59
Total	212.23	34.81	682.34

1.20.2.1 Exclusions

The following cost contributors were not accounted for in the OPEX estimate:

- Scope changes
- Escalation beyond the second quarter of 2023
- Financial cost
- Schedule delays
- Operating permits
- Currency fluctuations
- Reagent and Fuel Pricing fluctuations

1.20.2.2 Basis of Operating Costs

Mining costs are based on engaging a contract miner and budgetary proposals received in June 2022.

Processing labour costs were developed from the anticipated staffing plan, and Namibian wages rates. Power cost estimate was based on recent costs for IPP provided hybrid solar and diesel generation, the current local utility rate of US\$0.11/kWh, and the power consumption estimates from the electrical load list. Variable costs for reagents and consumables were applied based on historical and test work consumption rates and current local pricing for reagents. Fixed cost elements were derived from factoring and from previous operating budgets for the process.

1.21 ECONOMIC ANALYSIS

This PEA is preliminary in nature. The current basis of Project information is not sufficient to convert the in-situ Mineral Resources to Mineral Reserves, and Mineral Resources that are not Mineral Reserves do not have demonstrated economic viability. The PEA results provide an initial assessment of the Project economics based on preliminary information. This PEA is based on technical and economic assumptions which will be evaluated in more advanced studies.

1.21.1 Principle Assumptions

1.21.1.1 Resource

Table 1-25: 10-Year Mine plan for 4.0% TGC cut-off

Mineral Resource Assumptions	Value	UOM	Comments
Weathered			
Measured and Indicated Tonnes	2,540,623	Mt	MSA, Mining Report
Measured and Indicated Graphite Grade	5.03%	% TGC	MSA, Mining Report
Contained Graphite	127,881	t	MSA, Mining Report
Fresh			
Measured and Indicated Tonnes	3,556,327	Mt	MSA, Mining Report
Measured and Indicated Graphite Grade	5.44%	% TGC	MSA, Mining Report
Contained Graphite	193,293	Kt	MSA, Mining Report
Total			
Measured and Indicated Tonnes	6,096,950	Mt	MSA, Mining Report
Measured and Indicated Graphite Grade	5.27%	% TGC	MSA, Mining Report
Contained Graphite	321,174	Kt	MSA, Mining Report

1.21.1.2 Mining

Table 1-26: Mining Assumptions for Economic Evaluation

Mining Production	Value	UOM	Comments
Mineralized material Mining Rate- Mean	609,695	t/a	CES, MSA Mine Plan
Mineralized material Mining Rate - Steady State	631,450	t/a	CES, MSA Mine Plan
Waste Mining - Mean	404,504	t/a	CES, MSA Mine Plan
Strip Ratio	0.7	ratio	CES, MSA Mine Plan
Life-of-Mine	10	years	MSA, Mining Report

1.21.1.3 Processing

Table 1-27: Processing Assumptions for economic evaluation

Process Production	Value	UOM	Comments
Crushing			
Annual Throughput	631,450	t/a	Northern Graphite
Standard Operating Days	365	d/a	Northern Graphite
Standard Operating Hours	12	h/d	Northern Graphite
Availability	92.0%	%	Northern Graphite
Utilization	90.0%	%	Northern Graphite
Effective Utilization	82.8%	%	CES, Calculated
Mean Crush Rate - Instantaneous	174	t/h	CES, Calculated
Milling			
Annual Throughput	631,450	t/a	Northern Graphite
Standard Operating Days	365	d/a	Northern Graphite
Standard Operating Hours	24	h/d	Northern Graphite
Availability	92.0%	%	Northern Graphite
Utilization	90.0%	%	Northern Graphite
Effective Utilization	82.8%	%	CES, Calculated
Mean Grinding Rate (dry)	87	t/h	CES, Calculated
Flotation			
Throughput (dry)	87	t/h	Northern Graphite
Grade	5.3%	%	METPRO
Recovery	92.0%	%	METPRO
Concentrate Grade	95.0%	%	METPRO
Masspull	5.1%	%	CREO, Calculated
Concentrate Production (dry)	32,213	t/a	CREO, Calculated
Co-Deposited dry Tailings and Waste Rock			
Tailings deposition moisture	12.6%	% H2O m/m	CREO mass balance
IWD Phase 1 (non-ARD)	10,837,335	t	CREO, 2023

1.21.1.4 Financial, Taxes, Royalties and Levies

Table 1-28: Financial assumptions

Financial Assumptions	Value	UOM	Comments
Discount Rate	8%	%	CREO
Exchange Rate	18.5	NAD:USD	XE, 6-months
Royalties	2.0%	%	KPMG, 2021
Graphite Flake Price	1,550.00	USD/t TCG	Ex-Gate Okanjande, NG 2023
Corporate Income Tax, Namibia	37.5%	%	PWC, Dec 2021
Depreciation Period	3.00	a	DM-Northern Graphite
Annual Escalation Applied	6%	%/a	CREO, NCPIX
Escalation Period Applied	1.00	a	CREO

1.21.2 Key Performance Indicators

Table 1-29: Project Overall KPI's

Project KPI's	UOM	
Measured and Indicated Resource - Weathered	Mt, %TGC	2,54Mt @ 5.03% TGC
Measured and Indicated Resource - Fresh	Mt, %TGC	3,56Mt @ 5.44% TGC
Post Tax NPV @8% CoC	US\$	70,218,979
Post Tax IRR	%	36%
Post Tax and Royalty Discounted Cashflow	US\$	145,674,213
Post Tax Payback Period	a	4
Pre Tax NPV @8% CoC	US\$	120,396,195
Pre Tax IRR	%	47%
Pre Tax and Royalty Discounted Cashflow	US\$	239,077,435
Pre Tax Payback Period	a	3

Table 1-30: Key Project Cost Indices

Cost Indicators	UOM	Cost Index (US\$)
Mining & Haul OPEX/t Concentrate	\$/t con	135.88
Mining OPEX/t Total Mined	\$/t total mined	4.17
Mining OPEX/t Ore Mined	\$/t Ore mined	6.93
Process OPEX/t Concentrate	\$/t con	439.60
Process OPEX/t Graphite Concentrate produced	\$/t concentrate	22.43
Total OPEX/t Concentrate	\$/t con	635.07
Total C1 Cost/t Concentrate	\$/t con	666.07
Total CAPEX/t Concentrate	\$/t con	115.27
Total Expenses/t Concentrate	\$/t con	781.34

1.21.3 Cash Flow

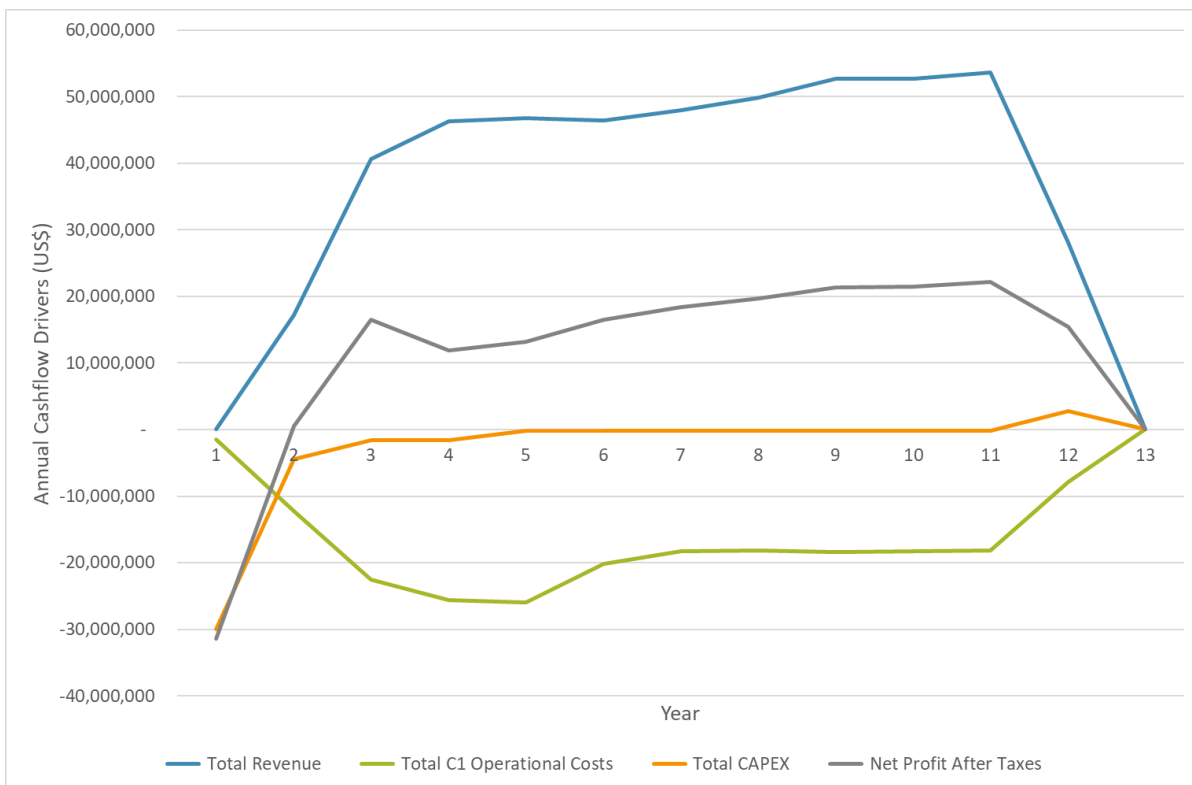


Figure 1-14: Okanjande Project Cash Flow

Table 1-31: Okanjande Project Cash Flow

Year	Mineralized Material Mining	Mineralized Material TCC	Mineralized Material Hauled and Processed	Concentrate Produced and Shipped (dry)	Revenue from Graphite Con Sales	Total OPEX	Total CAPEX	Total Taxable Income ¹	Income Tax	Net Profit After Tax	Cumulative Net Profit After Tax
	t/a	%TGC	t/a	t/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$
0	2023	-	-	-	-	(1,431,093)	(30,012,452)	-	-	(31,443,544)	(31,443,544)
1	2024	205,221	11,451	205,221	11,090	17,189,052	(12,251,842)	(4,388,840)	-	548,370	(30,895,174)
2	2025	520,946	27,080	520,946	26,225	40,648,084	(22,478,005)	(1,623,721)	(9,818,712)	16,546,358	(14,348,816)
3	2026	632,315	30,856	631,450	29,841	46,253,669	(25,548,953)	(1,623,721)	19,151,628	(7,181,860)	11,899,135
4	2027	632,315	31,172	631,450	30,146	46,726,295	(25,990,819)	(140,450)	19,606,179	(7,352,317)	13,242,709
5	2028	631,450	30,927	631,450	29,951	46,423,701	(20,200,907)	(140,450)	25,587,920	(9,595,470)	16,486,874
6	2029	631,450	31,920	631,450	30,909	47,909,230	(18,321,463)	(140,450)	29,447,316	(11,042,744)	18,404,573
7	2030	632,315	33,261	631,450	32,164	49,853,647	(18,210,247)	(140,450)	31,502,950	(11,813,606)	19,689,344
8	2031	632,315	35,124	631,450	33,961	52,639,837	(18,421,158)	(140,450)	34,078,229	(12,779,336)	21,298,893
9	2032	631,450	35,109	631,450	34,000	52,699,559	(18,285,094)	(140,450)	34,274,016	(12,852,756)	21,421,260
10	2033	631,450	35,772	631,450	34,639	53,689,696	(18,159,444)	(140,450)	35,389,802	(13,271,176)	22,118,626
11	2034	315,725	18,501	319,183	18,107	28,065,723	(7,868,722)	2,778,572	20,037,220	(7,513,957)	15,461,615
12	2035	-	-	-	-	-	-	(112,964)	-	-	145,674,213
13	2036	-	-	-	-	-	-	(66,148)	-	-	145,674,213
Total		6,096,950	321,174	6,096,950	311,031	482,098,493	(207,167,745)	(35,853,312)	239,077,435	(93,403,222)	145,674,213

¹ After Depreciation and carried forward losses

1.21.4 Sensitivity Analysis

The sensitivity of the PEA in terms of project NPV was evaluated in terms of the main revenue drivers – Graphite Price and Graphite recovery, as well as the main cost drivers – Power, Mineralized material Mining Cost, Mineralized material Hauling Cost and NAD : USD exchange rate. As per Figure 1-15 below, the project NPV is highly sensitive towards both Graphite Pricing followed Graphite Recovery (based on slope).

Table 1-32: Sensitivities Ranking (absolute slopes)

Parameter	UOM	Absolute Slope	Ranking	Class
Graphite Basket Price	US\$/t	189	1	Market
Graphite Recovery	%	130	2	Operational
Exchange Rate	NS\$:US\$	47	3	Market
Initial CAPEX	US\$	28	4	CAPEX
Power Cost (N\$/kWh)	N\$/kWh	19	5	OPEX
Total Mining Costs Variable	US\$/t	16	6	OPEX
Ore Crushing Costs	US\$/t	8	7	OPEX
NamPower Grid Connection CAPEX	US\$	3	8	CAPEX

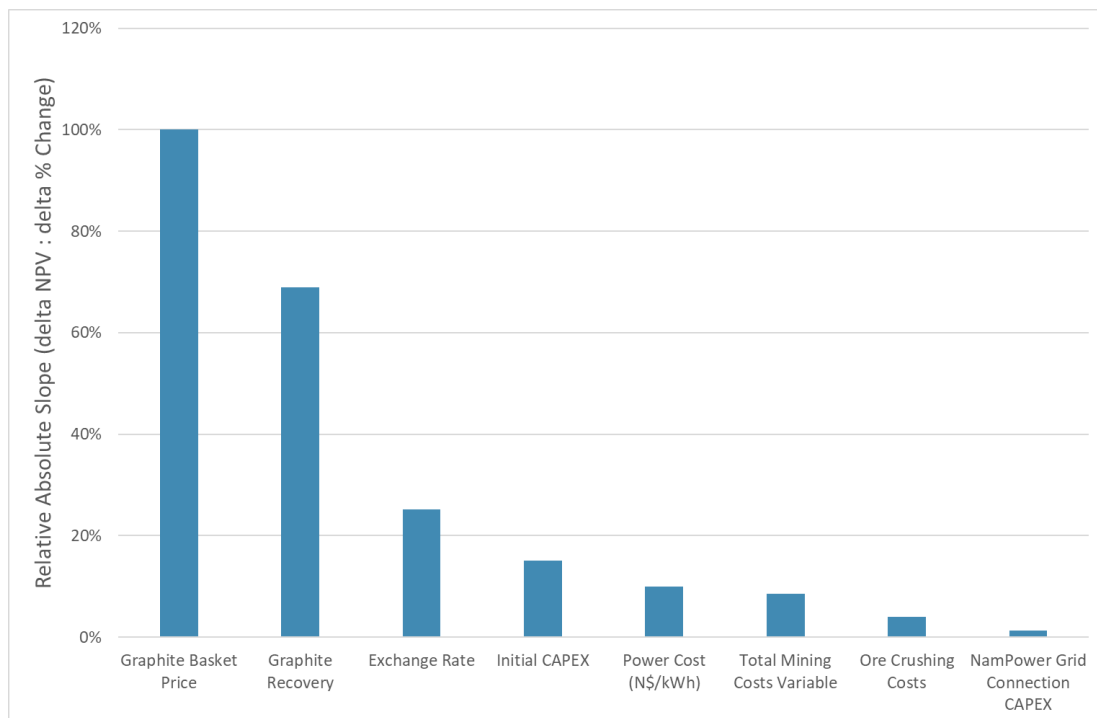


Figure 1-15: Sensitivity Pareto - Relative Absolute Slope

1.21.5 Taxes, Royalties and Other Interests

- **Namibian Corporate Income Tax** for non-Diamond Mining Companies = 37.5%,
- **Value Added Tax (VAT):** Any legal entity with an annual taxable turnover of greater the N\$500,000 is required to register for Value Added Tax (VAT). Value-added tax is payable on the taxable value of all goods sold or imported. The standard rate is 15%. The effect of VAT rebates was not considered in this economic analysis.
- **Withholding Tax Payable and Double Taxation Agreements:** Namibia has Withholding Tax and Double Taxation Agreements with certain countries. However, Canada does not fall in the list of treaty countries and as such this was not considered in this economic evaluation.
- **Depreciation:** The estimated cost recovery for calculation of Namibia income tax consisted of a 3 years 100% declining balance calculation, based on initial and expansion capital. The JV has tax losses of approximately US\$50 million which have not been applied to pre-tax income as the Company's tax advisors have not yet determined if they are available for use.
- **Royalties:** Royalties are levied in terms of the Namibian Prospecting and Mining Act as a percentage of the market value of the minerals extracted by licence holders in the course of finding or mining any mineral or group of minerals. Graphite concentrate will fall into the group of minerals "Semi-precious stones/Industrial metals/Non-Nuclear fuel minerals" for which the Royalty rate is 2%. Royalties were considered on this basis in the economic evaluation.

1.22 ADJACENT PROPERTIES

The Okanjande Mining Licence (ML196) is surrounded by an EPL 4717, which is also held by Imerys-Gecko (35,050 ha). Adjacent to this licence, to the northwest, is an EPL (7123) held by Cobe Investment CC (valid until 08/10/2023) for a range of minerals including industrial minerals (20,966 ha). To the north is an EPL (5847) held by Kunene Resources (Pty) Ltd for precious stones and industrial minerals (6,869 ha). This EPL is valid to 15/07/2022.

To the southeast is an EPL held by Josua Neshuku, EPL 7113 which expired in August 2021. To the east is an EPL 6734 held by Osino Gold Exploration and Mining (Pty) Ltd which expires on 09/06/2023. All adjacent properties are EPLs granted for precious minerals, dimension stone and industrial minerals.

1.23 OTHER RELEVANT DATA AND INFORMATION

None of the contributors are aware of any additional relevant data that might materially impact the interpretation and conclusion of this PEA Report

1.24 INTERPRETATIONS AND CONCLUSIONS

This report was prepared by a group of independent consultants, all qualified persons as defined by NI 43-101, to demonstrate the potential economic viability of open pit mining and processing, based upon the estimated Mineral Resources at the Okanjande Project (MLP). This report provides a summary of the results and findings to the level that would be expected for a Preliminary Economic Assessment. Standard industry practices and assumptions have been applied in this PEA.

Under the base case assumptions for the Project, the PEA indicates an undiscounted pre-tax cash flow of US\$239M, and a post-tax NPV at 8% discount rate of US\$70.2M. The resulting post-tax IRR is 36% for an initial capital investment of US\$34.6M. The Payback period for the Project is 4 years.

The updated project plan produces upside in terms of cashflow when compared to the previous un-discounted cashflow of US\$194.3M. The post-tax NPV of US\$70.2M compares well to the previous post-tax NPV of US\$65.1M. The lower IRR of 36% compared to the previous study IRR of 62% is the result of increase CAPEX related to the relocation of the Okorusu plant and new site infrastructure requirements. The lower IRR of the current study is offset by the benefit of reduced future relocation and site infrastructure costs related to planned expansion of operations.

The results of sensitivity analyses of post-tax cash flow and post-tax IRR show that the project is most sensitive to graphite price and recovery. Breakeven graphite price is US\$976.90/t and post-Tax NPV doubles at US\$2,152.80/t.

The Okanjande graphite deposit has demonstrated its potential for a large tonnage variable grade graphite deposit. The continuity of the deposit is well-defined. The current Mineral Resource is reported within pit shells. The Mineral Resource Estimate demonstrates the potential of the Okanjande Project.

The mineral process and metallurgy for this project has been developed since the late 1980's. While the process flow implemented for the project by Imerys-Gecko encountered difficulties achieving nameplate throughput and concentrate grade, the flowsheet was not based on successful test work conducted previously.

The 10-year mining plan (4%TGC cut-off) considers the mining of weathered as well as fresh graphite bearing material.

The modifications to the Okorusu graphite flotation flow sheet will correct the processing difficulties experienced by Imerys-Gecko and the flow sheet was tested and verified through a test work program completed at SGS in 2021. Tests indicated that the new modified process flow is expected to exceed performance in terms of recovery and concentrate grade, compared to the Imerys-Gecko flow sheet as well as that of the MINTEK flow sheet developed in the 1990's.

The Okanjande site currently has an ECCs in terms of the Environmental Management Act. It is necessary to amend both the ECC to incorporate the proposed changes of the project.

The interim report for geo-chemical testing on Okanjande mineralized material and waste rock indicated (Knight Piesold Consulting, 2022) that transitional and fresh material is acid generating and that weathered material is potentially acid generating. The conceptual lined co-deposited IWD is planned to be constructed on a natural rock formation of marble which should neutralize any weak acid that might form. The tailings and waste storage co-deposition facility represents best industry practice.

The cost estimates were developed to support a nominal mining rate of 631,450t/a. The process facility cost estimate is based on a nominal throughput capacity of 631,450t/a and producing 31,103t/a of upgraded and bagged graphite flakes per year. The cost estimates have been prepared in accordance with the recommended practices of the American Association of Cost Engineers (AACE) and are considered to have an accuracy of +40 - 30%.

Based on the Key Performance Indicators above, the Okanjande Project Makes a strong economic case for success.

1.25 RECOMMENDATIONS

This PEA study is preliminary in nature and is based on technical and economic assumptions which need to be evaluated in more advanced studies. The detail developed is sufficient for a PEA level study. However, additional work would be required to meet a pre-feasibility level.

- **Resources** – In order to fully evaluate the resource potential of the Okanjande deposit and immediate area, un-assayed drill intersections should be assayed, and a drill program carried out to extend and delineate the margins of the current resources and to evaluate nearby exploration targets.
- **Mining** – Optimization of the Mine design and schedule based on current resources.
- **Mineral Processing** – Extend process test work to include locked-cycle testing, variability testing and piloting if possible.
- **Environmental** – Complete advanced geo-chemical studies on tailings and waste rock for definitive classification of all waste rock and mineralized material zones in terms of ARD potential. Complete Mine Closure planning.
- **Capital and Operating Costs** – Refine cost estimation during follow-up studies.
- **Future Development** – The Okanjande deposit currently has fresh rock resources of 1,287k tonnes of contained graphite in the measured and indicated category and 359k tonnes in the inferred category. The Company may wish to undertake a Preliminary Economic Assessment with respect to expansion of the new processing plant at the mine site. This may reduce operational costs due to economies of scale. The currently site layout makes provision for such future expansions.

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2 INTRODUCTION

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

2.1 ISSUER

Northern Graphite Corporation (“Northern” or the “Company”), a Canadian company listed on the Toronto Venture Exchange (“TSX-V”), acquired 100% ownership of Imerys Gecko Holdings (Pty) Ltd which in turn owned 100% interest in Imerys Gecko Graphite (Namibia) (Pty) Ltd and Imerys Gecko Okanjande Mining (Pty) Ltd. The companies were rebranded to Northern Graphite Holdings (Namibia) (Pty) Ltd (“Holdings”), Northern Graphite Okanjande Mining (Pty) Ltd (“NGOM”) and Northern Graphite Processing (Namibia) (Pty) Ltd. (“NGP”). Collectively, Holdings, NGOM and NGP are referred to as “Northern Graphite (NG) Namibia” as they operate as one under joint management and control.

NGOM holds Mining License (“ML”) 196 (the “Property”) which covers the Okanjande graphite deposit and is located approximately 23km by road to the southwest of the town of Otjiwarongo, 230km north of Windhoek, the capital city of the Republic of Namibia (“Namibia”), and 388km east from the port of Walvis Bay. The previously producing Okanjande graphite Mine, active from 2017 to 2018, is located on the Property. The Okanjande graphite mine is currently under care and maintenance, with no mining taking place. NGOM also holds Exclusive Prospecting License (“EPL”) 4717 which surrounds the mining license.

NGP owns all the equipment necessary to process graphite-bearing material and produce a saleable concentrate. The equipment is currently located on land and in buildings currently leased from Okorusu Fluorspar (Pty) Ltd at the idle Okorusu Fluorspar mine site (“Okorusu”), all of which will be relocated to the Okanjande mine site. Okorusu is 60km, by road, northeast of Otjiwarongo and is located on Mining License ML 90, held by Okorusu Fluorspar (Pty) Ltd.

Historically, mining started at the Okanjande graphite Mine in August 2017, with the run of mine (“RoM”) graphite-bearing material trucked to Okorusu for processing and continued until October 2018 when operations at the Okanjande graphite Mine ceased and the Okanjande graphite Mine was placed on care and maintenance.

Imerys SA (“Imerys”) spent more than USD 50 million to retrofit the Okorusu plant to produce graphite concentrate. The process plant experienced several start-up issues and failed to meet design specifications in terms of throughput, recoveries and flake size distribution. The operation was placed on care and maintenance in October 2018.

This Preliminary Economic Assessment Report (PEA) was prepared by Creo Engineering Solutions (Pty) Ltd on behalf of the Company for the Okanjande Graphite Project (the Project). The Okanjande Mine and Processing plant is located approximately 60km southwest from the town of Otjiwarongo in the Otjozondjupa region of Namibia. Okanjande and the Okorusu process plant assets have both been included as part of the acquired JV. The project has been under care and maintenance since 2018 and is planned to be restarted by the client after executing a process plant relocation, several brown fields expansion projects, modifications and refurbishment activities.

2.2 TERMS OF REFERENCE

Northern Graphite commissioned a team of consultants in mid-2021 to conduct a Preliminary Economic Assessment study on the Okanjande Graphite Project. The study was updated in Quarter 1 of 2023 to include the relocation of processing operations to the Okanjande mine site. The Project intends to bring the in-situ (weathered and a portion fresh mineralized material) resource into production. A production rate of 31,103kt/a graphite concentrate of greater than 95% TGC is projected. This PEA focusses on bringing in only 6.1Mt of mineralized material into production from a total resource of 24.2Mt.

The purpose of this PEA Report is to provide an economic analysis of the potential viability of the Okanjande mineral resource. This report was compiled in accordance with the requirements and specifications of the CIM NI 43-101 standard.

This preliminary economic assessment is preliminary in nature, and no mineral reserves have been declared. Mineral resources that are not mineral reserves do not have demonstrated economic viability and thus there is no certainty that the preliminary economic assessment will be realized.

2.3 ABBREVIATIONS

Abbreviation	Description
AI	Bond Abrasion Index
BBWi	Bond Ball Work Index
BESS	Battery Energy Storage System
BRWi	Bond Rod Work Index
BV	Bureau Veritas
CIM	Canadian Institute of Mining
CRM	Certified Reference Material
CWi	Bond Crush Work Index
CZ	Central Zone
DLSZ	Deep Level Southern Zone
ECC	Environmental Clearance Certificate
EMA	Environmental Management Act
EMP	Environmental Management Plan
EPCM	Engineering, Procurement and Construction Management
EPL	Exclusive Prospecting License
HARD	half absolute relative difference
ICP-OES	Inductively Coupled Plasma - Optical Emission Spectrometry
IGG	Imerys Gecko Graphite (Namibia) (Pty) Ltd
IGOM	Imerys Gecko Okanjande Mining (Pty) Ltd
IWD	Integrated Waste Dump
JV	Joint Venture
LECO	Trademark for Laboratory Equipment Corporation
LIB	Lithium-ion Battery
mamsl	Meters above mean sea level
MEFT	Ministry of Environment, Forestry and Tourism
MIBC	Methyl Isobutyl Carbinol
ML	Mining License
NMZ	Northern Margin Zone
NGOM	Northern Graphite Okanjande Mining (Pty) Ltd
NGP	Northern Graphite Processing (Pty) Ltd
NP	Northern Platform
NZ	Northern Zone
OLZ	Okahandja Lineament Zone
PEA	Preliminary Economic Assessment
PFD	Process Flow Diagram

Abbreviation	Description
QA/QC	Quality Assurance / Quality Control
QP	Qualified Person
ROM	Run-of-Mine
RPEEE	Reasonable Prospects for Eventual Economic Extraction
RUL	Rössing Uranium Limited
SANAS	South African National Accreditation System
SEM	Scanning Electron Microscopy
SF	Southern Foreland of the Kalahari Craton
SFA	Size Fraction Analysis
SMC	SAG Milling Comminution
SMM	Stirred Media Mill
SMZ	Southern Margin Zone
SOP	Standard Operating Procedure
SPG	Spherical Graphite
SZ	Southern Zone
TGC	Total Graphitic Carbon
TSX-V	Toronto Venture Exchange
UAV	Un-manned Aerial Vehicle
UCS	Unconfined Compressive Strength
XRD	X-Ray Diffraction

2.4 SOURCES OF INFORMATION

Information contained in this report includes abstracts of professional sources for the different fields of expertise. The following table provides a list of the key sources of information incorporated into this document. See Table 2-1.

Table 2-1: Key sources of information

Source	Title
MSA Group	Technical Report: Northern Graphite Corporation Okanjande Mineral Resource Estimate Namibia, November 2021
MSA Group	Okanjande Project – Mining Report, December 2021
Knight Piésold Consulting	Environmental Scoping Study and Gap Analysis, January 2022
CREO Engineering Solutions	Okanjande Graphite Project: Preliminary Economic Assessment Study, July 2022
METPRO	Process Review and Optimization for the Okorusu Plant, October 2021

It should be noted that some of the historical reports used refer to the mineralization as “ore”. These reports were generated prior to the establishment of the NI-43-101 guidelines and should not be interpreted as being a proven, economically extractable, material.

2.5 QP SITE INSPECTION REPORT

E. Roux – Consulting Process Engineer, CREO Engineering Solutions (Pty) Ltd

Mr Etienne Roux, Director at CREO Engineering Solutions Pty. Ltd., is the Competent Person for the Mineral Processing and Metallurgical testing, Recovery Methods, Capital and Operating Costs and Economic Analysis. I have visited the site on 3 occasions: 19 July 2021, 13 December 2021, 18 January 2022 and 18 February 2022 with relevance pertaining to inspection of existing processing equipment, infrastructure, integrated waste dump, previous mining operations and general lay of land. I have read the instrument, and the parts of the report that I am responsible for have been prepared in compliance with the instrument.

R.N. Barnett – Consulting Associate Industrial Minerals Geologist, MSA Group (Pty) Ltd

Mr Robert Nicholas Barnett, Associate Consulting Geologist with MSA Group (Pty) Ltd, is the Competent Person responsible for the Geological Setting and Mineralization, Deposit Types, Exploration and Drilling, and co-responsible for Sample Preparation, Analyses and Security, and Data Verification. I have visited the site on two occasions from 28 September to 1 October and 18 to 22 October 2021 with relevance to geology, drill core and graphite mineralized material exposed in the mine pit. I have read the instrument and the parts of the report that I am responsible for and have been prepared in compliance with the instrument.

M Mohring – Consulting Associate Mining Engineer, MSA Group (Pty) Ltd

Mr Mark Richard Mohring, Associate Consulting Mining Engineer with MSA Group (Pty) Ltd, is the Competent Person responsible for the Mining Report. I have visited the site from 11 May 2021 to 14 May 2021, where I inspected the pit and associated infrastructure of the mining and processing operation. I have read the instrument and the parts of the report that I am responsible for have been prepared in compliance with the instrument.

3 RELIANCE ON OTHER EXPERTS AND QUALIFICATION FOR DISCLAIMERS

(Prepared by CREO Engineering Solutions Pty. Ltd.)

This Preliminary Economic Assessment Study (PEA) Report has been prepared by CREO on behalf of Northern Graphite Corporation, with inputs from third party. This PEA Report is a compilation of information received from the qualified persons representing each of the stakeholders/consultants. Each qualified person approved and validated their scope of work. Table 3-1 provides a list of stakeholders and consultants who contributed to the study.

Table 3-1: List of stakeholders and assignments

ASSIGNMENT	STAKEHOLDER/CONSULTANT
<ul style="list-style-type: none"> • Mineral Processing and Metallurgical Testing • Recovery Methods • Project Infrastructure • Market Studies and Contracts • Environmental Studies, Permitting and Social or Community Impact • Capital and Operating Costs • Economic Analysis • Compilation of Report 	Creo Engineering Solutions, Etienne Roux & Derick Louw
<ul style="list-style-type: none"> • Property Description and Location • Accessibility, Climate, Local Resources, Infrastructure and Physiography • History • Sample Preparation, Analyses and Security • Data Verification • Geological Setting and Mineralization • Deposit Types • Drilling • Adjacent Properties 	MSA, Robert Barnett
<ul style="list-style-type: none"> • Sample Preparation, Analyses and Security • Data Verification • Mineral Resource Estimates • Adjacent Properties 	MSA, Ipelo Gasela
<ul style="list-style-type: none"> • Mining Methods • Mineral Reserve Estimate 	MSA, Mark Mohring

3.1 CREO RELIANCE ON OTHER EXPERTS - DISCLAIMER

This report was prepared for Northern Graphite Corporation (the Client), by Creo Engineering Solutions (the Consultant), as an independent Consultant, and is based in part upon information and historical study work furnished by the Client and third-party consultants (see Table 2-1).

Information on the graphite market (section 19 Market Studies and Contracts) was compiled from a report commissioned from Benchmark Mineral Intelligence (“BMI”), as well as other industry sources, by Gregory Bowes P.Geo. who is the Chairman of the Board of Northern and not an Independent Person

While it is believed that the information, conclusions, and recommendations will be reliable under the conditions and subject to the limitations set forth herein, the Consultant cannot guarantee their accuracy. The opinions expressed in this report subject to the project responsibilities fulfilled by Creo Engineering Solutions, are Creo Engineering Solutions' best estimate taken from the data available, combined with experience and knowledge of the work. The Client and any other Third Parties that might review or use the information in this report are responsible for their independent analysis based on the information herein and must acknowledge the risks associated with such an operation.

3.2 MSA RELIANCE ON OTHER EXPERTS - DISCLAIMER

Neither MSA nor the authors of this Technical Report, are qualified to verify the legal status of ML 196 and EPL 4717. MSA has relied on a legal opinion by Ellis Shilengudwa Incorporated ("ESI"), dated 09 December 2021, that IGOM was the legal holder of 100% of the interest in ML 196 and EPL 4717, which are valid as of the date of the opinion. ESI is a corporate and commercial law firm based in Windhoek, is regulated by the Law Society of Namibia, and is a member of DLA Piper Africa, a Swiss Verein whose members are comprised of independent law firms in Africa working with DLA Piper. Furthermore, ESI have verified the Existing Lease Agreement in place and the form of the New Lease Agreement in which the agreement(s) provide Imerys-Gecko with the necessary rights, title and interest in and to the Leased Premises to allow for the conducting of agreed processing activities.

The present status of tenements listed in this report (Item 2 – Introduction preface, Item 4.2 – Mineral Tenure, Permitting, Rights and Agreements, and Item 4.3 – Environmental Liabilities) is based on information and copies of documents provided by Northern and ESI, and this Independent Technical Report has been prepared on the assumption that the tenements will prove lawfully accessible for evaluation. Comment on these legal agreements is for introduction only and should not be relied on by the reader. ESI has not provided comment on the status of ML90 which is not the focus of this Technical Report.

Similarly, neither MSA nor the authors of this report are qualified to provide comment on environmental issues associated with the Property.

4 PROPERTY DESCRIPTION AND LOCATION

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

Northern acquired of Imerys Gecko Holdings (Pty) Ltd, which in turn owned 100% interest in Imerys Gecko Graphite (Namibia) (Pty) Ltd and Imerys Gecko Okanjande Mining (Pty) Ltd.. Imerys Gecko Okanjande Mining (Pty) Ltd held ML 196 which covered the Okanjande graphite deposit. The companies were rebranded to Northern Graphite Holdings (Namibia) (Pty) Ltd (“Holdings”) and Northern Graphite Okanjande Mining (Pty) Ltd (“NGOM”).

Historically, mining started at the Okanjande graphite Mine, located on the Property, in August 2017. The run of mine (“RoM”) graphite-bearing material was trucked to Okorusu for processing, and this continued until October 2018 when operations at the Okanjande graphite Mine ceased and the mine was placed on care and maintenance.

4.1 LOCATION

The Property is located approximately 23 road km southwest of the town of Otjiwarongo, 230 km north of Windhoek, the capital city of Namibia, and 388 km from the port of Walvis Bay. The location of ML 196 is indicated in Figure 4-1 and the license area corner co-ordinates are listed in Table 4-1.

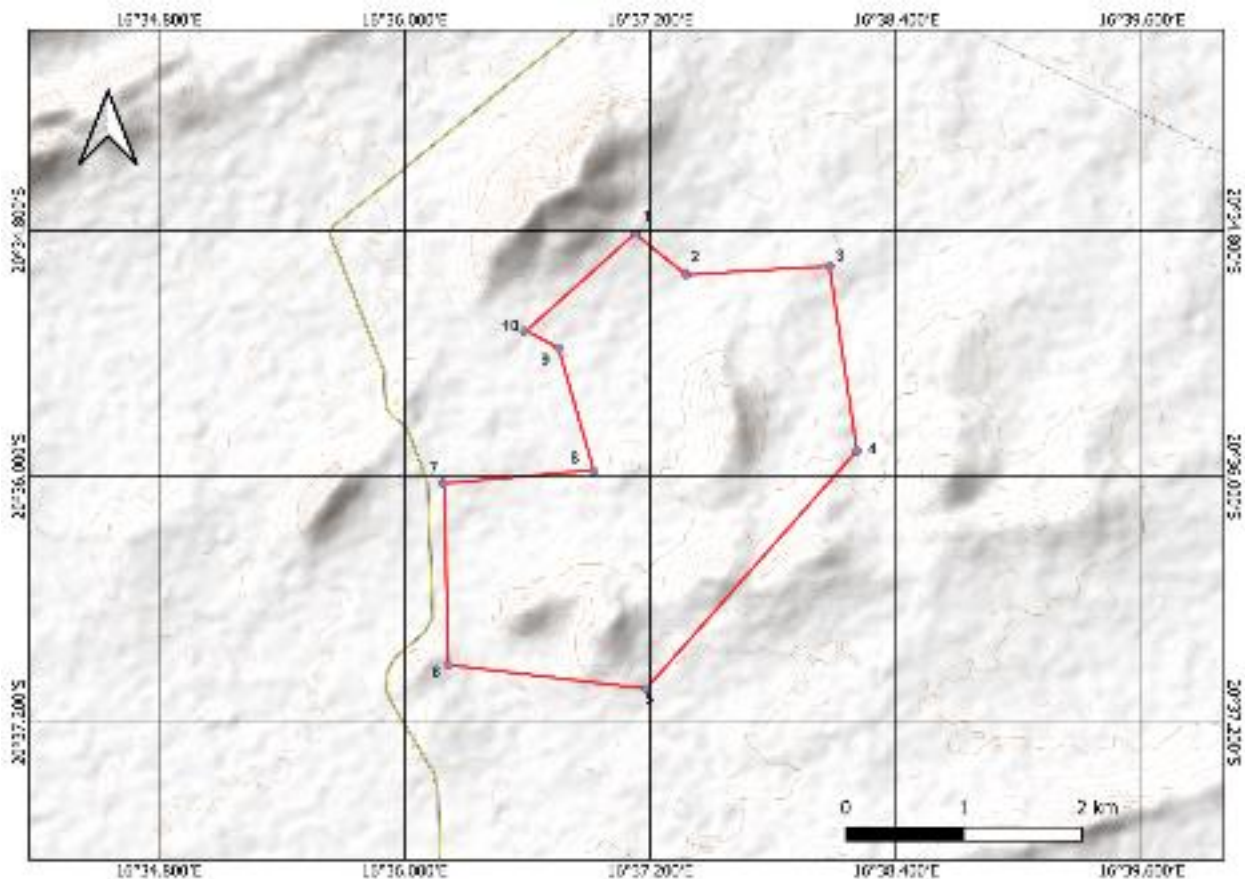


Figure 4-1: Locality map of licence area

Table 4-1 ML 196 corner co-ordinates (MSA, 2021):

Order	Lat Deg	Lat Min	Lat Sec		Long Deg	Long Min	Long Sec	
1	-20	34	48.8	S	16	37	7.6	E
2	-20	35	0.93	S	16	37	22.51	E
3	-20	34	58.41	S	16	38	4.56	E
4	-20	35	52.48	S	16	38	12.43	E
5	-20	37	2.17	S	16	37	10.77	E
6	-20	36	55.08	S	16	36	12.69	E
7	-20	36	1.94	S	16	36	11.32	E
8	-20	35	58.42	S	16	36	55.39	E
9	-20	35	22.27	S	16	36	45.16	E
10	-20	35	17.41	S	16	36	35.09	E

4.2 MINERAL TENURE, PERMITTING, RIGHTS AND AGREEMENTS

The Okanjande Mine is located within ML 196. ML 196 is valid until February 9, 2042 and covers an area of 903.4ha.

NGOM also holds an Exclusive Prospecting License EPL 4717 which surrounds the mining license, covers 46,670ha, and is valid until March 2023. The EPL 4717 Renewal Application was submitted in December 2022.

An Environmental Compliance Certificate (“ECC”) covering mining and exploration activities for Okanjande was applied for, and received from the national authorities on November 11, 2021, valid for three years until November 11, 2024.

4.3 ENVIRONMENTAL LIABILITIES

Historically, Imerys Gecko Okanjande Mining (Pty) Ltd carried out contract mining operations at the Okanjande graphite Mine during parts of 2017 and 2018 and mined approximately 300,000 tonnes of weathered mineralisation (most of which was transported to the Okorusu processing plant) and 400,000 tonnes of weathered waste rock. The waste rock and a stockpile of approximately 14,000 tonnes of graphite-bearing material remains on the Okanjande Property. Geochemical testing by Knight Piesold (Knight Piesold Consulting, 2022) of tailings and waste rock indicated that all weathered tailings and waste rock are potentially acid generating and that all transitional and fresh tailings and waste rock are acid generating. The waste rock has been deposited on a calc-silicate unit which has low permeability and provides natural buffering from any acidic seepage to occur. Mitigating measures are required to manage run-off from the historical waste dump. It is recommended that run-off be routed to the adjacent storm water pond.

5 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

(This section was compiled by The MSA Group Pty. Ltd. with inputs from CREO Engineering Solutions)

Namibia is located in the south-western part of Africa. It is bound by the Atlantic Ocean to the west, and between South Africa and Angola to the south and north respectively. With an estimated population of 2.1 million and an area of 824,292km², Namibia is one of the most sparsely populated countries in the world. The capital city is Windhoek, with approximately 350,000 inhabitants.



Figure 5-1: Namibia Location Map (Maps of the World, 2020)

Namibia is an independent country and a presidential republic with a constitution that follows democratic principles including freedom of speech, press and religion. The country's main industries are agriculture and mining (diamonds, uranium, copper, lead, zinc and marble). The fastest growing sector is the tourism industry due to the country's safety, ecological diversity and climate. The currency of Namibia is the Namibian Dollar (N\$ or NAD) which is pegged to the South African Rand and converts into USD at approximately N\$17.50 : US\$1.00 (8 February 2023).

Namibia's Ease of Doing Business ranking was updated by the World Bank in 2019 to 104, with a ranking of 1 being the most business-friendly regulations and total ranked nations being 190.

5.1 ACCESS AND INFRASTRUCTURE

5.1.1 Regional – Otjiwarongo / Otjozondupa

Otjiwarongo has a well-developed road network. It is situated at the junction of national road B1 that passes north-south through all of Namibia, the C38 to Outjo and further into the Kunene Region in Namibia's northwest, and the C33 to Karibib, which connects the coastal town of Swakopmund and the port city of Walvis Bay. The national road B1 is a major road link between central and northern Namibia and is a corridor for all traffic travelling from South Africa

and Botswana, through Namibia, to Angola. Otjiwarongo is also connected to the national railway grid, run by TransNamib.

Otjiwarongo is connected by road and rail to the port at Walvis Bay, a natural deep-water harbour. Walvis Bay is the largest port on the country's coast and is an important logistical port for the southern African region, providing import and export facilities for Namibia, Zambia, the Democratic Republic of Congo, and Botswana. The port has the capacity to move 1 one million containers a year. Bureaucratic and logistical problems at the city's competitor port at in Durban, South Africa, has resulted in the diversion of traffic to Walvis Bay.

5.1.1.1 Local Resources

Local resources required for labour, supplies and equipment will be insufficient from the nearby community to support the project. Some local resources can be utilised, but skilled mining personnel, consultants and suppliers will be available throughout Namibia to support the project and its operations.

During operations, Okanjande will have to source its key personnel outside of the immediate area, Otjiwarongo, the greater Namibia and possibly from the neighbouring country – South Africa.

The Otjozondjupa region has in the order of 150,000 inhabitants, approximately 30,000 of which live in the main city and capital of Otjiwarongo. Otjiwarongo has a large German-speaking population and Germanic influence is evident in many of the buildings. About 90% of the town's residents speak and understand Afrikaans, about 75% speak English and 35% German. The population density is low, even for Namibia, due to a large portion of the region being communal farming land.

Otjiwarongo's population sustains a diverse economy including cement manufacture, many service industries and institutions. The economics of the town mainly revolve around agriculture as most businesses are related to this sector. Its industries are not large but other key economic activities include the manufacture of polystyrene, tyre re-tread, taxidermy, Fabupharm (manufacturing a wide range of cosmetic and toiletry preparations as well as registered pharmaceutical products for the private sector and the Government Tender Market), brick makers, auction houses, Namaqua Meat (producing meat for the European export market), and charcoal production which is based on several nearby farms. It has several government and private primary and secondary schools and good medical services including doctors, dentists, pharmacists and a government and private hospital.

Some 55,000 people are employed in the Otjozondjupa region, of whom about 31,000 are male. The employment participation rate is approximately 70%; this is above the national average of 55%. The largest occupational group is labourers and other unskilled trades, which constitute 40% of all those employed.

5.2 CLIMATE, OPERATING SEASON

Most of Namibia has a subtropical desert climate characterized by hot summers and mild winters, great differences in day- and night-time temperatures, low rainfall and low humidity. Sunshine days are reported to average 300 days per year. Namibia experiences winter and summer at opposite times from Europe and North America. The winter (June to August) is generally dry. Although arid, the rainy season in the inland north-central areas and in the northeast is from November to March.

Rainfall in the Otjiwarongo area is fairly constant at 400–500mm per year; rain showers occur mainly during November to April, with the highest rainfall rates recorded in January and February. Yearly average mean temperatures are 21°C, with average monthly highs and lows of 29°C and 13°C respectively. However, much higher and lower daily temperatures often occur. The highest temperatures (in degrees Celsius) are experienced in January (an average of

22.7°C with highs of 37.2°C and lows of 12.1°C) and the lowest in June (an average of 15.2°C with highs of 28.5°C and lows of -1.9°C).

The Okanjande Mining Licence can be operated accessed all year.

5.2.1 Precipitation

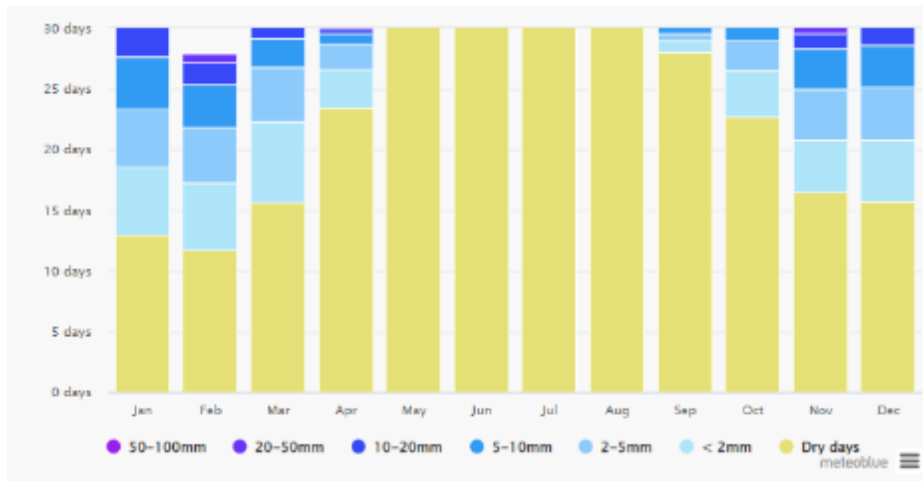


Figure 5-2: Average Rainfall - Otjiwarongo (Meteoblue, 2021)

5.2.2 Temperature

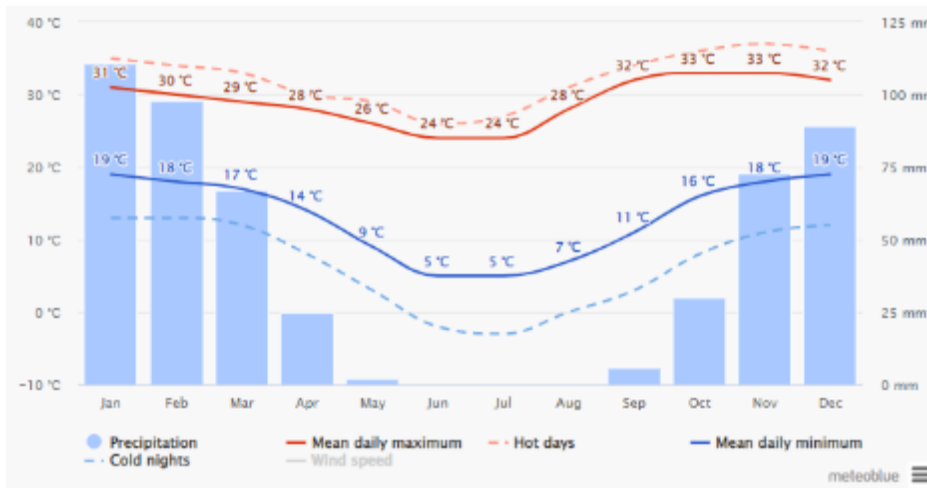


Figure 5-3: Mean Minimum and Maximum Temperatures and Precipitation Occurrence - Otjiwarongo (Meteoblue, 2021)

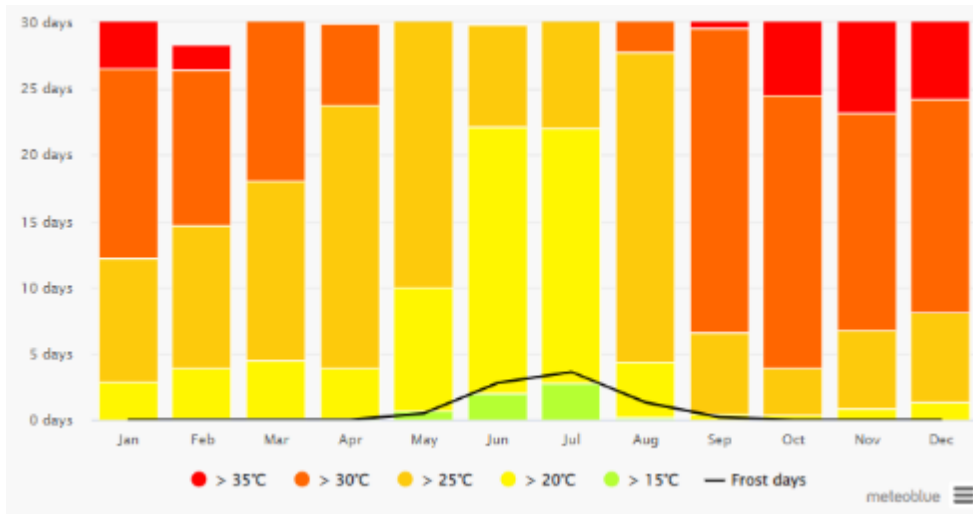


Figure 5-4: Monthly Temperature Distribution and Frost occurrence - Otjiwarongo (Meteoblue, 2021)

5.2.3 Wind

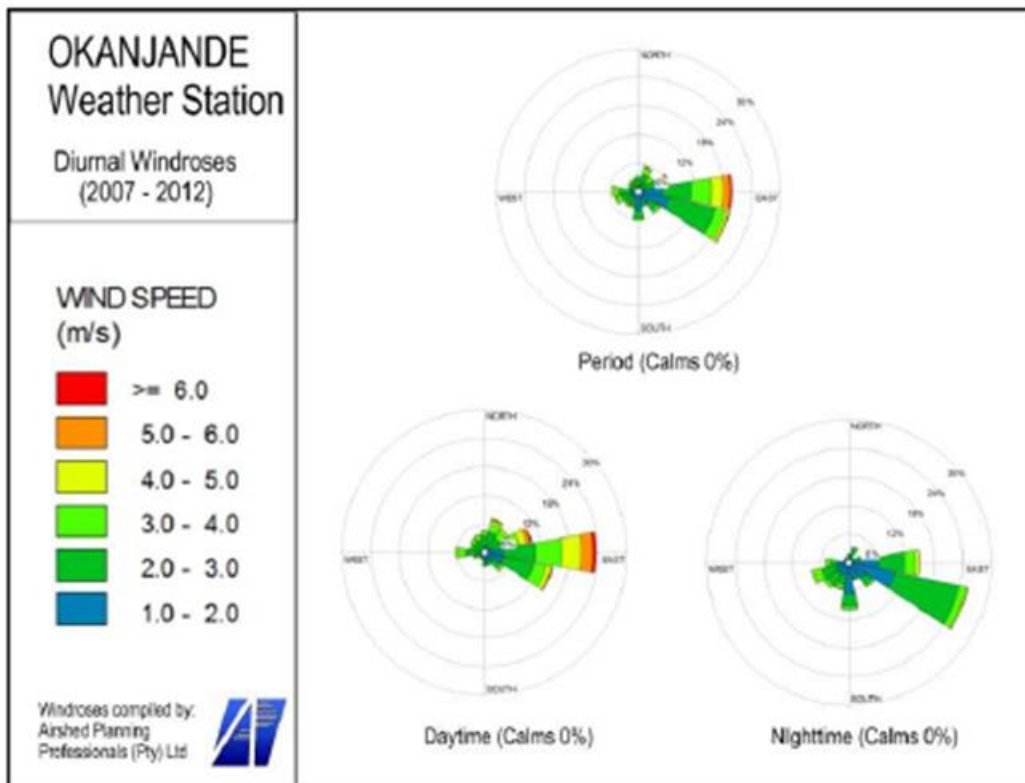


Figure 9-3: Otjiwarongo weather station without northerly component

Figure 5-5: Windrose - Okanjande (Otjiwarongo Weather Station)

5.3 PHYSIOGRAPHY

The Otjozondjupa region is characterized by a relatively flat plain with inselbergs and low ridges standing out above the general land surface. The general elevation is between 1,300m and 1,600m above mean sea level (“mamsl”). The Okanjande graphite deposit is situated on the lower west-facing slopes of one of the ridges. The proximity of other low ridges to the site means that the Okanjande deposit is effectively screened from public view.

The Project area forms part of the thornbush savanna with fairly thin soil and occasional patches of bedrock exposed at the surface. This habitat is widespread and homogeneous throughout central Namibia and contains no unique or singular features of high ecological importance. None of the species found on site are rare, threatened or endangered. The current land use in the area consists primarily of livestock farming and, to a lesser degree, hunting tourism operations. Because of the thin soils and low rainfall, the land has a very low agricultural capability rating. Grazing is only suitable on the lower slopes and valley areas as the soils on the upper parts of the ridges are too thin to support any specific land use.

6 HISTORY

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

In 1990, Rössing Uranium Limited (“RUL”), a subsidiary of Rio Tinto, noted the occurrence of flake graphite on a group of farms immediately south of Otjiwarongo. Subsequent reconnaissance geological programs revealed substantial amounts of flake graphite particularly on the farms Okanjande and Highlands. Between 1991 and 1993, RUL undertook several detailed studies pertaining to the Project including geological, drilling, metallurgical and environmental studies and a feasibility study was completed. The historical exploration by RUL and the related activities are discussed in detail in this report from Items 9 to 11. As part of the RUL Feasibility study, RUL estimated 34Mt of “Demonstrated Geological Reserves” at an average of 5.14% total graphitic carbon (TGC), plus 4.4Mt at an average grade of 6.28% TGC of “Inferred Geological Reserves” for a total of 38.4Mt at an average grade of 5.27% TGC (RUL, 1993; RUL, 1994). Part of the Demonstrated Geological Reserve that was shown to be amenable to mining by open pit methods was reported as a Mineable Reserve with 25Mt at an average grade of 5.65% TGC. Extensive exploration drilling and trial mining operations were carried out. A pilot plant was constructed, and metallurgical test work completed. Due to a decline in graphite prices, development was never pursued.

Gecko Namibia (Pty) Ltd (“Gecko”) subsequently acquired the rights to Okanjande. Its original development plan for the Okanjande deposit included building a processing plant, tailings storage facility (“TSF”), waste dumps and other infrastructure adjacent to the deposit. An EMP was compiled by Enviro Dynamics, approved in 2014 (“the 2014 EMP”) and remains valid. This development scenario was not pursued.

From 2014 to 2015, Gecko completed exploration activities including geophysical surveys and drilling. The detail of this work is described from Items 9 to 11. In 2015, Gecko contracted Kai Batla to complete an “NI 43 101 compliant” Mineral Resource of the Okanjande deposit. The Mineral Resource was completed using the historical RUL and the then current Gecko drillhole data. A total of 112Mt was estimated for the entire Mineral Resource, at an average of 4.32% TGC. In MSA’s understanding, this report was eventually used for internal purposes only.

In 2016 Gecko entered into a Joint Venture (“JV”) with Imerys S.A. (“Imerys”) to form Imerys Gecko Holdings (Pty) Ltd (“Holdings”). Pursuant to the terms of the JV, Imerys earned a 51% interest by financing the development of the Okanjande deposit and retrofitting the Okorusu fluorspar plant to process graphite-bearing material. Gecko retained a 49% interest in the JV.

The development scenario pursued by the JV consisted of mining weathered mineralisation at the Okanjande deposit and transporting it to the Okorusu fluorspar plant for processing. This plant was refurbished and modified to suit the requirements of graphite processing for a substantially lower cost than a green fields development, and it already had an approved TSF as well as access to grid power and water. A Mining Licence (ML196) was awarded for the Project in 2017 pursuant to the 2014 EMP and is valid until 2042. An ECC covering mining, processing and exploration activities for the Okanjande deposit was issued on 18 April 2018, valid for three years. A renewed ECC certificate was received from the national authorities on 11 November 2021, valid for a further three years.

The Okanjande graphite Mine/Okorusu processing operation started up in August 2017, however the processing plant encountered a number of difficulties with throughput, flake size and purity all being lower than expected. The operation was put on care and maintenance in October 2018. The Okorusu plant, as reconfigured by Imerys-Gecko, was based on the Imerys plant at Lac Des Iles in Quebec, Canada, and was unsuitable for the Okanjande deposit mineralization and did not reflect the successful flowsheet developed previously by RUL, through test work and pilot plant programs.

It is likely that there was poor liberation of the larger flakes at the coarse primary grind size of P80 ~425 microns and as a result, a large amount of the large graphite flakes were redirected to a regrind mill. This reduced throughput and resulted in flake size degradation.

One of the polishing mills was also being over-fed, creating a bottleneck and resulting in a reduced plant feed rate.

The Okorusu plant also employed spirals as a means of recovering the larger flakes and these were inefficient.

In 2018, the JV completed a review of the historical data, topographic survey and drilling at Okanjande and surrounding prospects within the Exclusive Prospecting Licence area. The activities related to exploration and drilling are discussed in detail from Items 9 to 11. Further in 2018, on behalf of the JV, Imerys completed an internal Mineral Resource estimate of the Okanjande deposit using historical drilling data collected by RUL and Gecko. None of the Imerys-Gecko drillhole data was used. A total of 60.2 Mt of Weathered and Fresh material was estimated at an average grade of 4.86% TGC (non-Code compliant).

7 GEOLOGICAL SETTING AND MINERALIZATION

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

7.1 REGIONAL GEOLOGY

The Okanjande deposit is located in a metamorphic complex in the north-central (“nCZ”) part of the Neoproterozoic Damara Orogen. The Damara Orogen comprises three highly oblique mobile belts; namely the north-northwest trending Kaoko Belt in the northwest, the Gariep Belt in the southwest and the north-eastern trending Damara (inland) Belt. These three mobile belts mark the convergence of three cratonic blocks, namely the Rio de la Plata, Congo, and Kalahari cratons, during the creation of the Gondwana Supercontinent (Lehmann et al., 2016).

The Damara belt is sub-divided into seven north-east trending tectonostratigraphic zones, named from north to south (Nascimento et al., 2017) (Regional Geology of the Damara Belt (Miller, 2013)):

- Northern Platform of the Congo Craton (“NP”);
- Northern Margin Zone (“NMZ”);
- Northern Zone (“NZ”);
- Central Zone (“CZ”), subdivided into Northern and Southern parts (“nCZ” and “sCZ”, respectively);
- Southern Zone (“SZ”) - also referred to as the Khomas Trough, includes the Okahandja Lineament Zone (“OLZ”) and the Deep-level Southern Zone (“DLSZ”);
- Southern Margin Zone (“SMZ”); and
- Southern Foreland of the Kalahari Craton (“SF”).

The CZ of the Damara Orogen consists of the Damaran metasediments that have undergone greenschist-amphibolite facies metamorphism, through multiple phases of deformation, and are intruded by several syn-/late-tectonic felsic igneous intrusions. The CZ is sub-divided by the northeast-trending Omaruru Lineament-Waterberg Fault into a northern zone, referred to as the nCZ, and a southern zone, known as the sCZ. The Omaruru Lineament distinguishes between the magnetically quiet domain of the nCZ and the disturbed part in the sCZ (Steven, 1993). The nCZ is bound by the Otjihorongo Thrust to the north (Lehmann *et al.*, 2016).

The Project is located in the nCZ and is underlain by Nosib Group sediments and upper Swakop Group shelf carbonates that are intruded by a number of syn- and late-tectonic granites and pegmatites (Steven, 1993).

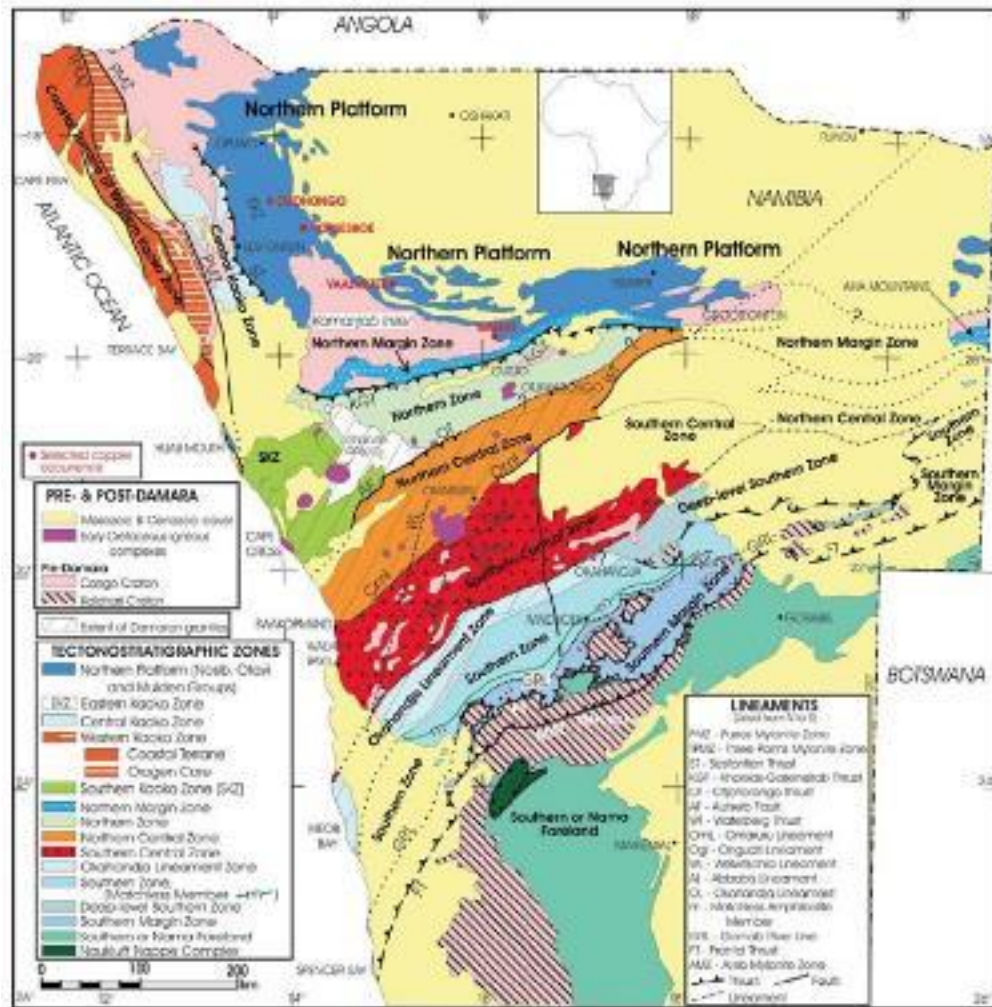


Figure 7-1: Regional Geology of the Damara Belt (Miller, 2013)

7.2 LOCAL GEOLOGY

The Nosib and Swakop Groups in the Project area have been intruded by syn- and post-tectonic pegmatites and granites, with the latter typically comprising two types of leucogranites. These pegmatites and granites occur as irregular and semi-disconcordant narrow bodies, which cross-cut the metasedimentary layering in places. Quartz veining also occurs in the area. The sedimentary rocks have been subjected to high grade regional metamorphism (RUL, 1993; RUL, 1994).

The Nosib Group comprises arkoses, feldspathic quartzites and ortho-quartzites that have been metamorphosed to feldspathic-quartzitic gneisses and graphitic quartzite, which host the graphite as an accessory mineral. The graphite mineralization is hosted locally in discrete units. Sulphide mineralization is prevalent in the Nosib Group. The Nosib Group is relatively resistant to weathering and forms a range of low hills that are observed in the area (RUL, 1993; RUL, 1994). It is not clear whether the host rock is a gneiss or quartzite; this requires further investigation.

The Swakop Group conformably overlies the Nosib Group, with its basal contact marked by a thin biotite schist. The lithologies of the Swakop Group include marble, locally intercalated with calcilicates, overlain by a thin layer of graphite-rich feldspathic quartzite, followed by a thick sequence of quartz-biotite schists and gneisses with thin calcilicate intercalations intruded by granites (RUL, 1993; RUL, 1994).

Cover rocks comprise Quaternary sediments, including alluvium, sand and gravel, which tend to accumulate in the valleys. Calcrete is developed locally, overlying the marbles and calc-silicates of the Swakop Group (RUL, 1993; RUL, 1994).

The area has been subjected to multiple phases of deformation, which have resulted in dome-and-basin structures with a WSW-ENE fold axis. The synformal structures are associated with steep and near-vertical foliations, while the antiformal structures are associated with gentler foliations (RUL, 1993; RUL, 1994).

Major north-south and northeast-southwest orientated faults cut across the region, which are typical of the Darama region (RUL, 1993; RUL, 1994).

7.3 PROPERTY GEOLOGY

In the Project area, the Nosib and Swakop Groups are tightly folded along a west-southwest fold axis, resulting in quartz-rich ellipsoidal domes of the Nosib Group, while the valleys comprise Swakop Group. The graphite mineralized gneisses occur on the flanks of the antiformal dome structures along valley margins (Figure 7-2) (RUL, 1993; RUL, 1994).

Graphite occurs locally as disseminated flakes in stratabound units. Graphite mineralization is ubiquitous in the Property (Figure 7-2), and several targets have been identified and investigated by the different owners over time, with the most successful target being the Okanjande deposit (RUL, 1993; RUL, 1994). Several other targets are currently defined that require further investigation.

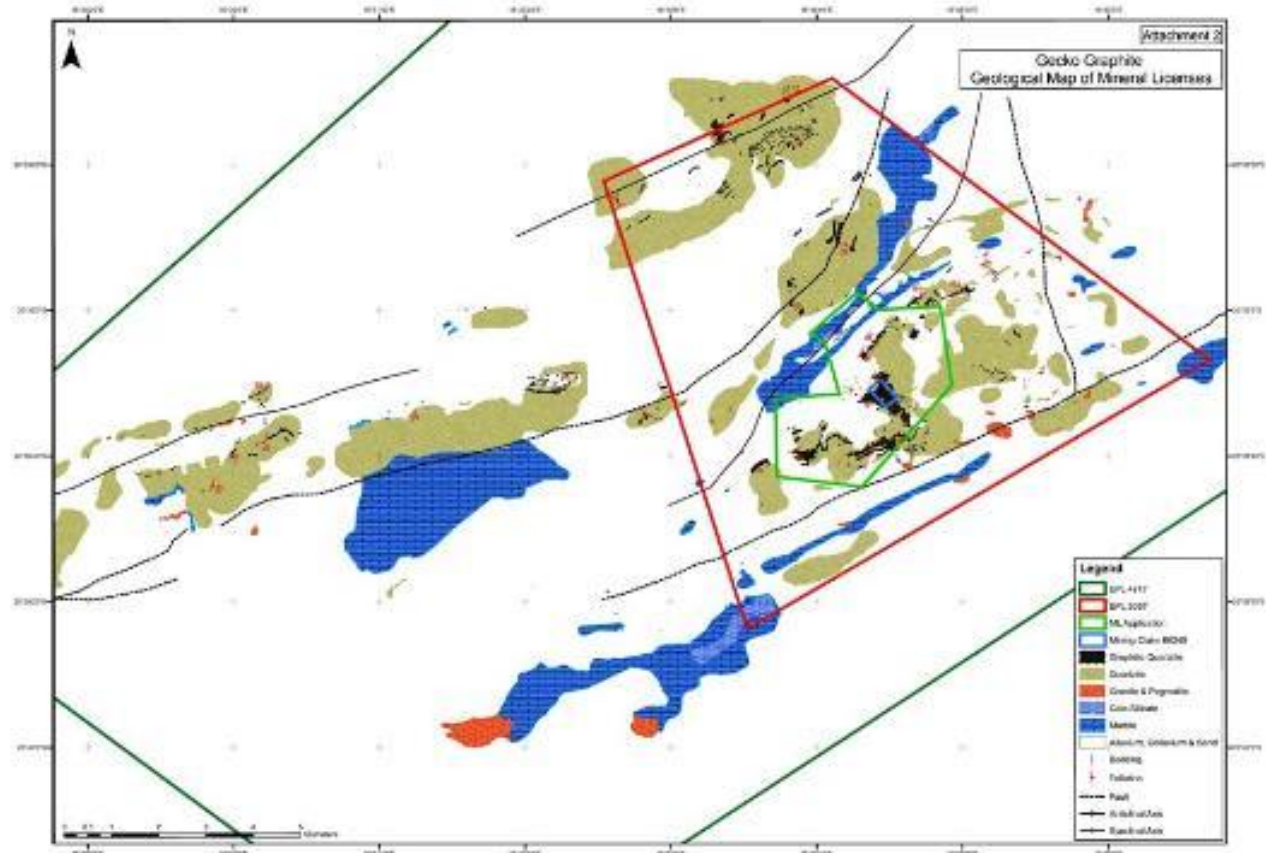


Figure 7-2: Property Geology Map (Gecko, 20xx)

The Okanjande deposit reaches its highest grade and is thickest in the centre, with graphitic carbon grades and mineralization thickness decreasing towards the peripheries. The contacts of the graphite units are difficult to define as they tend to be transitional. The graphite mineralization is typically defined over zones of significant total graphitic

carbon grade rather than discrete lithologies, since the mineralized and non-mineralized rocks can be petrographically and stratigraphically the same. Unmineralized / waste zones are typically graphitic poor gneisses/quartzite, interbedded within the graphitic gneiss/quartzite or irregular intrusive pegmatites and granites. Two other waste rocks identified at the Okanjande deposit include a pinkish coloured, phlogopite-rich quartzite, which is typically 2 cm to 3 cm thick; and a friable mica schist, which rarely exceeds 1 m in thickness. The mica schist may result in pit wall instability if exposed on bench faces (RUL, 1993; RUL, 1994).

The Okanjande deposit is a massive deposit with shallow dipping layering to the south-east. It extends on surface approximately 900m northeast to southwest and up to 820m from northwest to southeast. The mineralization extends from surface and has been defined by drilling up to a depth of 90 m below surface. The mineralization is open at depth. The deposit is extensively weathered to depths of generally between 10m and 20m. The thickness of the weathered zone increases in the southwest and northeast of the deposit. A transitional zone of partial weathering of 1m to 5m thickness overlies the fresh rocks. The thickness of the transitional zone increases to the northeast of the deposit. An altered zone was identified at depth in the western part of the deposit from core logging and petrographic studies. The altered zone is confined to shear zones in this area (RUL, 1993; RUL, 1994).

Towards the deposit edges, the geology appears to be more variable with lower grades, as observed in the northern and southern extensions of the deposit. More work is required in these areas to better understand these complexities.

The structure of the Okanjande deposit is tenuous. The present interpretation of the deposit structure is an isoclinal to recumbent fold structure that was refolded on a NW-SE axis. This is suggested by the small-scale structures identified in the rocks. Steeply dipping E-W and NE-SW trending faults have been observed to cut across the deposit (RUL, 1993; RUL, 1994).

The mineralized gneisses/quartzites have banded, porphyroblastic or magmatic textures. These gneissic textures relate to varying degrees of mineralization (Imerys-Gecko, 2018):

- The “banded mineralized material” has bands of high-grade graphitic quartzite interbedded with barren bands of quartzite;
- The “porphyroblast mineralized material” consists of massive high-grade graphitic quartzite with medium to large sillimanite porphyroblasts. This mineralization style can be of relatively high grade. Graphite flakes in this mineralization style have preferential orientations which result in a schistose fabric; and
- “Migmatitic mineralized material” is related to higher metamorphic grade and has magmatic folds that indicate early melts due to an increase in temperature during metamorphism. Generally, this mineralization style has the highest grades.

7.4 MINERALIZATION

RUL completed mineralogical studies on selected Okanjande deposit core samples at an independent laboratory, Mintek, Johannesburg. The mineralogical studies were completed on weathered and fresh mineralized samples from 12 drillhole cores using transmitted and reflected light microscopy and point counting methods, scanning electron microscopy (“SEM”) and X-ray diffraction (“XRD”).

The mineralogical investigations revealed that the graphitic samples generally contain similar minerals, although they vary significantly in both proportion and texture. The minerals identified are shown in Table 7-1.

Table 7-1: Okanjande Mineralogy

Major Constituents (>10%)	Minor Constituents (between 10% and 1%)	Trace Constituents (<1%)
Quartz	Muscovite	Rutile
K-Feldspar	Plagioclase	Zircon
Jarosite (K-Fe-sulphate)	Graphite	Apatite
	**Phlogopite	Chlorite
	**Biotite	+Chalcopyrite
	Sillimanite	Ce-Phosphate
	+Pyrrhotite	Monazite
	+Pyrite	Ba-K Feldspar
	*Fe-Hydroxides (Goethite)	Baryte
	*Haematite	+Molybdenite
		Wolframite
		Cassiterite
		+Ni-Sulphide
		Cr-spinel
		Xenotime

Source: RUL, 1993

Note: * in weathered mineralized samples

+ in fresh mineralized samples

** present in some rocks only

The major silicate minerals of the graphitic gneiss at Okanjande are quartz and K-feldspar. The most common mica is muscovite, comprising between 2% and 8% of the rock.

Sulphide mineral content in the samples varies between 2% and 7%. Pyrite and pyrrhotite are the most commonly occurring sulphide minerals, with approximately equal proportions. Other sulphide minerals were identified in trace amounts, including chalcopyrite, molybdenite and covellite (only observed in weathered rock). Sulphides occur in irregular patches or clusters of up to 20mm across, or in veins associated with quartz, and a small proportion of the sulphide is associated with graphite flakes.

Five sulphate minerals (barite, gypsum, jarosite, sidorotil and melanterite) were identified. Barite occurs in small quantities and is the only primary sulphate mineral, while the other minerals are as a result of weathering and oxidation of sulphide minerals. Jarosite can make up more than 10% of the weathered rock and some of the jarosite particles are closely associated with graphite flakes. Sidorotil and melanterite are soluble iron sulphate minerals which occur in minor amounts in the weathered rock, commonly along joints and fractures. Sulphate minerals may occur at depth in fresh mineralisation due to oxidation along joints, faults and shears.

Rutile content in the Okanjande deposit samples is between 0.2% and 1.3% with an average of 0.8%. Rutile particle size is between 0.05mm and 0.40mm, with an average of 0.20mm. Rutile grains are commonly enclosed in gangue minerals and most have no association with graphite. A small proportion of rutile has inclusions of graphite, zircon and sulphides. SEM analysis indicates that some rutile grains are high in vanadium occurring in solid solution within the rutile structure.

The graphite content of the analysed samples ranges between 3% and 18% TGC. The average flake size ranges between 500µm and 800µm. More than 85% of the graphite flakes occur as straight undeformed, well-crystallised, euhedral flakes, and free of interlaminated gangue minerals in unaltered mineralized material. Only 3% to 12% of the graphite flakes are interlaminated with gangue minerals. Most of the flakes that are interlaminated with gangue minerals are very large flakes that occur in the form of books. A small proportion of the flakes (<5%) occur as tiny flakes or grains enclosed in the feldspar or muscovite grains.

In the western part of the deposit, approximately 15% of the graphite flakes in weathered rocks at depths below the observed levels of oxidation have altered textures. The altered graphite flakes are associated with localised shear stresses at depth along a zone trending northeast to southwest. Altered graphite flakes are curved, bent or broken, reducing flakes to smaller sizes. Graphite flakes with this texture are interlaminated with gangue minerals and are commonly rimmed with muscovite.

8 DEPOSIT TYPES

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

Graphite mineralization occurs in three types of deposits worldwide; microcrystalline, vein graphite and crystalline flake graphite (Simandl et al., 2015):

- Microcrystalline graphite or amorphous graphite deposits have partially ordered graphite flakes. The deposits are found in low regional metamorphic or along contact metamorphic environments of coal seams (Simandl et al., 2015).
- Graphite vein deposits are made up of graphite veins typically found in metasedimentary belts that are metamorphosed to upper amphibolite and granulite facies. In these belts, vein graphite deposits are found in in pegmatite like bodies (quartz, graphite and pyrite precipitated in layers) in anticline fold hinge zones. Graphite veins are currently only mined in Sri Lanka. Vein graphite product is commonly made up of graphite-rich fragments that are typically 0.5 cm to 0.8 cm in diameter (Simandl et al., 2015).
- Crystalline flake graphite deposits are commonly hosted by marbles and paragneiss that have been subjected to upper amphibolite to granulite facies metamorphism. These deposits can also occur in iron formation, quartzite, pegmatite, syenite and, less commonly in serpentinized ultramafic rocks. Graphite mineralization in paragneiss occurs in thick sequences, and mineralization tends to be evenly distributed with total graphitic carbon grades in the order of 2% to 3% or less (Simandl et al., 2015).

The Okanjande deposit fits the description of a crystalline flake graphite deposit in a paragneiss. The original rocks at the Okanjande deposit were arkoses, feldspathic quartzites and ortho-quartzites that have been subjected to high grade metamorphism to become feldspathic-quartzitic gneisses and graphitic quartzites.

The Okanjande deposit is a massive deposit with shallow dipping layers to the southeast. Its surface extent is approximately 900m northeast to southwest and up to 820m from northwest to southeast. The mineralization extends from surface and has been defined by drilling up to a depth of 90m below surface. The mineralization is open at depth.

9 EXPLORATION

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

Exploration had been carried out by various companies since 1990. In addition to drilling work and analyses of the drillhole samples (documented in Items 10 and 11), the following exploration was undertaken:

- 1990 – 1993: RUL carried out topographic and aerial surveys;
- 2003 – 2010: Solvay – no work undertaken;
- 2010 – 2016: Gecko completed mapping, soil sampling, pitting, geophysical surveys and topographic surveys; and
- 2016 – to date: Imerys-Gecko completed a review of the exploration data and a topographic survey.

9.1 RÖSSING URANIUM LIMITED (RUL) EXPLORATION

According to the RUL reports (RUL, 1993; RUL,1994), topographic surveys and aerial black and white photography were completed at Okanjande deposit subsequent to drilling.

The RUL work was coordinated on a local grid.

9.2 GECKO EXPLORATION

In 2014 and 2015, Gecko completed geological mapping, soil sampling, pitting and geophysical surveys. Soil sampling and pitting were completed as part of the environmental investigations on an area identified for the establishment of a plant. Gecko completed extensive geological mapping of exposed outcrops, and geological maps of the Project area were produced as a result of the mapping programme (Figure 9-1).

Gecko contracted Fugro Airborne Surveys (“Fugro”) to fly a fixed wing TEMPEST electromagnetic and magnetic survey at 200m line spacing over the Project area. After second level processing, a set of conductivity depth images were produced by the contractor, together with the imagery on the terrain and magnetics of the area. Later, Imerys-Gecko planned holes based on these surveys (Figure 1-9).

In 2014, at the time of drilling, Gecko contracted Mr Zaan Wichtman of African Land Survey to complete a topographic survey, as well as collar surveys of the historical drillholes and the drillholes drilled by Gecko.

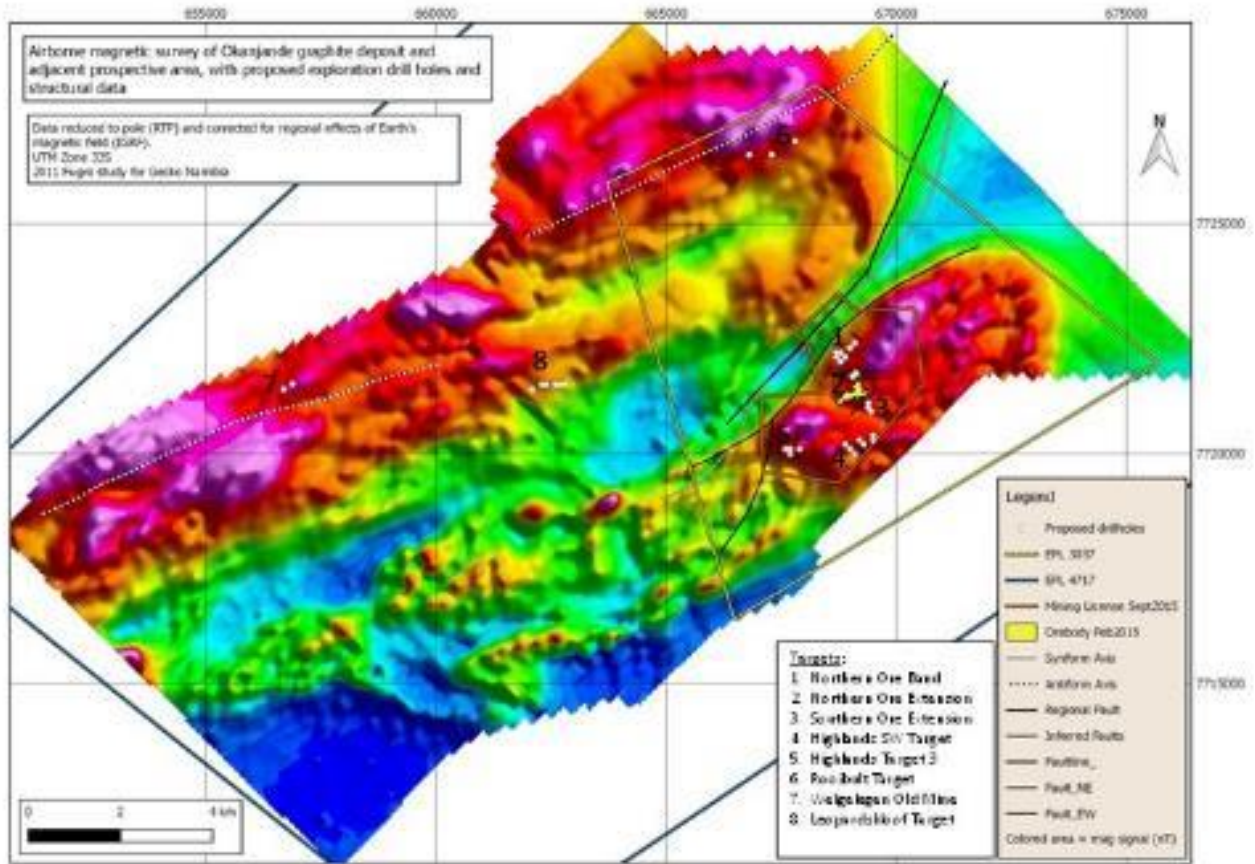


Figure 9-1: Gecko airborne magnetic survey with planned Imerys-Gecko collars

9.3 IMERYS-GECKO EXPLORATION

In September 2018, Imerys-Gecko contracted Mr H. L. Strydom of Strydom & Associates to complete a topographic survey of the Project area, which included a pit and several stockpiles. The survey was completed using unmanned aerial vehicle (“UAV”), commonly known as a drone, with a Leica SR530 GPS system. The survey used several pre-marked, surveyed ground control points and was based on the local UTM33 projection. The method is expected to have an error of less than 0.025 m in the X, Y and Z co-ordinates. This topography was used to limit the vertical extent and deplete the mined-out areas from the Mineral Resource model.

9.4 NORTHERN GRAPHITE EXPLORATION

No exploration work has been conducted by or on behalf of, Northern Graphite Corporation.

10 DRILLING

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

Drilling at the Okanjande deposit was completed by RUL (early 1990s), Gecko (2014 to 2016) and Imerys-Gecko (2018).

10.1 RUL DRILLING

The descriptions of the RUL drilling were derived from the RUL reports (RUL, 1993; RUL, 1994). The work described in this section was completed prior to the QP’s involvement in the Project, therefore the QP did not observe any of the processes described during their implementation.

In 1990, RUL drilled 66 percussion drillholes and 25 diamond drillholes, with 11 of the diamond drillholes twinning the percussion drillholes at the Okanjande deposit (also known as Target 2). Percussion drilling was used to rapidly delineate the Mineral Resource, while being cost-effective. Diamond drilling was used to provide detailed geological information, to obtain samples for metallurgical test work and to drill where the sites were not accessible for the percussion drilling equipment. Poor recoveries, slow drilling rates and mud losses were encountered during the diamond drilling, especially in weathered areas, which limited its effectiveness as the primary drilling method.

The drillholes were either drilled at an approximate azimuth of 320° and a dip of 60° or vertical. The drillholes were completed along southeast to northwest lines at an average line spacing of 50m, with the southwestern part of the deposit drilled at 100m line spacing. The holes were drilled 50m apart on each line (Figure 10-1). Most of the drillholes reached a depth of 60m, with the minimum and maximum depths being 40m and 90m, respectively.

In the area where the bulk sample was taken, RUL drilled 11 short percussion holes of up to 25m in depth.

Assay information is available for a total of 5,978m of RUL drilling at the Okanjande deposit, which is made up of 4,376m drilled by percussion and 1,602m drilled by diamond drilling. Logging information is only available for 296.7m of diamond drillhole core.

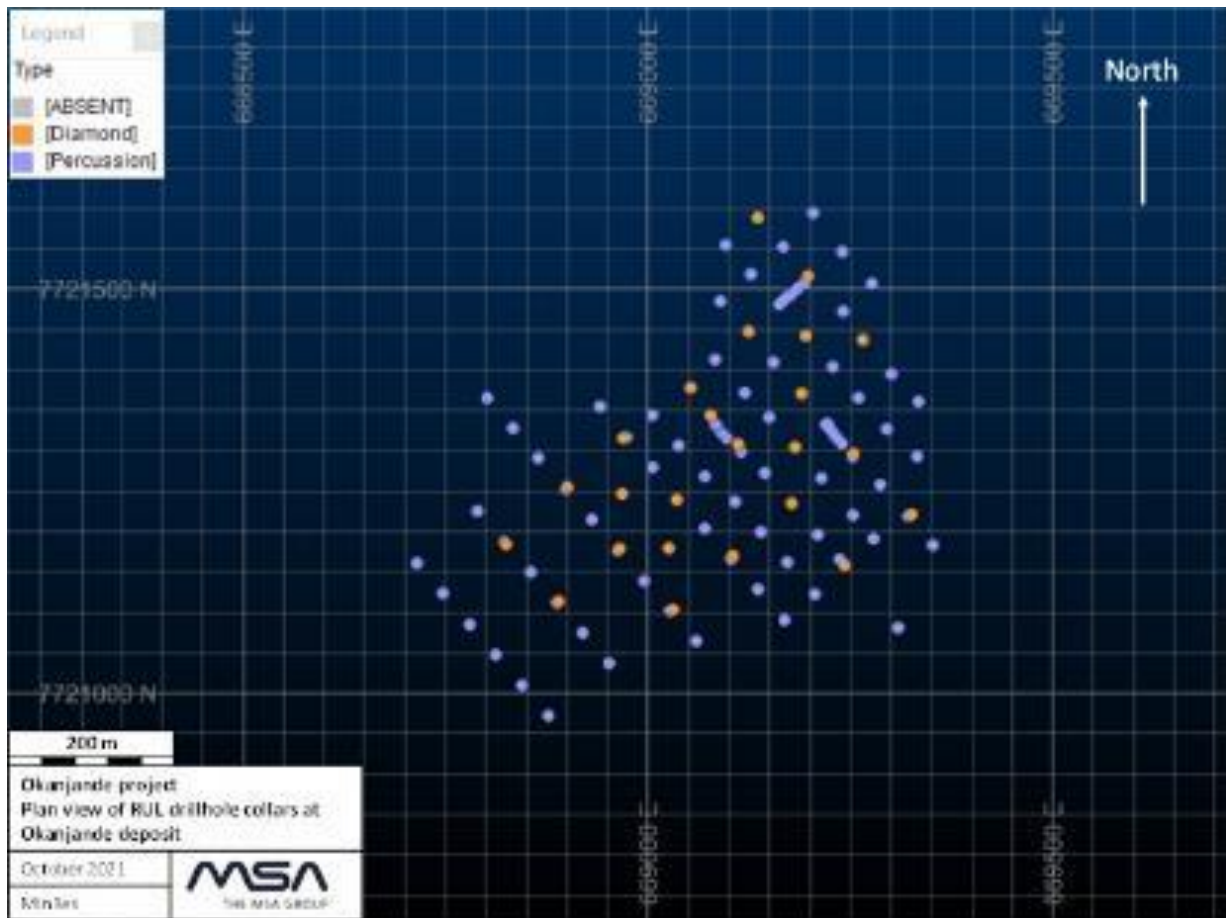


Figure 10-1: Collar positions of RUL drillholes at Okanjande deposit, colour-coded by type (MSA, 2021)

According to RUL reports (RUL, 1993; RUL, 1994), RUL originally identified four drilling targets. Target 2 is the Okanjande deposit discussed in the preceding paragraphs.

- Target 1 is adjacent to the district gravel road of D2515 and had one hole drilled, which did not intersect mineralization;
- At Target 3, to the southwest of the main Okanjande deposit, RUL completed nine percussion drillholes. RUL reported that two drillholes intersected mineralization but the results for the rest of the drillholes are not discussed in the RUL reports, thus MSA assumed that they did not intersect significant mineralization. RUL completed follow up diamond drilling of two holes, however these did not intersect mineralization and the cores were not sampled; and
- Target 4 forms part of a domal structure on the southwestern portion of the Good Hope Farm and no holes were drilled by RUL.

Data is not available for the targeting drilling described above and therefore the results cannot be verified. The QP is not relying on the information from the RUL targeting drilling for its interpretation of the deposit.

Three holes (OAP80, OAP81, OAP82) were drilled by RUL to the north of the main Okanjande deposit for sterilization purposes for a primary crusher site. These holes intersected mineralization in an area now referred to as the Northern Extension, and thus the position of the crusher plant was moved further away from the Okanjande deposit. The QP is not relying on the information from the RUL sterilization drilling as it is incomplete and is instead relying on verifiable data from later Gecko drilling in the Northern Extension for its interpretation and modelling of the Mineral Resource.

RUL also completed three percussion drillholes 600m to the northwest of the main Okanjande deposit for sterilization purposes for a tailings area site. The drillholes intersected mineralization, and the area became known as the "Tailings

Area Mineralized Zone” and is now referred to as the “Northern Mineralized Band”. Assay data are not available for the sterilization holes and therefore the results cannot be verified.

Percussion drilling was collared with 150mm diameter drilling, subsequent to which 115mm diameter drilling was used for the rest of the holes. RUL used HQ drilling for the diamond holes.

The drillhole collars were surveyed in a local grid system by the Rössing Mine Survey department.

10.1.1 Density Measurements

According to the RUL reports (RUL, 1993 and RUL, 1994), density measurements were completed on selected core samples at Mintek, which resulted in an average density value of 2.61t/m³ for the weathered mineralisation and 2.73t/m³ for the fresh mineralised material. The original data for these density measurements are not available. This density data was not used in the current Mineral Resource estimate, because it could not be verified.

10.2 GECKO DRILLING

The descriptions of the Gecko drilling were derived from an internal Imerys-Gecko Mineral Resource report (Imerys, 2018), e-mail communications with Gecko (2021) and from observation made on the data provided. The work described in this section was completed prior to the QP’s involvement in the Project, therefore the QP did not observe any of the processes described during their implementation.

From 2014 to 2015, Gecko drilled 91 diamond drillholes, which include infill drilling, twin drilling of the RUL drilling and exploration drilling in the fringes and outside the main Okanjande deposit. Eleven RUL percussion drillholes and two RUL diamond drillholes were twin drilled. The infill drilling resulted in approximately 25m spacing between section lines towards the centre of the deposit. Exploration drilling was in the peripheries and in the Northern and Southern Extension of the main Okanjande deposit, the “Northern Mineralized Band”, and approximately 2km northeast of the main Okanjande deposit (Figure 10-2).

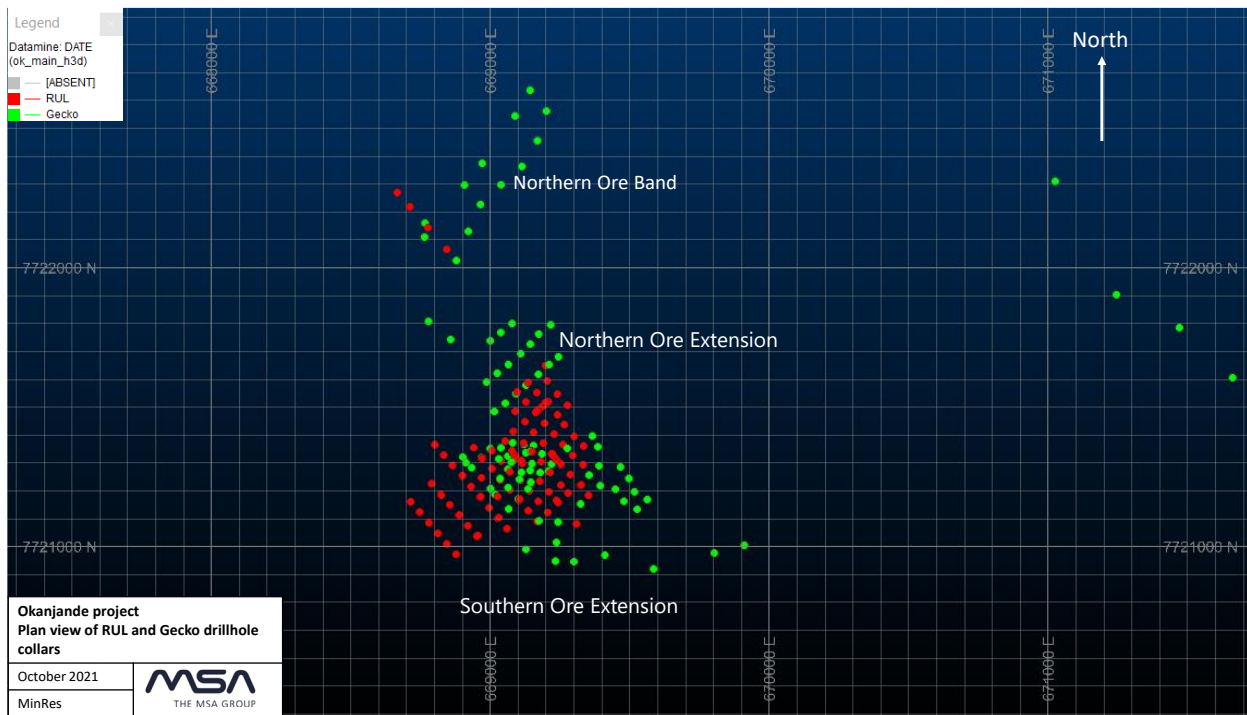


Figure 10-2: Location of Gecko drillhole collars (MSA, 2021)

All Gecko holes were undertaken by diamond drilling. Holes were drilled at an azimuth of 320° and dip of 60°, or vertical in the central parts of the Okanjande deposit, while the rest of the holes were drilled at various orientations including vertically.

According to Gecko, the drillholes were planned by the geologist on site, with advice from Kai Batla, Holdings (Pty) Ltd (“Kai Batla”) who was the advising consulting company to Gecko at the time. The drillhole collars were positioned by surveyors in the field. During drilling, cores were removed by the drilling contractor from the core barrels and placed in core boxes. The start and end of each core run was marked by the drilling contractor using plastic “stick ups”.

Core logging was completed by two geologists, Mr. Eugene van der Heever and Mr. Konrad Amunime. The geologists were trained by the consulting geologists, Mr. Pieter van Wyk and Mr. Oliver Krappmann. Mr. Pieter van Wyk supervised the team, checked their logging, marking and cutting of the core to ensure that they abided to the logging and sampling protocols. Upon completion, the core logging was first checked by Gecko’s senior geologist, Mr Gideon Kalumbu, and Mr Pieter van Wyk conducted the final checks for inconsistencies and/or errors. Core recovery was not routinely recorded, and remarks were made by exception for areas of interest.

The entire cores were photographed prior to sampling, and the full core photograph database is available in digital format, with the exception of OKD008, OKD012 and OKD013, which are missing

In 2014, Gecko contracted Mr Zaan Wichtman of African Land Survey to complete a topographic survey and to conduct collar surveys of the historical drillholes and its own holes. The survey instrument used was Trimble R8 GPS System with a TSC3 controller, and the survey was completed on the Namibian Bessel (Schwarzeck) system LO22/17 and was later converted to UTM WGS84 33S. A total of 58 Gecko holes and 40 RUL collars were surveyed by Mr Zaan Wichtman. The rest of the RUL collars were geo-referenced from historical plans. Figure 10-3 shows the drillhole collar positions as surveyed or georeferenced from plans.

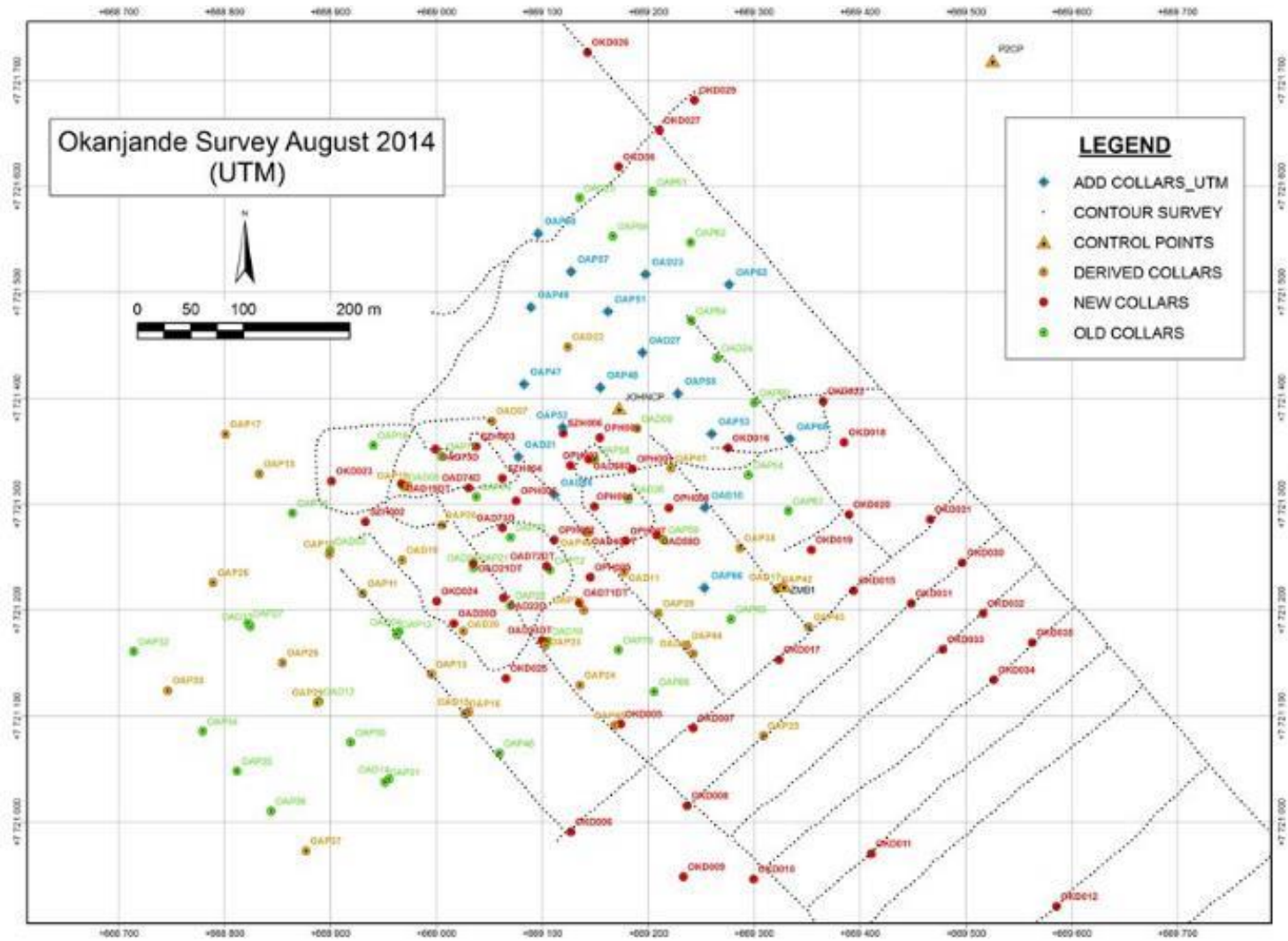


Figure 10-3: 2014 Survey plan of drillhole collars (African Land Survey, 2014)

In December 2014, Gecko contracted Mr Herman Strydom of Strydom & Associated Land Surveyors to survey an additional 27 collar positions located in the Northern Extension and in areas outside the Okanjande deposit. The collar surveys were completed using a Leica GPS900 series instrument on the Namibia Bessel System in the LO22/17 projection, which was later converted to UTM WGS84 33S.

10.2.1 Density Measurements

Density was not measured on a systematic basis during the Gecko drilling programme. A selection of twenty-three core samples were selected, of which two were fresh graphitic core, nineteen were weathered graphitic core and two were fresh waste core.

The two fresh mineralized rock density measurements are 2.74g/cm³ and 2.70g/cm³, while the two fresh waste rock density measurements are 2.58g/cm³ and 2.60g/cm³. The weathered graphitic rock density values range between 2.18g/cm³ and 2.62g/cm³ with an average of 2.40g/cm³.

There is no report on the method of density measurement and the results could not be verified. Therefore, the results were not used in the Mineral Resource estimate.

10.3 IMERYS-GECKO

The Imerys-Gecko work described in this section was completed prior to the QP's involvement in the Project, therefore the QP did not observe any of the processes described during their implementation. The Imerys-Gecko processes described in the following sections were based on the Imerys-Gecko standard operating procedures (SOPs) and email communications with Imerys-Gecko (Machoko, 2021).

From June to September 2018, Imerys-Gecko completed a total of 44 diamond drillholes within the Okanjande deposit and surrounding prospects. Twelve of these were infill holes drilled in the southwestern portion of the deposit, to reduce the drill line spacing from 100 m to 50 m (Figure 10-4). These holes were drilled at an approximate azimuth of 320° and a dip of 60° and at 50 m spacing within the drill lines. These holes are within the current Mineral Resource area.

Imerys-Gecko identified additional drilling targets from the airborne survey and mapping previously completed by Gecko:

- Imerys-Gecko completed four drillholes at the “Northern Mineralized Extension” (Figure 10-4). Two of the drillholes intersected significant mineralization (greater than 2% TGC), while the other two holes drilled 30 m away did not intersect mineralization. Imerys-Gecko recommended that follow up mapping be completed at the “Northern Mineralized Extension” to improve understanding of the local geology;
- Three shallow drillholes were completed at the “Southern Mineralized Extension” (Figure 10-4). Two of the drillholes intersected significant grade mineralization, but narrower than previously intersected by Gecko in the area. These holes may not have been drilled deep enough to intersect the full mineralized package;
- Out of the six drillholes completed at the “Northern Mineralized Band”, four drillholes intersected significant mineralization that is approximately 13m thick on average (Table 10-1), which is narrower than mineralization intersected by Gecko drillholes in the south-west part of the deposit, that was 40m thick on average (Figure 10-4);
- Seven drillholes were completed at the SW Target that were drilled on three north-west to south-east lines spaced approximately 320m apart, with a drill spacing of approximately 150m within the drillhole lines (Figure 10-4). Six holes intersected mineralization (Table 10-1). The hole that did not intersect mineralization plots in the south-east area of the target and was not sampled;

- Out of four drillholes completed by Imerys-Gecko at the Highlands Target 3 (Figure 10-4), two drillholes intersected mineralization (Table 10-1). The other two drillholes did not intersect significant mineralization; one of the drillholes was sampled and the other one was not sampled. The holes that did not intersect significant mineralization are located between the holes that intersected significant mineralization. This suggests that the mineralization is discontinuous in this area;
- Imerys-Gecko completed three drillholes approximately 5km north of the Okajande deposit at Rooibult (Figure 10-4), with only one drillhole intersecting shallow, thin significant grade mineralization (Table 10-1). The other drillholes were not sampled;
- The two drillholes completed at the Welgelegen Old Mine (Figure 10-4), approximately 12km west of the Okajande deposit, intersected mineralization over thicknesses of 21m and 35m (Table 10-1); and
- Three drillholes approximately 250m apart along an east-west line were completed approximately 6km to the west of the Okajande deposit at the Leopardkloof Target (Figure 10-4). One drillhole intersected significant mineralization and the others were not sampled. (Table 10-1).

Imerys-Gecko considered three of the targets to be of a higher priority for further drilling; the “Northern Mineralized Band” (farm Okanjande), SW Target (farm Highlands) and Old Mine (farm Welgelegen).

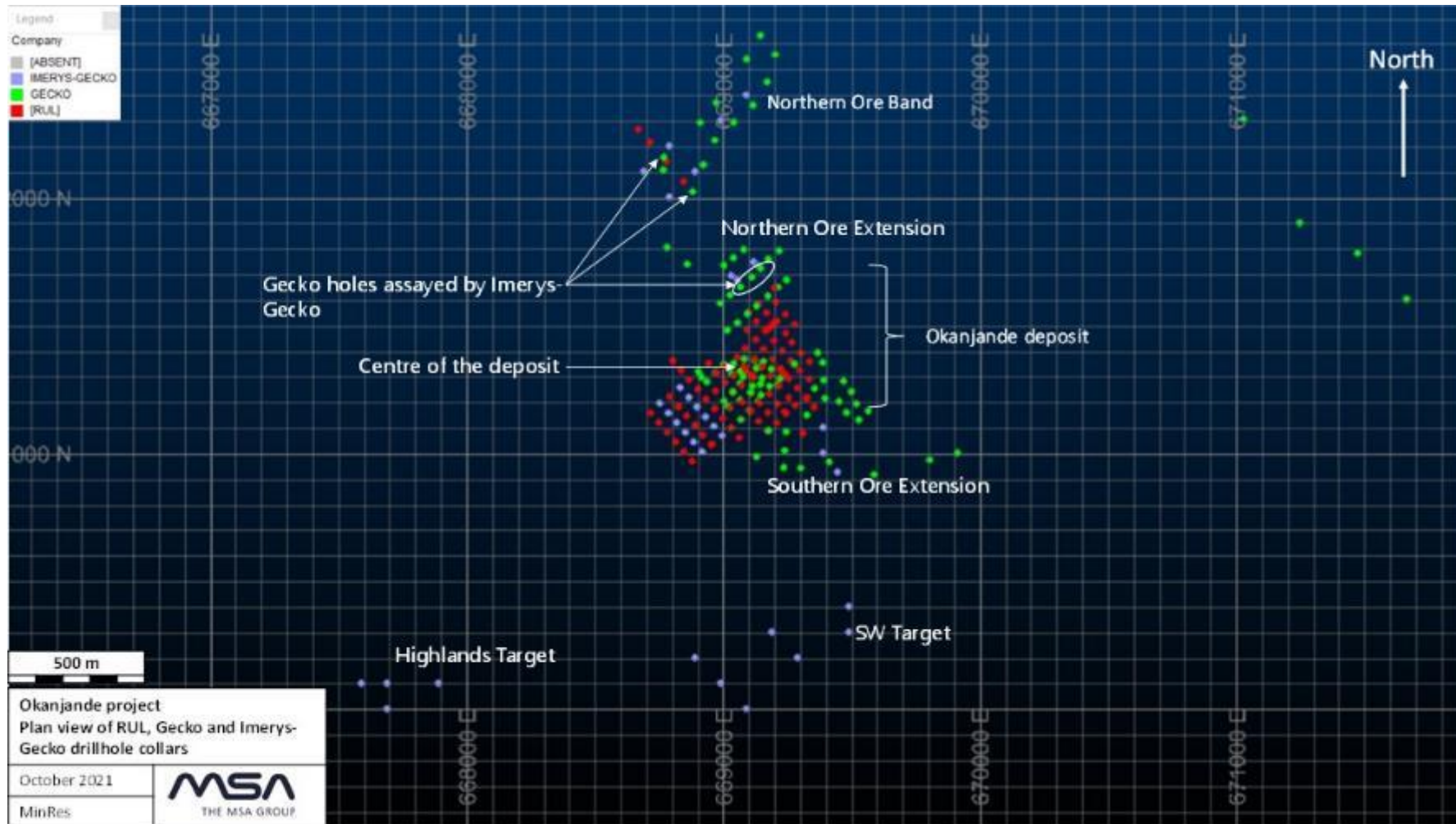


Figure 10-4: Drillhole collars of the different drilling campaigns at and around the Okanjande deposit (MSA, 2021)

Table 10-1: Imerys-Gecko significant intersections outside of the Okanjande deposit (MSA, 2021)

Prospect	Drillhole Identity	Depth From (m)	Depth To (m)	Length (m)	TGC average (%)
"Northern Mineralized Band"	OkaPBH_027	2.00	25.00	23.00	3.12
	OkaPBH_030	14.00	27.00	13.00	3.27
	OkaPBH_033	0.50	10.00	9.50	4.47
		29.00	34.40	5.40	2.47
	OkaPBH_034	10.00	28.00	18.00	5.01
"SW Target"	OkaPBH_013	0.36	6.00	5.64	5.82
	OkaPBH_014	7.00	47.20	40.20	3.70
	OkaPBH_015	26.00	29.15	3.15	7.69
	OkaPBH_017	2.00	16.00	14.00	7.29
	OkaPBH_018	5.00	27.00	22.00	3.75
	OkaPBH_019	2.00	12.00	10.00	4.58
		24.00	26.03	2.03	5.02
"Highlands Target 3"	OkaPBH_009	0.20	31.00	30.80	4.31
	OkaPBH_012	3.00	30.00	27.00	2.12
"Rooibult"	OkaPBH_037	0.16	0.16	4.00	2.15
"Welegelegen Old Mine"	OkaPBH_007	3.00	24.00	21.00	5.75
	OkaPBH_008	4.00	39.00	35.00	3.48
"Leopardkloof"	OkaPBH_002	3.00	25.00	22.00	4.13

The drillholes were collared using HQ drilling size and the rest of the holes were drilled using NQ drilling. After a drilling run was completed, the drilling contractor collected core from the core barrel and placed it in a core box, where the beginning of each run was marked on a plastic core block. The core was transported from the drill site to the logging and processing area, where a geologist checked and verified the plastic core blocks. Broken core was gathered, and core length measured. Core logging was completed by a geologist on standard log sheets. The logging included descriptions of lithology, core recovery, rock quality designation, colour, weathering, texture and mineralization. The logged information was never captured electronically.

A handheld GPS was used for the initial positioning of the drillhole collars, thereafter the surveyor picked up the collar positions prior to drilling. Imerys-Gecko contracted Brighton Zhou of Top Range Survey & Mapping cc to complete general survey work at Okanjande mine site, including the collar survey of the Imerys-Gecko exploration holes.

The drillhole data were provided by Imerys-Gecko in numerous Excel spreadsheets.

10.3.1 Density Measurements

Core density was not measured during the drilling campaign.

10.4 DENSITY MEASUREMENTS COMPLETED POST THE DRILLING CAMPAIGN

Due to the lack of sufficient verifiable density data from the three drilling programmes, the QP requested for density to be measured on a selection of drillhole cores under his supervision. A total of 144 samples were selected from eleven drillholes in the main Okanjande deposit. Densities were measured on-site and measurements for nine of the drillholes were completed under the QP's supervision. Weathered core was coated with lacquer to prevent water being absorbed into core samples. Sections of the core that had undergone secondary (i.e., post-drilling) weathering, mainly those exhibiting degradation of sulphides, were not used for density measurements.

Samples were weighed both wet and dry suspended under a digital scale (Figure 10-5). The QP is satisfied that the density data collected from the samples in this manner is acceptable for Mineral Resource estimation.



Figure 10-5: Density measurement scale and apparatus with lacquer coated core (MSA, 2021)

10.5 NORTHERN GRAPHITE DRILLING

No drilling has been conducted by or on behalf of, Northern Graphite Corporation.

11 SAMPLE PREPARATION, ANALYSES AND SECURITY

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

11.1 RUL CAMPAIGN

The following description is sourced from the RUL Feasibility Study and geological reports (RUL, 1993; RUL, 1994), and from observations made on the data provided. The work was completed historically, and the QP did not observe the processes.

Percussion drilling chips were collected at 1m intervals. The sample material was collected directly from the cyclone underflow and was homogenised by passing it through the riffle splitter three times prior to collecting the sample. A sample of 2kg to 3kg was collected from the riffle splitter and bagged for assay. The percussion chips were originally kept but are no longer available for inspection.

Diamond drillhole cores were cut longitudinally into half, then cut longitudinally in half again. The quarter core samples were collected at 2m intervals continuously through the entire length of the drillhole core. Where the core was broken and could not be cut, typically at the beginning of the hole, the broken material was homogenised and then half was taken as a sample for assay.

RUL submitted samples to Scientific Services in Windhoek for sample preparation, entailing crushing, pulverising and homogenising. Sub-samples of pulverised material were dispatched to Scientific Services in Cape Town, an independent commercial laboratory, for graphitic carbon, sulphur and titanium oxide analyses. The QP is not aware whether the two Scientific Services laboratories were accredited at the time. According to the RUL reports (RUL, 1993 and RUL, 1994), sulphur and total carbon were analysed using LECO, carbon grade in carbonate was analysed using wet chemistry (no further details provided) and titanium oxide was analysed using X-Ray Fluorescence (“XRF”). Total graphitic carbon was calculated as the difference between the total carbon and carbon in carbonate assays. A limited number of percussion holes were assayed for sulphur. Though all diamond drillholes were assayed for sulphur, there are unexplainable gaps in the sulphur assay data.

A total of 94 field duplicate samples were collected as well as 25 field duplicates for umpire assays at a second laboratory (RUL, 1993 and RUL, 1994). It is not clear which laboratory was used for the umpire assays, and the results of neither the duplicates nor the umpire samples are available. No further quality assurance and quality control (QA/QC) samples are mentioned in the RUL reports (RUL, 1993 and RUL, 1994).

There are no further details provided on sample security.

Given the lack of QA/QC applied or available, the QP is reliant on indirect QA/QC for the RUL data, i.e., by examining global bias between different exploration campaigns that had more acceptable QA/QC protocols and twin drilling (refer to section 12.2 RUL Twin Drilling).

11.2 GECKO CAMPAIGN

The following description is sourced from a report by Gecko (2014) on the sample preparation of the drillhole samples, the Bureau Veritas (BV) assay certificates, BV quotations, from e-mail communications with Gecko (Krappmann, 2021), and from observations made on the data provided. The work was completed historically, and the QP did not observe any of the processes.

The cores were logged and marked up for sampling by geologists. Sample positions were marked on the core using a permanent marker at 1 m intervals, while adhering to the geological contacts identified during logging. The cores were split longitudinally in half using a diamond saw and again into quarter core for sampling purposes. The cores were

broken along sample markings, individual samples were placed into bags with uniquely numbered sample tickets and sample numbers were written on the outside of the bags. Visibly un-mineralized intervals were not sampled. All remaining cores were stored in galvanised core trays on site in a core shed. In 2018, a secure core storage facility was built next to the Okanjande mine offices and all the cores are currently stored there.

Samples were dispatched by Gecko to Bureau Veritas (BV) in Swakopmund, a SANAS accredited independent commercial laboratory, for sample preparation. The samples were accompanied by a sample submission form completed and signed by Gecko. The sample preparation at BV Swakopmund included crushing and pulverising (no further details provided).

Upon receipt of the pulverised material from BV at Gecko's premises in Nonidas, outside of Swakopmund, Gecko homogenised the material and split it using a riffle splitter to prepare between 30 g and 50 g sub-samples for assaying. The pulverized sub-samples were incorporated into a new sample stream, which included certified reference material ("CRM") and pulp duplicates at an insertion rate of 4% for both. Blanks were inserted initially and were later discontinued. The assay results for the blanks are not available. The samples were submitted to BV Swakopmund for dispatch to BV Rustenburg.

The samples were assayed for sulphur and total graphitic carbon ("TGC") at BV in Rustenburg, which was a SANAS accredited independent commercial laboratory at the time (the laboratory is now closed). Sulphur was analysed using a LECO combustion analyser. For TGC assays, the samples were acidified and roasted to remove carbonate and organic carbon respectively. The residual carbon was determined using a LECO combustion analyser. The TGC method used is a SANAS accredited method, with the code I-9421-QA-022. The titanium oxide assays were completed at BV Swakopmund using the schedule code of PF101 which involves a sodium peroxide fusion with an Inductively Coupled Plasma - Optical Emission Spectrometry ("ICP-OES") finish. Drillhole samples from holes drilled at the periphery of the Okanjande deposit were randomly selected for sulphur assays.

Gecko used five graphitic CRMs, a titanium oxide CRM and pulp duplicates for its QA/QC samples submitted to BV in Swakopmund. The results of the TGC and S assays of the CRMs are shown in Table 11-1.

Table 11-1: CRM results for Gecko sampling (MSA, 2021)

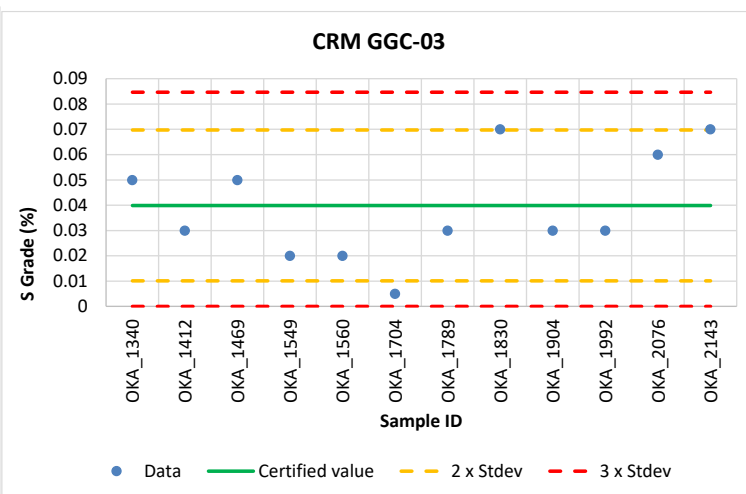
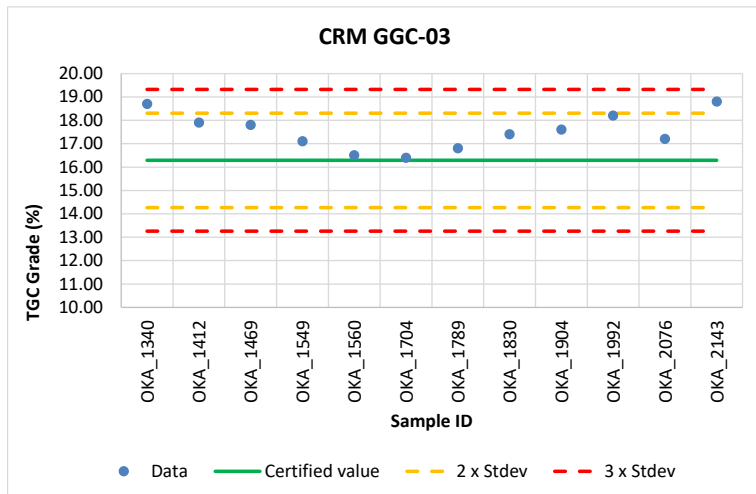
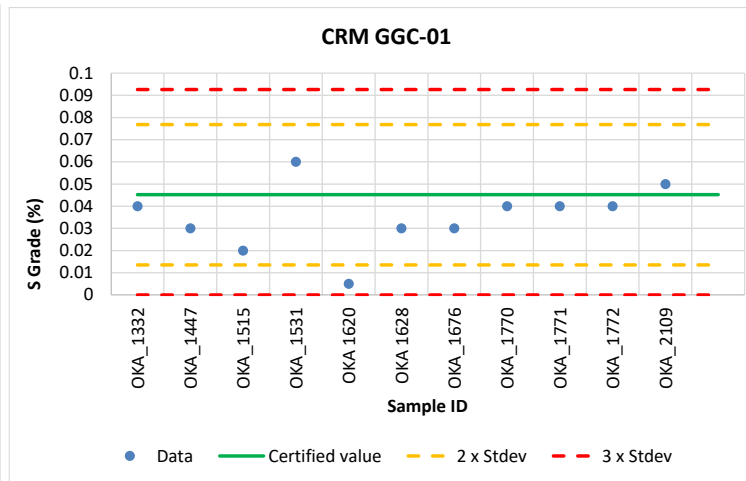
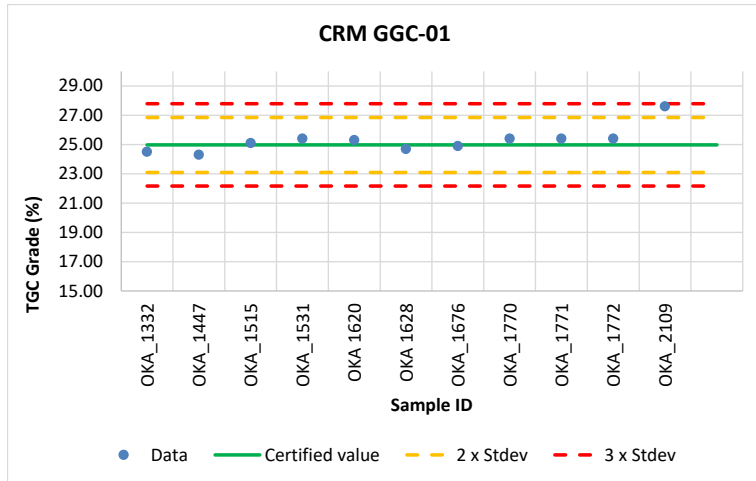
CRM name	Element	Certified value (%)	Standard Deviation	Number of samples	Bias	Number of samples outside *limits
GGC-09	TGC	2.41	0.27	37	2%	0
	S	4.59	0.13	37	**0%	7
GGC-10	TGC	4.79	0.29	10	**7%	1
	S	4.40	0.19	10	**10%	3
GGC-01	TGC	24.97	0.94	11	1%	0
	S	0.05	0.02	11	-23%	0
GGC-04	TGC	13.53	0.64	24	3%	0
	S	0.05	0.02	24	**2%	1
GGC-03	TGC	16.29	1.01	12	-7%	0
	S	0.04	0.01	12	24%	0

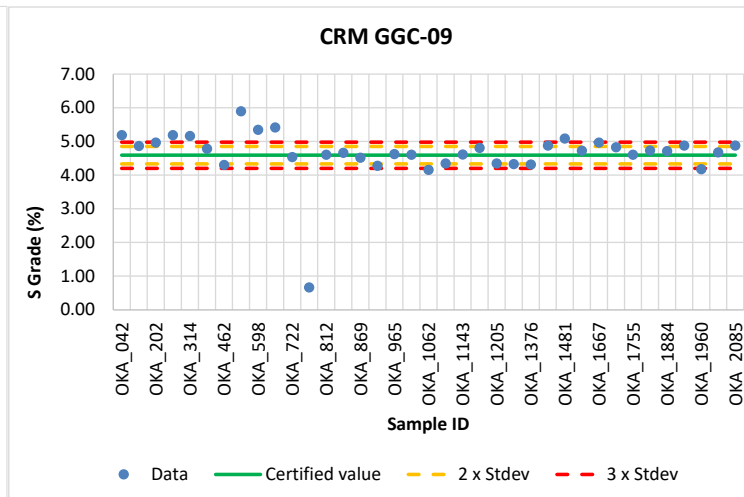
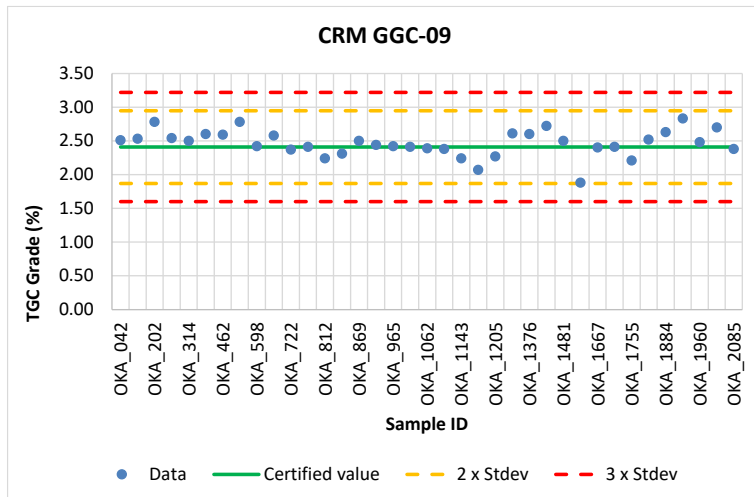
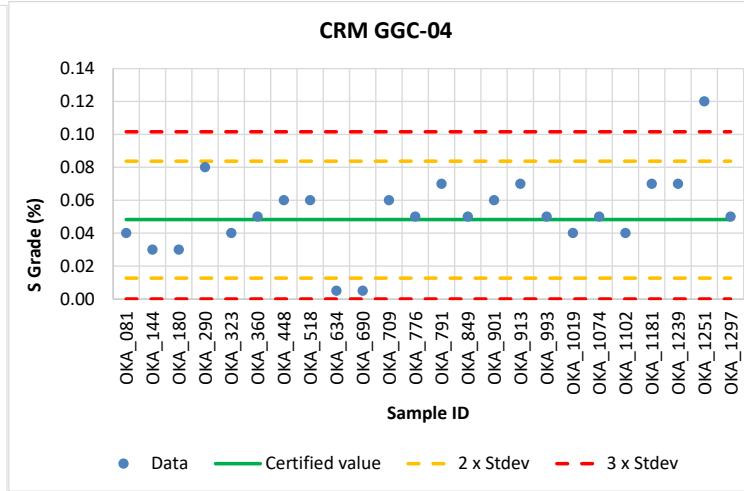
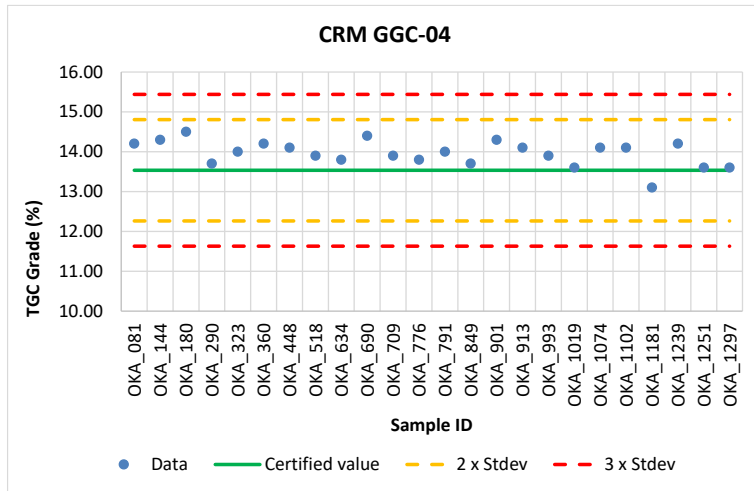
Note: *limit is three times standard deviation of the certified value
**bias is calculated after removing failing samples

The graphitic CRMs have TGC grades ranging from 2.41% to 24.97% (Table 11-1 and Figure 11-1). Four of these cover the grade ranges of the deposit, but the TGC grade of CRM GGC-01 is outside the range of the Okanjande mineralization.

The CRMs show acceptable accuracy for the TGC assays, with all the samples, except one GGC-10 sample, returning values within control limits of three times the standard deviation of the certified value. Four out of the five CRMs show a positive bias for the TGC assays of between 2% and 7%. The fifth CRM shows a negative bias of 7% (Table 11-1 and Figure 11-1).

The certified CRM sulphur grades are either low grade (0.04% / 0.05%) or relatively high grade (4.40% / 4.56%). The CRM analyses with the largest relative biases are the low grades ones, although the absolute differences are negligible (Table 11-1 and Figure 11-1). One of the higher sulphur grade CRMs has a failure rate of 19% and the other 30%, although the most frequently used one had minimal average bias. The QP is of the opinion that the sulphur results are acceptable because almost half of the assays that plot outside the three standard deviation limits fail by a small margin, and the biases are within a significant level of 10%. It is also worth noting that sulphur does not form part of the Mineral Resource but was rather estimated to allow assessment of any impact sulphur may have on the Project and the environment.





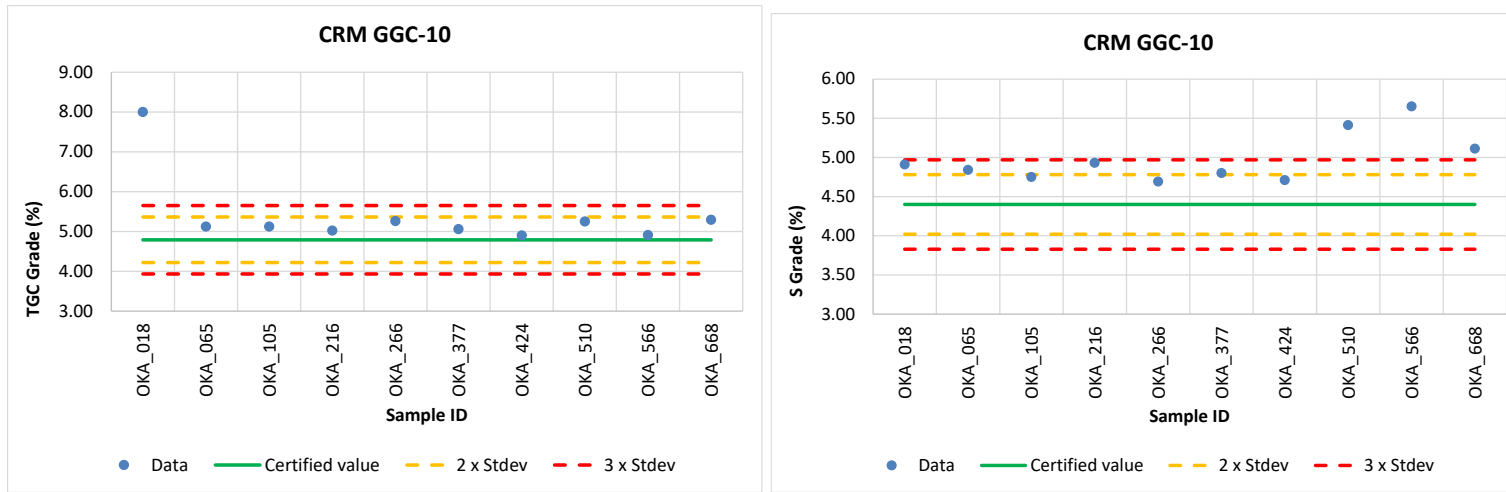


Figure 11-1: Control charts for the Gecko CRM's (MSA, 2021)

A total of 96 pulp duplicates were submitted to the laboratory for analyses with the rest of the samples. The results show good precision, where 90% of the TGC assays have a half absolute relative difference (“HARD”) of less than or equal to 2%, and 90% of the S assays have a HARD value of less than or equal to 4% (Figure 11-2). This shows better precision than the industry benchmark where 90% of pulp duplicates are expected to have a HARD value of 5% or less.

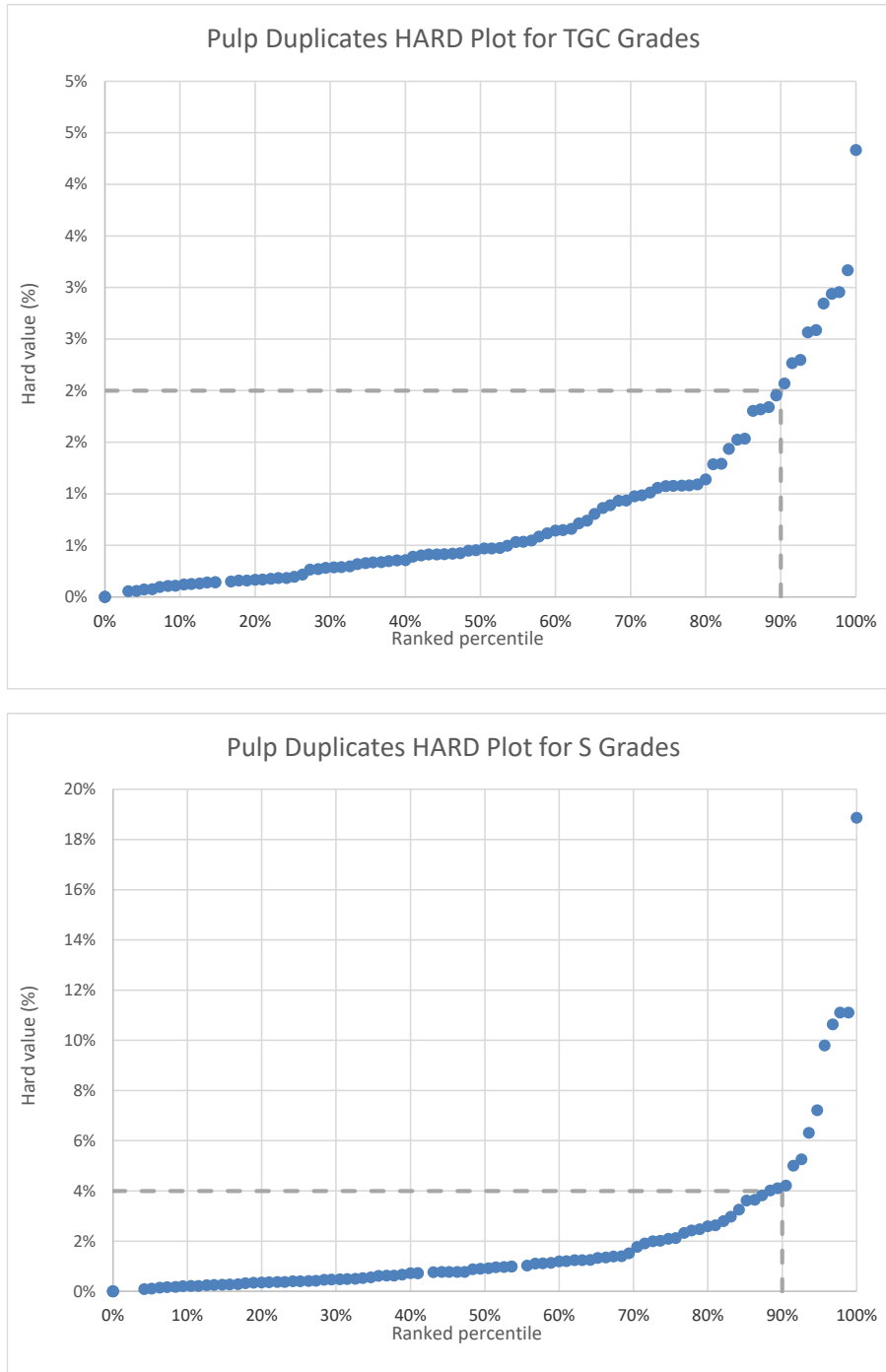


Figure 11-2: HARD plots for the Gecko pulp duplicates (MSA, 2021)

It is the QP’s opinion that the Gecko TGC and S assays have been demonstrated by QA/QC processes to be of acceptable accuracy and precision to use in Mineral Resource estimation.

11.3 IMERYS-GECKO CAMPAIGN

The descriptions of the processes in this section were sourced from the Imerys-Gecko standard operating procedures (“SOP”), BV assay certificates and from observations and interpretations made on the data provided. The work was completed historically and the QP did not observe any of the processes.

The samples from the Imerys-Gecko cores were taken at 1m intervals continuously through the entire core length while adjusting for geologically logged boundaries. The samples were recorded on a sampling sheet. The cores were cut longitudinally in half using a diamond saw and one half was cut again longitudinally in half to generate quarter core. The cores were broken at sample beginning and end markings, and individual samples were placed in plastic bags together with manila sample tags labelled with the unique sample number. The bags were sealed with twine and another sample label was tied to the twine. Visibly un-mineralized intervals were not sampled. The remaining cores are stored in galvanised steel core trays in a secure core storage facility near the Okanjande mine office.

Imerys-Gecko completed a review of the previous exploration data at the Project and identified that several holes drilled by Gecko were not sampled, although they were logged to have intersected graphite mineralization. These holes were drilled towards the edges of the deposit, but the reason for the holes not being sampled was not clear. Imerys-Gecko selected five of these holes for sampling (OKD027, OKD029, OKD036, OKD038, and OKD040) and another hole (OKD039) was re-sampled.

The Imerys-Gecko samples were transported by courier to BV Swakopmund, a SANAS accredited, independent commercial laboratory. Upon receipt of samples, the laboratory signed a form to maintain chain of custody. Sample preparation was completed at BV Swakopmund, where the samples were crushed, split and pulverised. The pulverised sub-samples were dispatched to BV Centurion (South Africa), a SANAS accredited, independent commercial laboratory, where the samples were analysed for S and TGC using method codes ACT-TPM-013 and ACT-TPM-028, respectively. The samples were analysed for S using LECO through a C-S/S Analyser. For TGC analyses, the samples were reacted with hydrochloric acid to remove the carbonate, followed by roasting of the sample at 425°C to remove the organic carbon, and the residue was then analysed using LECO through the Total Combustion by C-S Analyser. BV Centurion is SANAS accredited for both analytical methods.

Imerys-Gecko inserted duplicates as quarter core samples at a rate of 5%, for a total of 58. Provision was also made in the sampling sequence for reporting of laboratory duplicates every 12 samples. The laboratory duplicates were inserted at the laboratory. Since these duplicates are normally used for the laboratory’s internal QA/QC, these samples were not assessed in this report.

The Imerys-Gecko field duplicates show good precision for TGC and S, where 90% of the data have a HARD value of less than 15% for TGC and less than 13% for S (Figure 11-3). This is better than the industry benchmark where 90% of the data is expected to have a HARD value of 20% or less for field duplicates and demonstrates that the quarter core sample is an adequate sub-sample for this style of mineralization.

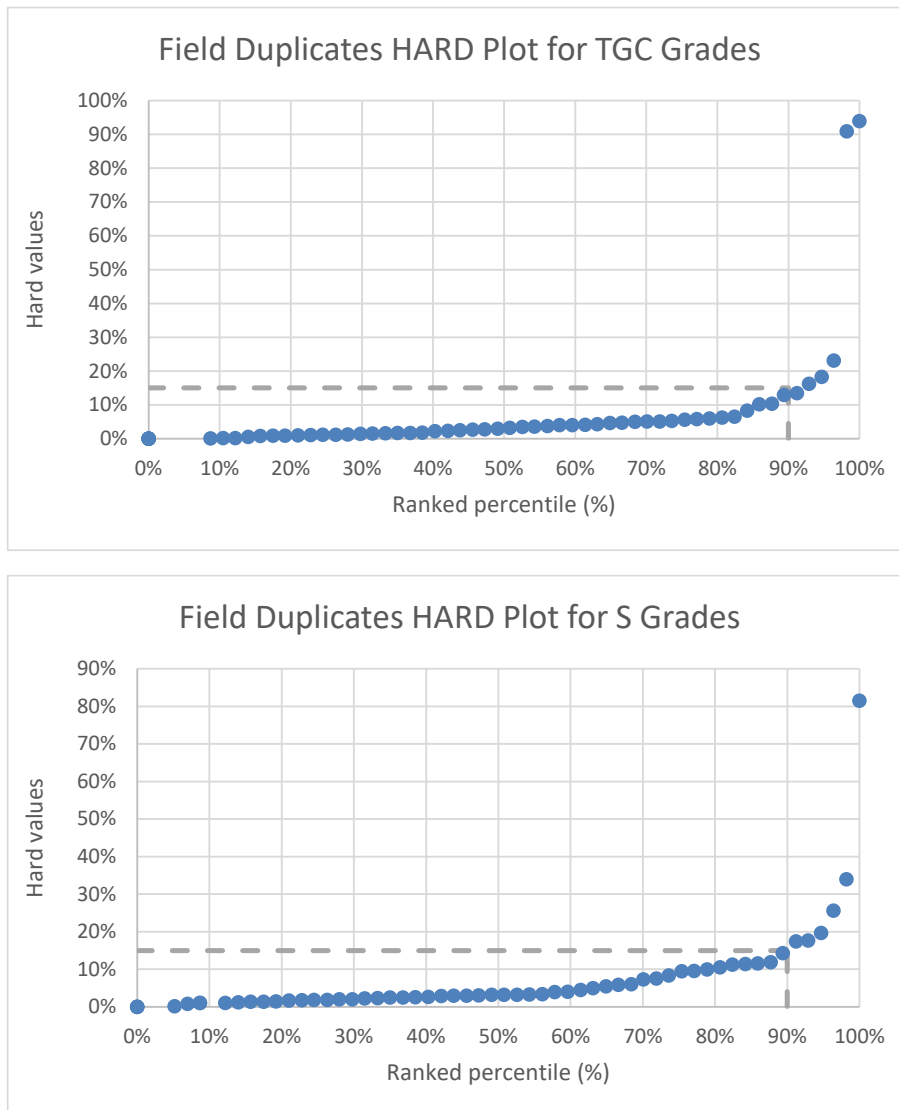


Figure 11-3: HARD plots for the Imerys-Gecko field duplicates (MSA, 2021)

The precision of the Imerys-Gecko assays is good. Accuracy was not assessed, as CRMs were not inserted and umpire analyses were not performed. The degree of contamination is not known, as blank samples were not used. The QP is therefore partly reliant on indirect QA/QC for the Imerys-Gecko data, i.e., by examining global bias between different exploration campaigns that had more acceptable QA/QC protocols (refer to Item 12 - Data Verification).

11.4 **NORTHERN GRAPHITE**

No sampling has been conducted by or on behalf of, Northern Graphite Corporation.

12 DATA VERIFICATION

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

12.1 QUALIFIED PERSON

The Geology QP, Mr Rob Barnett, established an audit trail of the electronic drilling data that is captured in several spreadsheets that were provided to MSA.

Mr Barnett also completed two site visits, one between 28th September and 1st October and the second between October 18-22, 2021.

The Mineral Resource QP, Ms Ipelo Gasela, completed the statistical analyses on the data to determine their appropriateness to use in a Mineral Resource estimation. Ms Gasela did not visit the site.

12.1.1 Electronic Data Review

A selection of original assay certificates was compared to data recorded in the spreadsheets for several drillholes (OKD016, OKJ2, OKJ5 and OPH001) and no discrepancies were found.

Weathering depths, as recorded in drillhole logs, were compared to sulphur analyses to verify the depth of weathering. Unexpected S values for the stated weathering state were observed in the data for seven drillholes. Mr Barnett compared these drillhole logs with core box photographs and found that in three drillholes the logs were accurate regarding depth of weathering, while in the other four drillholes the sulphur analyses were a more accurate measure of weathered depth.

12.1.2 Site Visits

On the first site visit, Mr Barnett inspected the graphite mineralization exposed in the Okanjande mine pit and inspected cores stored in the mine core yard. Drillhole cores of holes OKD017, OKD019 and OKD020 were compared to the drillhole logs and good correlation was found. The visual mineralogy of the graphite gneiss/quartzite was observed to be comparable to that of graphite gneiss he has observed at other graphite projects in Mozambique, Tanzania and Malawi. The graphitic quartzite at the Okanjande deposit was not observed at the other projects Mr Barnett is familiar with; however, the flake size in this lithology is similar to that occurring in other graphite gneiss.

Mr Barnett also investigated the drillhole cores of three drillholes which had graphitic mineralization logged but do not have sample records (OKD006, OKD026, OKD037 and OKJ12). He found that only OKJ12 had no mineralization worth sampling and that OKD006, OKD026 and OKD037 should be sampled as they are mineralized.

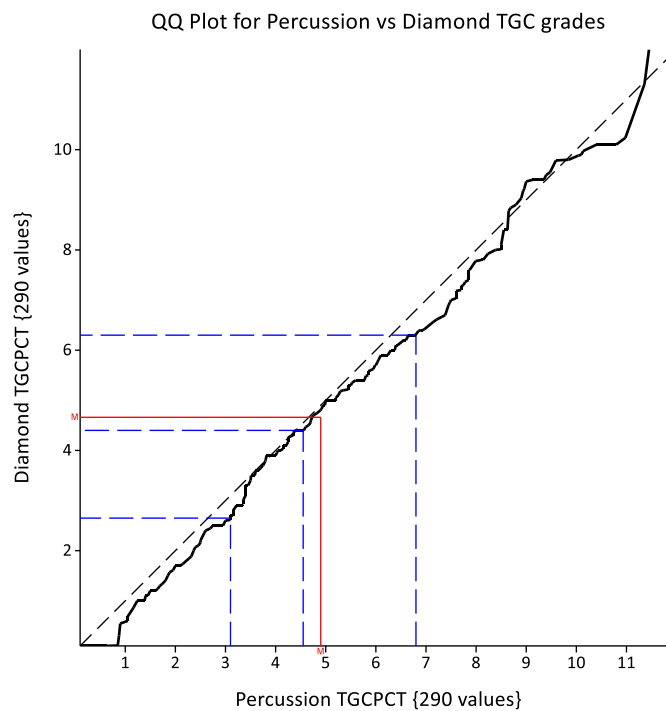
The drillhole core boxes are stored in a fenced off shed at the mine office site. The shed is roofed and has open sides. Drillhole core boxes are stacked separately for each drillhole and their locations are shown on a control plan for ease of retrieval. Core box markings are fading and it was recommended that the boxes be re-marked. Some core has undergone secondary weathering due to sulphide weathering, with the weathered core having a yellow and brown staining. In the most extreme cases the core has been altered to a clay-like texture with froth on the surfaces.

It was proposed that core density measurements should be undertaken while avoiding cores affected by secondary weathering. Density measurements became the primary purpose of the second site visit.

Mr Barnett visited the processing plant, which is “mothballed”, at Okorusu mine where he inspected flake graphite product still in stock that was sourced from Okanjande. The flake product has the typical shape and form of flake graphite products observed at other similar projects.

12.2 RUL TWIN DRILLING

MSA completed a quantile-quantile (“QQ”) plot comparing the grades of the samples obtained by RUL percussion drilling to the twin diamond drillholes completed by RUL. The QQ plot indicates that there is no bias between the sample assays of TGC from the two drilling methods (Figure 12-1). Most of the percussion holes were not assayed for sulphur, therefore only three of the twinned percussion holes that were assayed for sulphur were used in the bias test. The QQ plot shows a significant positive bias of 16% towards the percussion holes for sulphur grade. The bias is higher at depth where grades are higher and thus at the shallow depth, where most percussion drilling was undertaken, the bias is minimal overall (Figure 12-1).



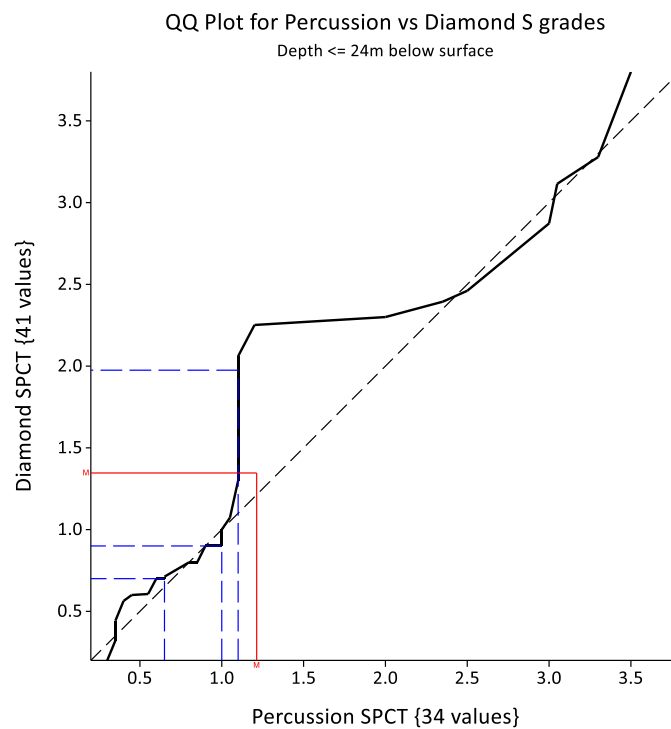
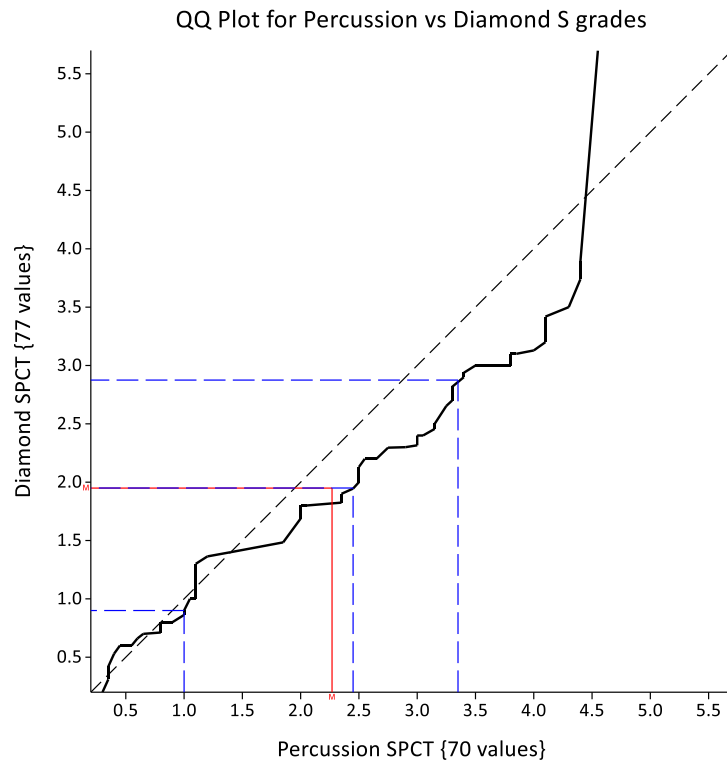


Figure 12-1: QQ plot between RUL percussion drillhole sample assays and the RUL twinned diamond drillhole sample assays – TGC (top), all S (middle), S up to 24 m depth (bottom) (MSA, 2021)

The QPs did not observe any of the RUL drilling. The results of the statistical tests from the twin drilling reveal that samples from the RUL percussion and diamond drilling have TGC grades that are not significantly biased

with respect to one another. It is therefore the QP's opinion that the two datasets can be used together in Mineral Resource estimation.

The S grades show a high bias towards the percussion samples, which appears to be more significant at greater depths where S grades are higher. There are only two percussion holes at depths exceeding 24m, therefore the inclusion of percussion data in the estimation has limited risk on the S estimates at depth.

12.3 GECKO VERIFICATION WORK

12.3.1 Gecko Check Sampling of RUL Drillhole Samples

Gecko completed a check sampling programme on a selection of the remaining core intersections drilled by RUL. Gecko completed the check sampling by collecting 1m samples of quarter core from the remaining RUL core, and a total of 129 check samples were assayed. The samples were assayed for TGC and S at Bureau Veritas in Rustenburg (South Africa), which was an independent and accredited commercial laboratory at the time. The remaining cores were photographed. The Gecko check samples were not accompanied by QA/QC samples.

For statistical comparison with the RUL data, the Gecko check sample assays were composited to 2 m composites. The same intervals corresponding to the RUL 2m samples were not always sampled by Gecko therefore not all intervals in the selected holes were compared.

The TGC assays show a strong positive relationship between the original and check sample assays, though with some scatter, resulting in a correlation co-efficient of approximately 0.94. A scatter is expected as the check sampling was carried out on quarter cores and some differences between two sides of a core can be expected. The two datasets show almost no bias to one another, evidenced by a best fit line of $y = 0.9888x$ (Figure 12-2).

The check sample S assays show a strong positive relationship with the original assays, with some scatter, resulting in a correlation co-efficient of 0.95. The check sample assays show a positive bias against the original assays of approximately 20%, which results in a best fit linear equation of $y = 1.19x$. This indicates that the sulphur grade is potentially under-estimated in the RUL data, or over-estimated in the Gecko check assays.

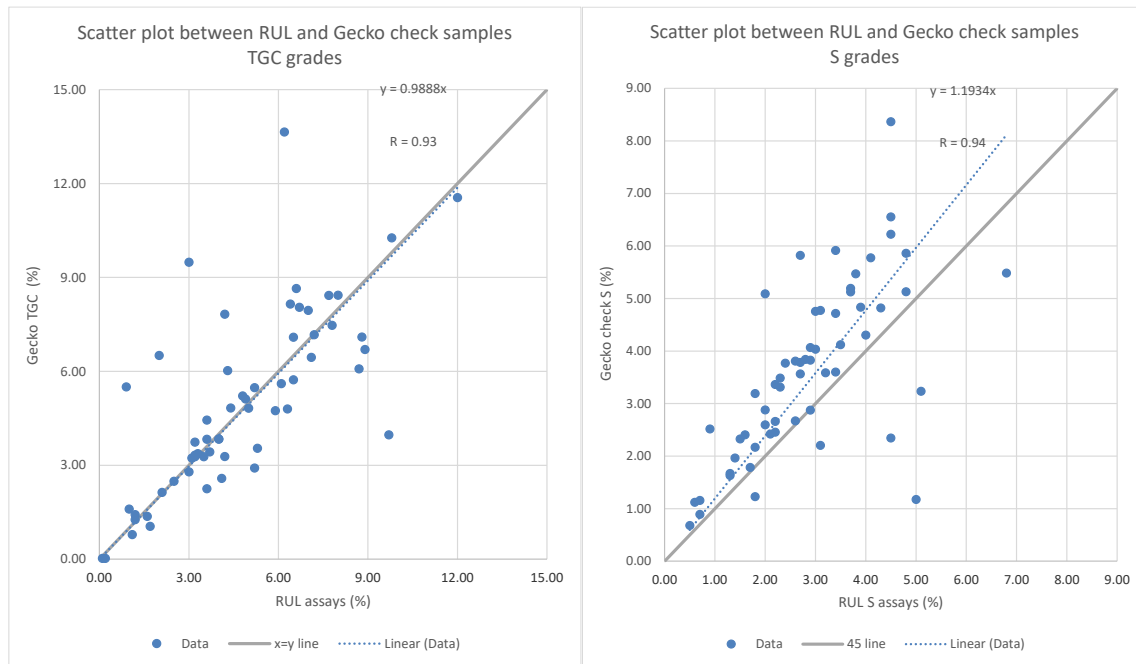


Figure 12-2: Scatter plot of the assays of the Gecko check samples versus RUL samples - TGC (left) and S (right) (MSA, 2021)

The QPs did not take the verification samples and only completed a statistical check of the check sample assays against the original samples. It is the QP’s opinion that the TGC assays of the RUL samples compare within acceptable limits to the Gecko check samples. However, there is a significant bias between the RUL and Gecko check sample sulphur assays.

12.3.2 Gecko Twin Drilling of RUL Drillholes

Gecko completed twin drilling of 11 RUL percussion holes and two diamond drillholes. The Gecko twin drillhole samples included samples of three CRMs; namely CRM GGC-09, CRM GGC-10 and CRM GGC-04. All but one of the CRM samples returned assays within the acceptance limit of three-standard deviations of the certified value but with minor biases (from 3% to 7%) to the certified values. The QP is of the opinion that the CRMs demonstrated good accuracy of the TGC assays.

The CRM sulphur assays show poor results, with 42% of the GGC-09 CRM and 33% of the GGC-10 CRM sample assays outside the acceptance limit of three-standard deviations of the certified value (Table 12-1 and Figure 12-3). Most of the failures are by a small margin (Table 12-1 and Figure 12-3). The average biases of the assays of the two high grade sulphur CRMs are less than 10% of the certified values, after the removal of the failures. It is therefore the QP’s opinion that the CRMs show adequate accuracy for S assays. Sulphur is not part of the Mineral Resource, but the grades were estimated to be able to assess any impact sulphur may have on the Project.

Table 12-1: CRM results for Gecko sampling for the twin drilling (MSA, 2021)

CRM name	Element	Certified value (%)	Standard Deviation	Number of samples	Bias	Number of samples outside *limits
GGC-09	TGC	2.41	0.27	12	3%	0
	S	4.59	0.13	12	**0%	5
GGC-10	TGC	4.79	0.29	9	**7%	1
	S	4.40	0.19	9	**9%	3
GGC-04	TGC	13.53	0.64	9	4%	0
	S	0.05	0.02	9	-17%	0

Note: *limit is three times standard deviation of the certified value
 **bias is calculated after removing failing samples

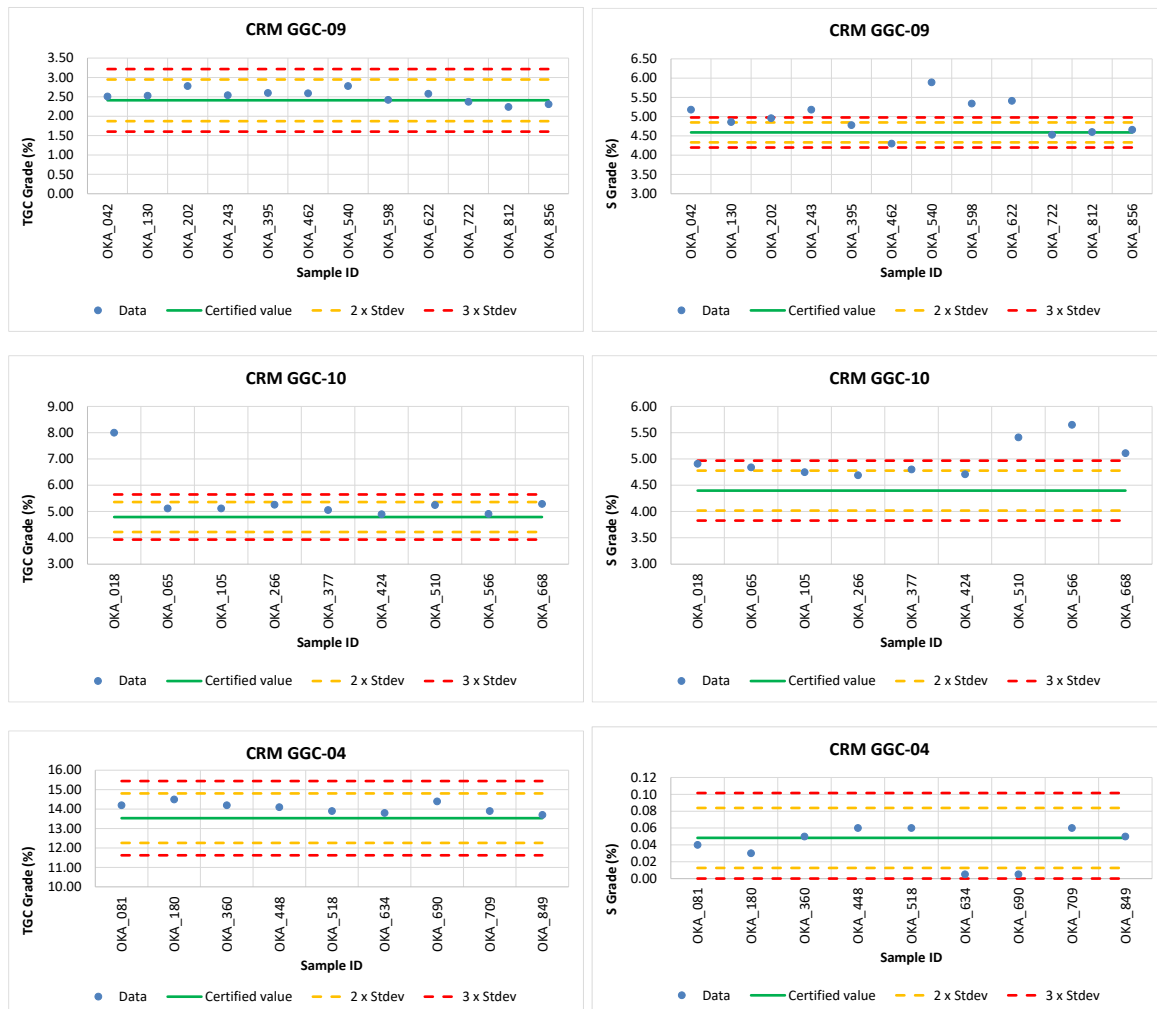


Figure 12-3: Control charts for the CRMs used for Gecko twin drillhole sample assay QA/QC (MSA, 2021)

For statistical test purposes, the data were composited to 2m intervals and a threshold of 2% TGC was applied to both sets of data to account for un-mineralized intervals that were not sampled in the Gecko drilling campaign.

The QQ plot of the paired hole TGC assays demonstrate that there is minor overall bias between TGC sample assays of the RUL and Gecko drilling (Figure 12-4). Due to the limited number of sulphur assays on RUL drillhole samples, only the two twinned RUL diamond drillholes had sulphur assays which could be used in the QQ plot. The QQ plot shows insignificant bias in the sulphur grades of the two datasets (Figure 12-4).

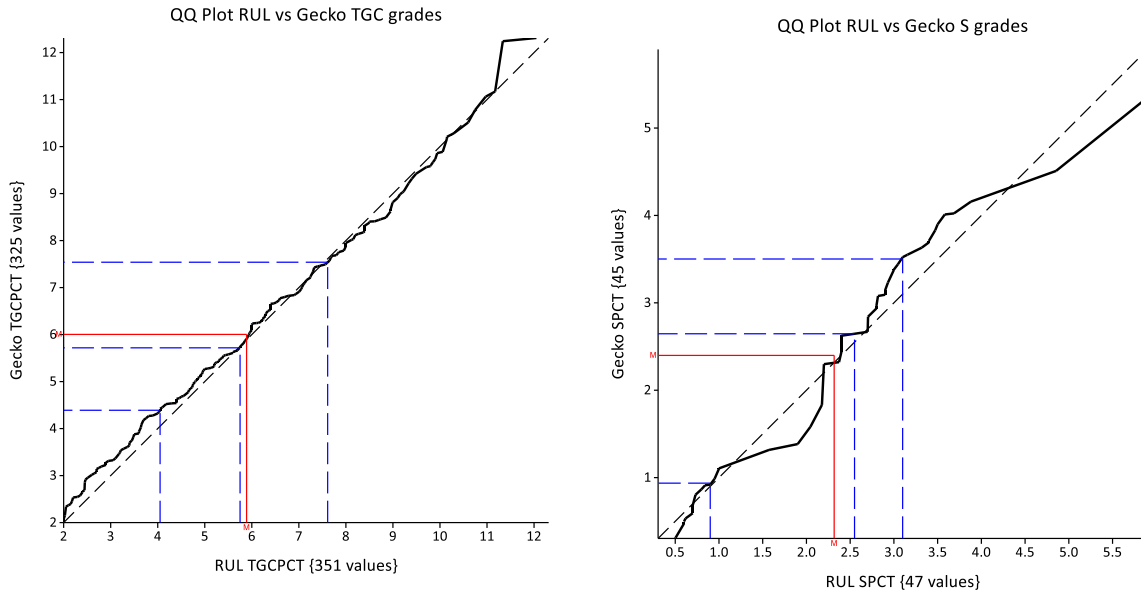


Figure 12-4: Quantile-quantile (QQ) plot between the RUL and Gecko twin drilling sample assays– TGC (left) and S (right) (MSA, 2021)

The QPs did not observe or supervise the twin drilling programme and only completed the statistical tests, which indicate that there is no significant bias between the RUL and Gecko sample TGC assays. Therefore, it is the QP’s opinion that the two TGC datasets verify each other and are suitable to use in a Mineral Resource estimation. The RUL diamond drillhole and the Gecko twin drillhole sulphur grades also do not show significant bias with respect to one another. Therefore, it is the QP’s opinion that the two datasets are considered acceptable to use in a Mineral Resource estimation.

12.4 IMERYS-GECKO VERIFICATION

12.4.1 Imerys-Gecko Check Sampling of Gecko Drillholes

In 2018, Imerys-Gecko completed check sampling of OKD039, a Gecko diamond drillhole. A good linear correlation was observed between the original Gecko sample assays and the Imerys-Gecko check sample assays for TGC and S, with correlation coefficients of 0.99 and 0.98, respectively. However, the original Gecko TGC assays are on average approximately 6% higher than the Imerys-Gecko check sample assays, with a tendency to be lower in the low-grade ranges and higher in the high-grade ranges. The sulphur assays are on average 8% higher than the Imerys-Gecko check sample assays with a tendency to be higher for assays greater than 2% (Figure 12-5).

The QP did not observe or carry out the check sampling and only completed the statistical analyses. In the QP’s opinion, biases between the Gecko and Imerys-Gecko assays are not material.

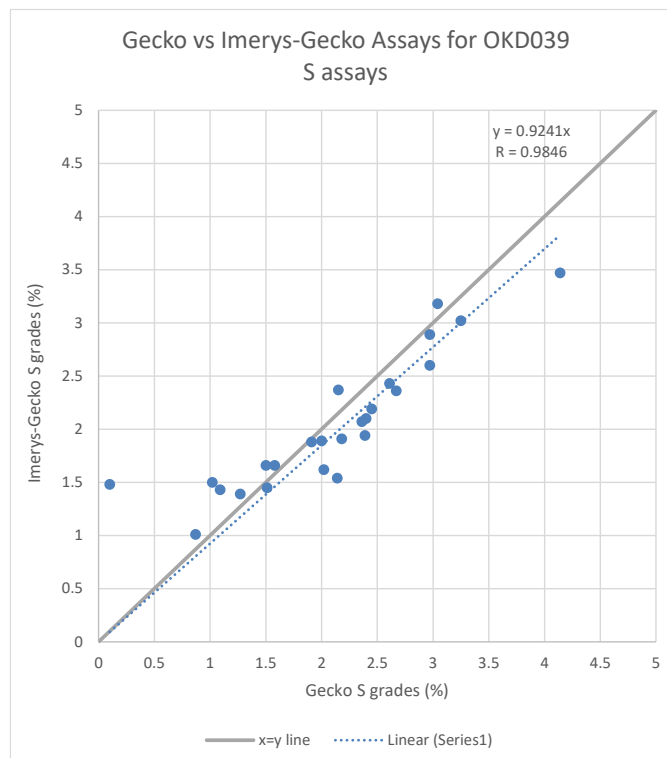
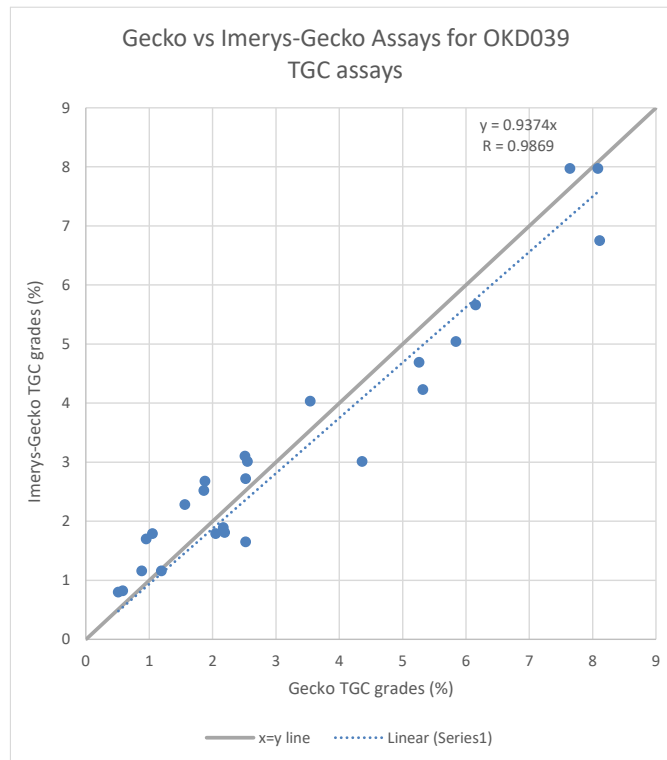


Figure 12-5: Scatter plot for OKD039 check sample assays - TGC (top) and S (bottom) (MSA, 2021)

12.4.2 Bias Test Between Imerys-Gecko and RUL Drilling

The QP completed a bias test between the TGC assays from the Imerys-Gecko drillhole samples and the RUL drillhole samples in the southwestern part of the deposit. The data were composited to 2m intervals and a threshold of 2% TGC applied for both sets of data to account for un-mineralized intervals that were not sampled in the Imerys-Gecko drilling campaign.

The TGC QQ plot shows that the two datasets have similar TGC grade distributions, with the Imerys-Gecko average TGC grade being 2.5% higher. The QQ plot between the RUL and Imerys-Gecko sample sulphur grades show slight conditional bias (Figure 12-6) and the Imerys-Gecko average sulphur grade is 4% lower.

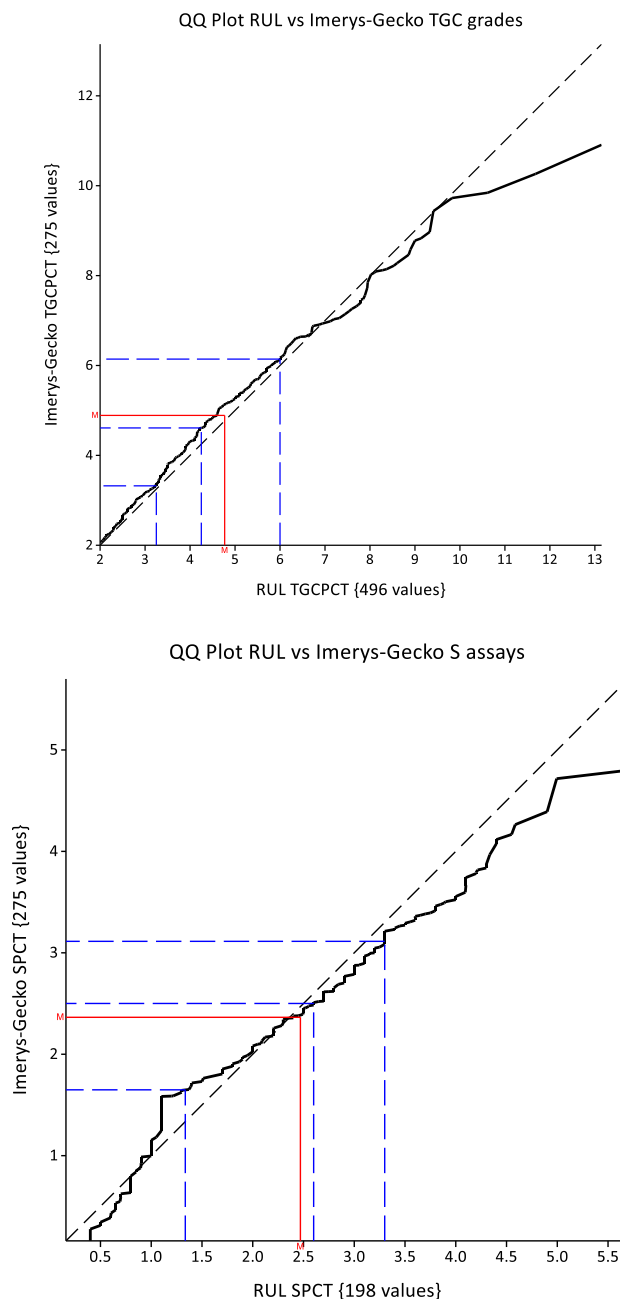


Figure 12-6: QQ plot between Imerys-Gecko and RUL drillhole sample assays in the southwest part of the Okanjande deposit – TGC (top) and S (bottom) (MSA, 2021)

The QPs did not observe or supervise any of the drilling programmes and only completed the statistical tests.

There is no significant bias between the Imerys-Gecko and the RUL TGC and S drillhole sample assays, therefore it is the QP's opinion that the two datasets verify each other, and this allows for the two datasets to be used together in the Mineral Resource estimation process.

13 MINERAL PROCESSING AND METALLURGICAL TESTING

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

13.1 EARLY PROCESS TEST WORK – MINTEK

Early investigations were carried out by Rossing Uranium Limited and Mintek in South Africa in the period 1989-1992. Salient outcomes of the early test work programs are given below:

13.1.1 Preliminary investigation into the upgrading of Graphite Ore, 1989

Batch tests produced high grade graphite concentrate – 93% Carbon

Recommended flowsheet from test campaign: Rod Milling, Rougher Flotation, three Pebble Mill re-grinding stages, three stages of flotation and magnetic separation.

Middling fractions (cleaner tails, WHIMS magnetics): 55% recovery at 88.7% carbon content. Best performance Magnetics middling's: to be recycled to second pebble mill. Cleaner Tails: showed benefit of recycling middling's.

13.1.2 Preliminary tests to recover flake graphite from the Okanjande deposit, 1990

A flow sheet was tested consisting of Rod Milling, Rougher Flotation, Pebble Mill regrind and 3 stages for cleaner flotation.

Samples from 2 distinct areas were test ranging in head grades from 9.5 to 10.5% fixed carbon as graphite.

Concentrates in the region of 93% C were produced from both areas.

A Flash Flotation stage was added, which increased the yield of >300um flakes, but did not affect the yield of >106um flakes.

Single stage versus 2 stage pebble regrind was investigated and determined that 2 stages of regrinding does not improve graphite concentrate.

Pebble mill versus Delaminator was tested as regrind stage and it was determined that the Delaminator produced higher grade graphite in all fractions than the Pebble mill, but reduced the fraction of >106 flake by 12%. Details regarding the Delaminator are not available.

13.1.3 Preliminary tests on drill core samples from the Okanjande deposit to recover flake graphite, 1991

Milling and flotation tests completed on 32 drill core samples with an average head grade of 6% fixed carbon as graphite and produced concentrate grades ranging from 83% to 96% at recoveries in excess of 80%.

Optimal flow sheet developed from the test program was Rod Milling, Rougher Flotation, Pebble Mill regrinding of rougher concentrate and 3 stages of cleaner flotation.

Low pH in the Rougher Float circuit for altered mineralized material samples resulted in poor selectivity, but this was improved to produce a rougher concentrate of 81% carbon at a recovery of 93,4% through pH adjustment to 10 utilizing sodium hydroxide.

High intensity magnetic separation failed to increase concentrate grade of large flakes containing sulfur/sulfides.

13.1.4 Metallurgical Commissioning of the Okanjande Graphite Beneficiation Pilot Plant, 1991

Rossing Uranium ran a pilot campaign on the Okanjande deposit to recover flake graphite in 1991. Mintek assisted in the metallurgical commissioning of the pilot plant. General observations and conclusions from the pilot campaign:

- Crushing of the Okanjande mineralized material led to much dust generation and loss of graphite product on windy days. Dust control required at crushing, transfer points and stockpiles.
- Closed circuit rod milling with 2mm classification screen applied without sanding in rougher cells. High recovery obtained from rougher cells.
- Pebble milling applied for polishing.
- Cleaning flotation noted free gangue, but investigation indicated graphite attachments leading to floating of gangue. More vigorous cleaning flotation conditions recommended (higher impeller speeds, shorter residence times).
- TEB was used as flotation reagent.
- Free and visible gangue was a constant problem in the final concentrate. Further polishing milling and gravity separation stages were recommended to improve the quality of the final concentrate. Gravity separation was shown to be more effective for larger flake sizes. Remaining gangue was mainly mica.
- Best process flow sheet tested was judged to be: Rod Milling with 2mm classification followed by 2 stages of Rougher flotation and one stage of Scavenger flotation. Rougher concentrate was polished in a Pebble Mill followed by 6 stages of Cleaner flotation with internal recirculation.

13.1.5 Autogenous Milling and Flotation of Okanjande Ore, 1992

Weathered and Fresh mineralized material samples proved to be amenable to autogenous milling during laboratory testing in 1992. Closed circuit milling with 2mm screen classification was used and resulting in 40% circulating loads and a screen undersize P80 of 425um.

Autogenous milled samples produced similar flotation performance than rod milled samples, but fresh mineralized material autogenous milled samples produced coarser graphite flake sizes than Rod milling. Weathered mineralized material produced coarser graphite flakes in a crushing and rod milling circuit, indicating rod milling damaged flakes while grinding of fresh mineralized material due to hard quartz grits.

Fresh Mineralized material milling indices for AG and SAG milling were found to be: 2.5kWh/t and 3.1kWh/t respectively.

Weathered mineralized material drop tests indicate high friability, but with enough competent rocks in between to allow for autogenous milling.

Two distinct pebble types were distinguished: relatively pure milky white quartz and high graphitic content (10% TGC) black pebbles. The high graphite pebbles were shown to wear significantly faster than the white quartz pebbles, indicating that the AG mill will result in a load of mainly milky quartz pebbles which could be ideal grinding media in a pebble mill.

Semi-Autogenous milling produced significant graphite flake damage.

13.1.6 The Purification of Okanjande Graphite Concentrate: Exploratory Work, 1992

Exploratory test work was conducted in the pursuit of producing 99.5 and up to 99.9% concentrate purity.

Roasting of graphite concentrate with either sodium hydroxide or sodium carbonate, followed by successive leaching with water, sulfuric acid and hydrofluoric acid produced greater than 99.9% purity.

Roasting of graphite concentrate with sodium hydroxide followed by successive leaching with water and sulphuric acid produced graphite concentrate with greater than 99.5% purity.

13.1.7 Pneumatic Flotation of Graphite at Okanjande, 1992

Column and Jameson float cells were testing in the pilot plant at various functional locations. In general, it was found that the column cells produced slightly better recoveries at similar concentrate grades to that obtained from the Jameson cells, but required additional frother addition.

Oxidation of sulphides in the tailings facilities resulted in low pH return water, which necessitated pH adjustment to prevent poor flotation response.

Sulphide recovery to concentrate from fresh mineralized material remained a problem throughout and indicated floatability of sulphide minerals present in the fresh mineralized material. Oxide mineralized material in the column cell could not produce higher than 91% grade.

Increasing froth washing in the column cell could not increase concentrate grade, indicating incomplete liberation of graphite flakes.

13.2 EARLY MINERALOGICAL CHARACTERIZATION - MINTEK

Mineralogical investigations were carried out by MINTEK over the period of 1991 to 1992.

13.2.1 A Mineralogical Examination of the Okanjande Graphite Ores, March 1991

Seventeen borehole core intersections were tested. The predominant minerals were Quartz, K-Feldspar, Graphite, Mica, Pyrite, Pyrrhotite and Sillimanite. Mean graphite flakes sizes were found to be in the region of 0.5-0.7mm while maximum flake size ranging from 1.0 to 2.0mm were observed.

Oxidized samples were found to contain notable amounts of jarosite and other iron oxides and hydroxides.

Two distinct mineralized material textures were described:

1. Altered Mineralized material with distorted graphite flakes which contained closely intergrown gangue minerals. Deformation appeared to be the result of shearing and deformation of the mineralized material. Full liberation of graphite expected to be difficult to achieve.
2. Un-Altered mineralized material containing undistorted and largely inclusion-free graphite.

The most common phase interlaminated with graphite flakes was Muscovite and Sericite. Sulphide minerals were also frequently found to be interlaminated with graphite.

It is recommended to identify the shear zone in the mineralized material body which would contain the altered mineralized material with poor beneficiation characteristics.

13.2.2 A Mineralogical Investigation of the Sulphide minerals present in graphite concentrates from Okanjande, July 1991

Graphite concentrates prepared from Okanjande mineralized material were examined for the occurrence and nature of sulphide minerals.

The sulphide phases present were Pyrite, Pyrrhotite, Chalcopyrite and Molybdenite. The sulphide phases were present as free grains as well as occlusions within the graphite flakes. The fraction of occluded sulphides was the highest for the coarse graphite (>300um) fraction and the lowest in the fine graphite (38um).

Chalcopyrite and Molybdenum were found to preferentially occur in the fine fractions of graphite, perhaps due to their milling characteristics.

13.2.3 A Mineralogical Investigation of the Sulphate Minerals present in graphite Ore samples from Okanjande, July 1991

Fifteen samples of Okanjande graphite ores were examined. The main sulphate phases present were Jarosite, Barite, Gypsum, Siderotil and Melanterite. Siderotil and Melanterite are water soluble and mainly found in shear and fractures as well the altered mineralized material zones at depth.

The iron sulphate minerals are mainly present in the weathered mineralized material, but do occasionally occur in fresh mineralized material along fractures and joints.

The presence of iron sulphates are the result of weathering.

13.3 RECENT TEST WORK

13.3.1 Gecko MINTEK test program 2015

The test work program comprised comminution characterisation, mineralogy and flotation test work to develop a base case flowsheet. The flowsheet development test work was conducted on weathered material and the final flowsheet was tested on the fresh material.

13.3.1.1 Flotation Testing

The optimal processing flowsheet selected (Figure 13-1) consisted of rougher and cleaner flotation cells with scavengers on both tailings to maximise recoveries. The cleaner concentrate was milled in a pebble mill to liberate graphite flakes attached to gangue with minimum flake breakage. A double stage pebble mill was required for maximum graphitic carbon recovery. The base case flowsheet utilised DOW 200 as frother without the addition of a collector.

Test work on the Weathered material showed that it was possible to obtain 95% Graphitic C at 85% recovery. The inclusion of the -53mm size fraction into the final product is critical to obtain the overall recovery required. This fraction should form part of the cleaning flotation stages after the first pebble mill. Graphitic carbon recoveries obtained on the fresh material were low at 67% at the reagent dosages tested, but this could possibly be optimized.

The JKTech drop weight and abrasion tests provide the mineralized material specific parameters for use in the JKSimMet Mineral Processing Simulator software. In JKSimMet, these parameters are combined with pilot data, equipment details and operating conditions to analyse and/or predict milling and crusher performance.

The results are given in Table 13-2. The classifications are relative to all the other samples in the JKTech data base. The value for Axb gives the resistance of the sample to impact breakage. Guidance for the interpretation of Axb values are given in Table 13-3. The smaller this value is, the greater the resistance to impact breakage. The weathered graphite sample depicted an A*b value of 85.50 indicating that the sample was soft.

Resistance to abrasion is indicated by the Ta parameter in the Table above. Smaller value of Ta indicates more resistance to abrasion breakage. The Ta value of the weathered graphite sample was 0.79 indicating that the sample was soft.

Table 13-2: JKTech Dropweight Results - Weathered

Sample	SG	Ta	A	b	Axb
Weathered	2.38	0.79	67.3	1.27	85.5
Classification:	67 to 127 (70-90 Percentile) = SOFT				

Table 13-3: JKTech Hardness Classification

JKTech DB Percentile	JKTech Hardness Classification
Hardest 10% (A*b range ~ 0 – 30)	Very Hard
10% to 30% (A*b range ~ 30 to 39)	Hard
30% to 40% (A*b range ~ 39 to 43)	Moderately Hard
40% to 60% (A*b range ~ 43 to 56)	Medium
60% to 70% (A*b range ~ 56 to 67)	Moderately Soft
70% to 90% (A*b range ~ 67 to 127)	Soft
90% to 100% (A*b range ~ 127 to 956)	Very Soft

13.3.1.2.2 SAG Milling Comminution (SMC) Tests

JKTech Dropweight tests were completed on Weathered and Fresh samples. The SMC test uses the same device as the JKTech drop weight tester. The SMC test generates a Drop weight index (DWi), which is a measure of the rock strength when broken under impact. The parameters obtained from the testing are again used in the JK SAG mill models to predict throughput, power draw and product size distribution with the aid of their database of information on the performance of continuously fed pilot and production scale mills. The results are given in Table 13-4.

The value for Axb gives the resistance of the sample to impact breakage. The smaller this value is, the greater the resistance to impact breakage. Guidance for the interpretation of Axb values are given in Table 13-3. The weathered and fresh samples depicted Axb values of 57.6 and 57.2 respectively indicating that the sample were moderately soft.

Table 13-4: SMC Test Results - Weathered and Fresh

Sample	SG	DWi	Mla	A	b	Axb	Classification
Weathered	2.42	4.19	14.90	68.60	0.84	57.60	Moderately Soft
Fresh	2.7	4.70	14.70	65.00	0.88	57.20	Moderately Soft

13.3.1.2.3 Uniaxial compressive strength (UCS) tests

The UCS tests were only completed on Weathered samples. The Uni-axial compressive strength (UCS) test investigates the competency of a rock sample by assessing how a particular material breaks in a mill under an applied load. The Uni-axial compressive strength is a function of the maximum force recorded at the point of ductile/brittle transition of a sample under slow compression. The results are given in Table 13-5.

The UCS test results indicated that the sample was soft with an average uniaxial compressive strength value of 46.7Mpa. From the UCS guidance for interpretation of results the weathered samples can be classified as Soft. See Table 13-6.

Table 13-5: UCS Test Results - Weathered

Sample	UCS Minimum (MPa)	UCS Average (MPa)	UCS Maximum (MPa)
Weathered	25.4	46.7	65.0
Classification:		SOFT	

Table 13-6: UCS Test Classification Guide

UCS (MPa)	50-100	100-150	150-250	>250
Classification	Soft	Medium	Hard	Very Hard

13.3.1.2.4 Bond Crushability Work Index (CWI) test

The Bond crushability tests were only conducted on Weathered samples. This test is conducted to obtain a measure of the materials crushability (CWi), which is expressed as the energy required to accomplish a given crushing operation. Individual rocks are placed on a holder within striking distance of two opposing swinging hammers. By adjusting the height of the hammers or impact angle, it is possible to measure the breakage or crushing strength of the rock and calculate the impact work index. The test was conducted on the weathered material at a feed size range of -76 mm to + 50 mm. The results are given in Table 13-7. The average CWi index of 12kWh/t corresponds to a soft crushing material following the guidance in Table 13-8.

Table 13-7: CWI test results - Weathered

Sample	CWi Minimum kWh/t	CWi Average kWh/t	CWi Maximum kWh/t
Weathered	10.5	12.0	13.3

Table 13-8: BWi Classification Guide

CWi (kWh/t)	<10	10-14	14-18	18-22	>22
Classification	Very Soft	Soft	Average	Difficult	Very Difficult

13.3.1.2.5 Bond Abrasion Index (Ai) tests

Abrasion index tests were conducted on both weathered and fresh samples. The test is used to obtain an indication of the relative abrasiveness of a mineralized material compared to a standard metal sample. The sample is placed in an apparatus with a rotating paddle which impacts the sample. The measure of mass loss of the paddle over a determined time is used to calculate an abrasion index.

The abrasion test results are given in Table 13-9. Based on their average Ai values of 0.23 and 0.28 respectively, both the Fresh and Weathered are Slightly Abrasive.

Table 13-9: Ai Test Results - Weathered and Fresh

Sample	Paddle Mass Before (g)	Paddles Mass After Test (g)	Abrasion Index Ai	Life Factor LF
Weathered	93.75	93.52	0.23	1.93
Fresh	90.72	90.44	0.28	1.67

Table 13-10: Ai Classification Guide

Ai	<0.1	0.1-0.4	0.4-0.6	0.6-0.8	>0.8
Classification	Non-Abrasive	Slightly Abrasive	Medium Abrasive	Very Abrasive	Extremely Abrasive

13.3.1.2.6 Bond Rod Work Index (BRWI) tests

Bond Rod Work Index tests were conducted on both weathered and fresh samples. The Bond rod work index test was performed to provide information that will quantify the energy requirements and performance for open circuit rod milling. A special tilting Bond rod mill was used for this test. The test results are presented in Table 13-11.

The Weathered sample Bond Rod Work Indices were 10.78kWh/t and 12.06kWh/t for the limiting screen sizes of 1180µm and 600µm respectively, indicating that it is Medium Hard with respect to Rod Milling.

The Fresh sample Bond Rod Work Indices were 10.32kWh/t and 10.51kWh/t for the limiting screen sizes of 1180µm and 600µm respectively, indicating that it is Medium Hard with respect to Rod Milling.

Table 13-11: BRWi Test Results: Weathered and Fresh

Sample	Limiting Screen (μm)	F80 (μm)	P80 (μm)	Nett Production (g/rev)	BRWi (kWh/t)
Weathered	1180	10,501	840	13.3	10.78
Weathered	600	9,848	454	7.5	12.06
Fresh	1180	8,588	800	14.4	10.32
Fresh	600	9,515	456	9.5	10.51

Table 13-12: BRWi and BBWi Classification Guide

BRWi (kWh/t)	7 - 9	10 - 14	15 - 20	>20
Classification	Soft	Medium	Hard	Very Hard

13.3.1.2.7 Bond Ball Work Index (BBWi)

Bond Ball Mill Index tests were conducted on weathered and fresh samples. This test is aimed at providing useful information for the design of grinding circuits, and in particular, to estimate the energy requirements for closed circuit ball milling. It is also used to predict and continually evaluate the performance of commercial ball mills. The test results are presented in Table 13-13.

The Weathered sample produced a Bond Ball Mill index of 11.60 kWh/t, indicating that it is Medium hard in terms of ball milling. The Fresh sample produced a Bond Ball Mill Index of 9.48kWh/t, indicating it is Soft to Medium hard in terms of ball milling.

Table 13-13: BBWi Test Results: Weathered and Fresh

Sample	Limiting Screen (μm)	F80 (μm)	P80 (μm)	Nett Production (g/rev)	BBWi (kWh/t)
Weathered	600	2,557	446	4.64	11.60
Fresh	600	2,453	459	6.25	9.48

13.3.1.2.8 Summary of Hardness Testing Results

The results of the hardness test conducted in 2015 forms basis for the process design criteria for the comminution circuit design and is presented in Table 13-14.

Table 13-14: Summarized Hardness Indices

Hardness Test	Samples	Result
JK Tech Dropweight	Weathered	Axb = 85.5 (Soft)
SAG Milling Comminution Tests (SMC)	Weathered and Fresh	Weathered Axb = 57.6 (Moderate) Fresh Axb = 57.2 (Moderate)
Uni-Axial Compressive Strength (UCS)	Weathered	UCS = 65.5MPa (Soft)
Bond Crushability Index (CWi)	Weathered	CWi = 12.0kWh/t (Soft)
Bond Abrasion Index (Ai)	Weathered and Fresh	Weathered Ai = 0.23 (Slightly Abrasive) Fresh Ai = 0.28 (Slightly Abrasive)
Bond Rod Mill Work Index (BRWi)	Weathered and Fresh	Weathered BRWi = 11.60kWh/t (Medium Hard) Fresh BRWi = 9.48kWh/t (Soft-Medium Hard)
Bond Ball Mill Work Index (BBWi)	Weathered and Fresh	Weathered BBWi = 12.06kWh/t (Medium Hard) Fresh BBWi = 10.51kWh/t (Soft-Medium Hard)

13.3.1.3 Mineralogy

The aim of the mineralogical investigations was to determine the liberation characteristics and to determine the resulting processing equipment. A representative pulverized sub-sample was analysed using qualitative X-Ray diffraction (XRD) to obtain the bulk mineralogical assemblage. Polished sections of each size were prepared for examination by optical microscope with the aim of establishing textures and liberation characteristics of the graphite flakes and their association with gangue minerals.

Test work products were analysed using the following methods: Total C organic C, proximate analysis (high grade concentrates) and inductively coupled plasma optical emission spectrometry (ICP-OES) on feed samples.

The bulk mineralogical composition of the crushed weathered feed sample indicated major gangue minerals to be quartz, mica and feldspar in varying amounts. Sulphur is contributed as a result of the presence of various sulphides including chalcopyrite, pyrite and pyrrhotite. In addition, sulphur is contributed by jarosite, which is an oxidation product of these sulphides. Rutile occurs in amounts less than the detection limit for XRD, but the presence of rutile was confirmed using scanning electron microscope (SEM) analysis as less than 1% by mass.

Table 13-15: Bulk Mineral Analysis with SEM observations

Mineral	Formula	Relative Abundance
Quartz	SiO ₂	xxx
K-Feldspar	KAlSi ₃ O ₈	xx
Plagioclase	(Ca,Na)(Si,Al) ₄ O ₈	xx
Mica	KAl ₂ (Si,Al) ₄ O ₃ (OH) ₂	xx
Clinocllore	Mg ₅ Al(Si,Al) ₄ O ₁₀ (OH) ₈	x
Graphite	C	xx
Amphibole	Ca ₂ [Mg ₄ (Al,Al,Fe)]Si ₇ AlO ₂₂ (OH) ₂	x
Chalcopyrite	CuFeS ₂	x
Pyrite	FeS ₂	y
Pyrrhotite	FeS	y
Sphalerite	(Zn,Fe)S	y
Fe-oxide/hydroxide	FeOOH/Fe ₂ O ₃	y
Rutile	TiO ₂	y
Zircon	ZrSiO ₄	y
Kaolinite	KFe ₃ (OH) ₆ (SO ₄) ₂	y
Barite	BaSO ₄	y
Monazite	(Ce,La,Nd,Th)PO ₄	y
Ba K-Feldspar	(Ba,K)AlSiO ₈	y
Sillimanite	Al ₂ SiO ₅	y

xxx = 20-50% mass;

xx = 5-20% mass;

x = <5% mass,

y = below detection but observed by SEM

13.3.1.4 Liberation

Liberation characterization of the crushed samples indicated that graphite flake liberation is inversely related to flake size with ~50% liberation in the 500-850µm size and greater than 90% liberation in the minus 200µm flake size. See Table 13-16.

Table 13-16: Liberation Characteristics of Graphite flakes in Jaw Crusher Product

Size Fraction (µm)	% Locked / Attached	% Liberated
1700 – 3350	99	1
850 – 1700	93	7
500 – 850	51	49
425 – 500	32	68
212 – 425	31	69
106 – 212	6	94
<106	<2	>98

Rod milling tests indicated that graphite flake liberation increases with milling time, with the liberation results of the 15-minute milling test, shown in Table 13-17.

Table 13-17: Liberation Characteristics Graphite Flakes on Rod Milled Product (15min milling time)

Size Fraction (µm)	% Liberated	% Attached	% Locked
>600	98.77	0.96	0.27
500 – 600	98.40	1.60	0.00
425 – 500	96.81	3.19	0.00
300 – 425	94.20	3.99	1.81
38 – 300	100.00	0.00	0.00

The first cleaner concentrate of the base case flow sheet was analysed. Flake size fraction above 425µm contained between 19-24% gangue by volume. Size fractions below 425µm contained 10-15% gangue by volume.

Gangue minerals observed in the concentrate include clay (kaolinite), quartz, plagioclase, K-feldspar, pyrite and goethite. Scanning electron microscopy shows that gangue occurs interlayered between the graphite flakes and delamination may not separate all the gangue minerals from graphite. Separation of this type of gangue might be possible by chemical leaching.

Table 13-18: Liberation Characteristics of Graphite Flakes in 1st Cleaner Concentrate (MINTEK Base Case Flowsheet)

Size Fraction (µm)	% Fully Liberated	% Liberated + Inclusions	% Attached	% Locked
>500	77.2	17.7	3.0	2.0
425 – 500	76.7	20.1	2.0	1.2
300 -425	75.8	21.9	1.4	0.9
212 – 300	80.8	17.8	1.2	0.2
150 – 212	81.2	18.1	0.7	0.0
106 – 150	77.2	22.2	0.3	0.2
53 – 106	84.3	15.7	0.0	0.0
45 – 53	80.2	19.3	0.6	0.0
38 – 45	86.4	13.0	0.6	0.0
25 – 38	89.9	10.1	0.0	0.0
<25	93.7	6.3	0.0	0.0

The test program concluded that the optimum liberation size for graphite flakes is 200µm. And primary grind size of 80% passing 425µm was recommended as feed to the flotation plant.

13.3.1.5 Conclusions and Recommendations of the Test program

- 200µm was selected as the optimum liberation size based on the mineralogy alone. However, to preserve coarse graphite flakes, 80%-425µm grind was selected as the optimum graphitic carbon liberation grind.
- A large proportion of graphite reported to the -53µm size fraction (down to -25µm), pointing out that it will not be possible to produce a discardable barren fines fraction.
- Sulphur grades in the weathered material was shown to not be of concern, since grades repeatedly reported well below the targeted 0.1% S.

- It is possible to produce a final concentrate product at 95% Graphitic Carbon content and with a recovery of 85% when processing (floating) the -53 μ m fraction.
- All flotation tests were performed using frother only (Dow200).
- Reagent optimisation test work, pH control test work and Mineralogy on the rougher tails of the fresh mineralized material is recommended to determine if recoveries could be improved.

13.3.2 Flowsheet Confirmation Test Work, METPRO/SGS 2021

Data reviewed by METPRO suggested that a more conventional graphite flotation flowsheet similar to the Bisset Creek process would be more suitable for the Okanjande mineralization. It was postulated that two polishing stages may be beneficial to maximize the +180 μ m yield. The suitability of the proposed revised flowsheet was evaluated in a series of seven cleaner flotation tests that were carried out at SGS Lakefield using one weathered and one fresh sample, which were shipped from site. The weathered sample was a composite of four random samples taken from the mineralized material stockpile at the Okorusu plant. The fresh sample was a composite taken from six drill holes at depths between 55 and 71m.

The first six cleaner flotation tests were carried out on the weathered sample and focused primarily on the cleaning circuit. The objective of the tests was to achieve a combined graphite concentrate grade of at least 95% C(t) while minimizing flake degradation. A secondary objective was to design the flowsheet to maximize the use of the existing equipment.

Process variables that were investigated in the flotation program included:

- Elimination of ball mill and scavenger flotation
- Single and double polishing mill
- Classification of intermediate concentrate and SMM grinding
- Cleaning circuit without classification of intermediate concentrate

The first test (NA-1) investigated the degree of liberation of the +300 μ m graphite flakes after the primary grind and a very short polishing grind. The +300 μ m concentrate yield was high at 18.4% but this mass is inflated by a lot of gangue material, as it graded a low 54.8% C. The -300 μ m concentrate was subjected to a second polishing stage, cleaned, and then classified at 180 μ m. Even after two stages of polishing, the -180 μ m concentrate failed the minimum grade requirements. These results confirm the assumption that the graphite flakes are poorly liberated after the primary grind.

The next set of tests (NA-2 to NA-4) evaluated different grind times in the polishing and SMM grinding applications with classification of the intermediate concentrate prior to SMM grinding. These three tests produced consistent results.

The purpose of test NA-5 was to determine if the flowsheet could be simplified by eliminating the classification of the intermediate concentrate and treated the entire intermediate concentrate in one SMM. However, this resulted in noticeably higher +48 mesh and -48/+80 mesh flake degradation.

The flowsheet was then simplified by eliminating the ball mill and scavenger flotation stages in test NA-6 which produced a very high graphite concentrate and good flake size distribution but resulted in elevated graphite losses. Recoveries declined from 92.5% to 85.1%. The graphite losses may be reduced with more aggressive froth removal and/or longer rougher flotation retention time. Further, graphite units in intermediate streams are considered losses during open circuit testing and a large percentage of these graphite units are recovered

into the concentrate during closed circuit operation. These potential changes (removal of ball milling and scavenger flotation) were not implemented in the final process flow sheet.

All tests employed a simplified reagent regime of diesel and MIBC without the use of a gangue dispersant. An aliphatic alcohol frother such as MIBC has the benefit of a less persistent froth, which reduces the risk of elevated gangue entrainment.

Table 13-19: Summary of SGS Flotation Tests (METPRO/SGS, 2021)

Test	Sample	Combined Conc Grade %C(t)	Recovery %C(t)	% Mass >300µm	% Mass -300µm/ +180µm
NA-1	Weathered	84.6	86.4	18.4	40.8
NA-2	Weathered	95.6	88.1	11.4	42.8
NA-3	Weathered	94.9	92.7	11.6	40.7
NA-4	Weathered	96.3	91.6	9.9	37.8
NA-5	Weathered	95.5	92.5	8.6	34.3
NA-6	Weathered	96.9	85.1	10.0	41.3
NA-7	Fresh	95.6	92.4	21.2	46.3

The combined concentrates of the seven cleaner tests were submitted for a size fraction analysis (SFA) to better illustrate the impact of circuit modifications on the flake size distribution and grades. The mass recovery into the various size fractions and the associated total carbon grades of the SFAs are presented in Figure 13-2 and Figure 13-3, respectively. The charts illustrate the overall robust and consistent response of the weathered material to changes in the grinding conditions between tests NA-2 and NA-6. The much coarser flake size distribution of the fresh material is evident in Figure 13-2.

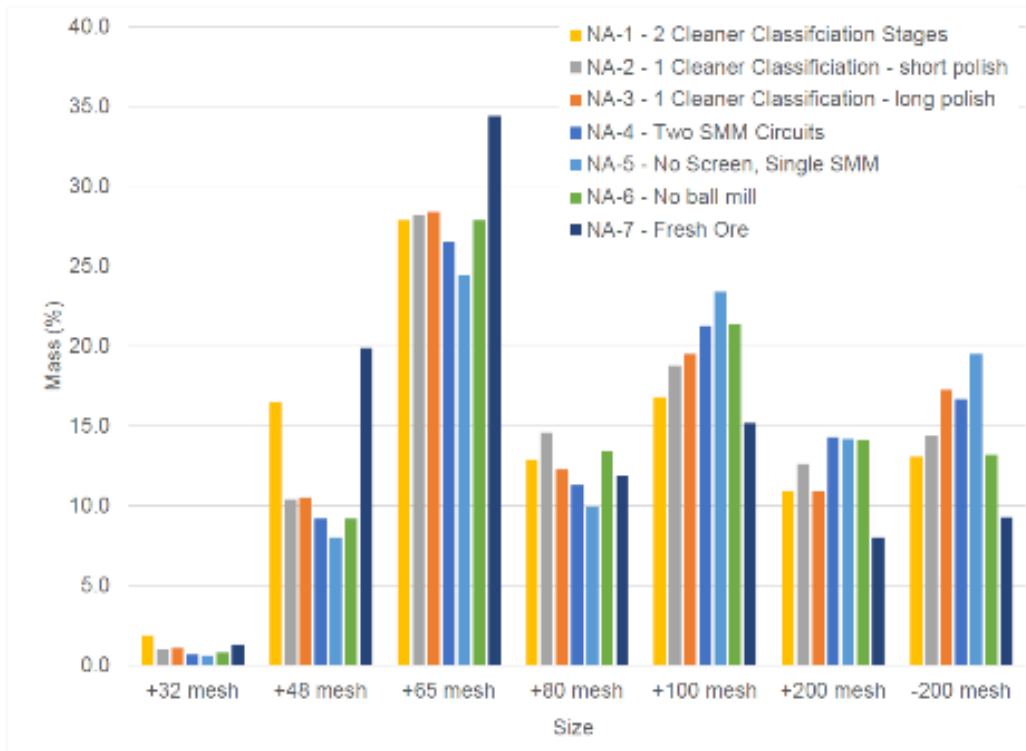


Figure 13-2: SGS Tests Concentrate Mass Distribution (METPRO/SGS, 2021)

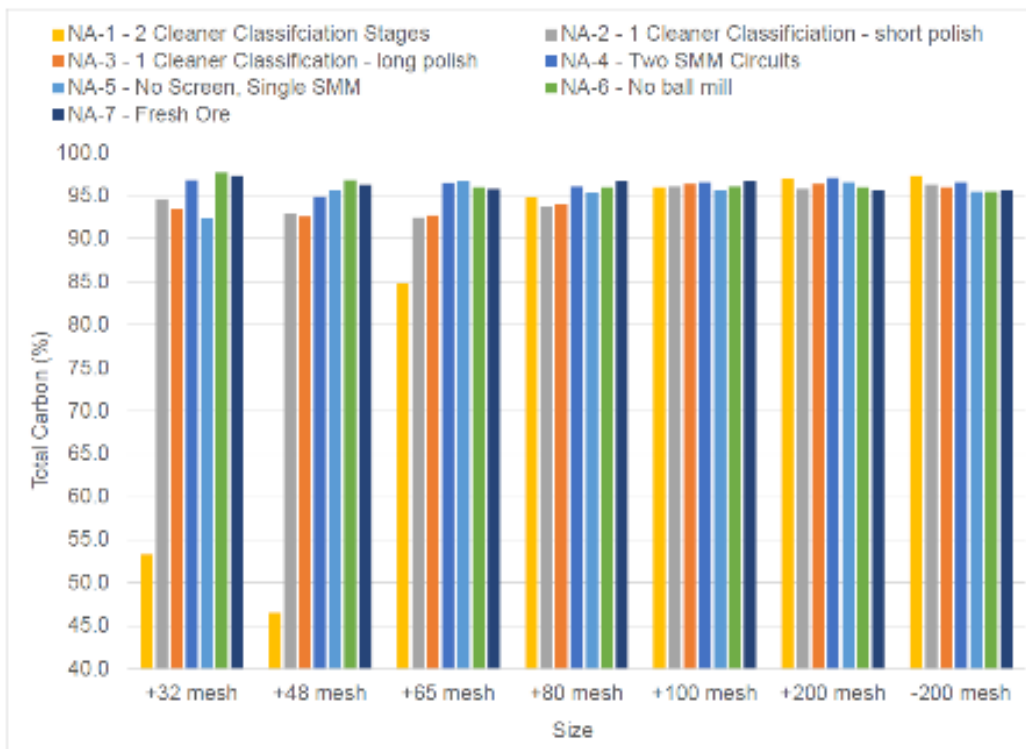


Figure 13-3: SGS tests Concentrate Carbon Mass Distribution (METPRO/SGS, 2021)

A comparison of the flake size distribution of the combined concentrate of the operating plant, Mintek laboratory and pilot testing, and the flotation testing performed at SGS is presented in Table 13-20. The SGS program matched and exceeded the results of the original Mintek work and confirmed that the current plant flowsheet is not suitable for treating the Okanjande mineralization.

Table 13-20: A Comparison of Graphite Flake Distribution between Plant and Lab (METPRO, 2021)

Flake Size (µm)	Okorusu Plant 2017-2018 (%)	MINTEK Lab tests (%)	MINTEK Pilot Plant (%)	SGS Weathered tests (%)	SGS Fresh Test (%)
>300	2	5	13	10	21
>180	16	35	35	41	46
>106	37	26	23	30	20
<106	45	34	29	19	13

13.3.3 Tailing Filtration Tests

Relocation of the processing plant to the Okanjande mine site necessitated a new approach for tailings disposal. Due to the limited water supply available at the Okanjande site, dry tailings disposal was selected. A plate and frame filter press plant was selected for dewatering duty, due to the ability to achieve very low moisture in discharged filter cake.

Filtration test work was completed on tails samples taken from the existing tailings storage facility at Okorusu site. The tailings filtration test work was completed by Latham Filtration Technology in South Africa, during February 2023 as part of plate and frame filter press selection and sizing for their budgetary proposal for a tailings filtration plant. Results of the filtration tests are given in Table 13-21. The filter tests were run on a scale plate and frame test unit.

Table 13-21: Tails Filtration Test Details

Slurry Feed		
Dry solids loading	%	22.5
Sg of slurry	t/m ³	1.18
Filtration Performance		
Filtration time	M	11.17
Average filtration flux rate	l/m ² /h	557.9
Average dry solids flux rate	kg/m ² /h	148.122
Cake Drying		
Air dry pressure	kPa	500
Air dry time	M	5
Wet bulk density cake	SG	1.51
Cake moisture	%	16.1
Cake characteristics		
Cake characteristics	Firm/Sticky	Firm breaks up easily after discharge
Cake discharge	Good	slight residue left on cloths

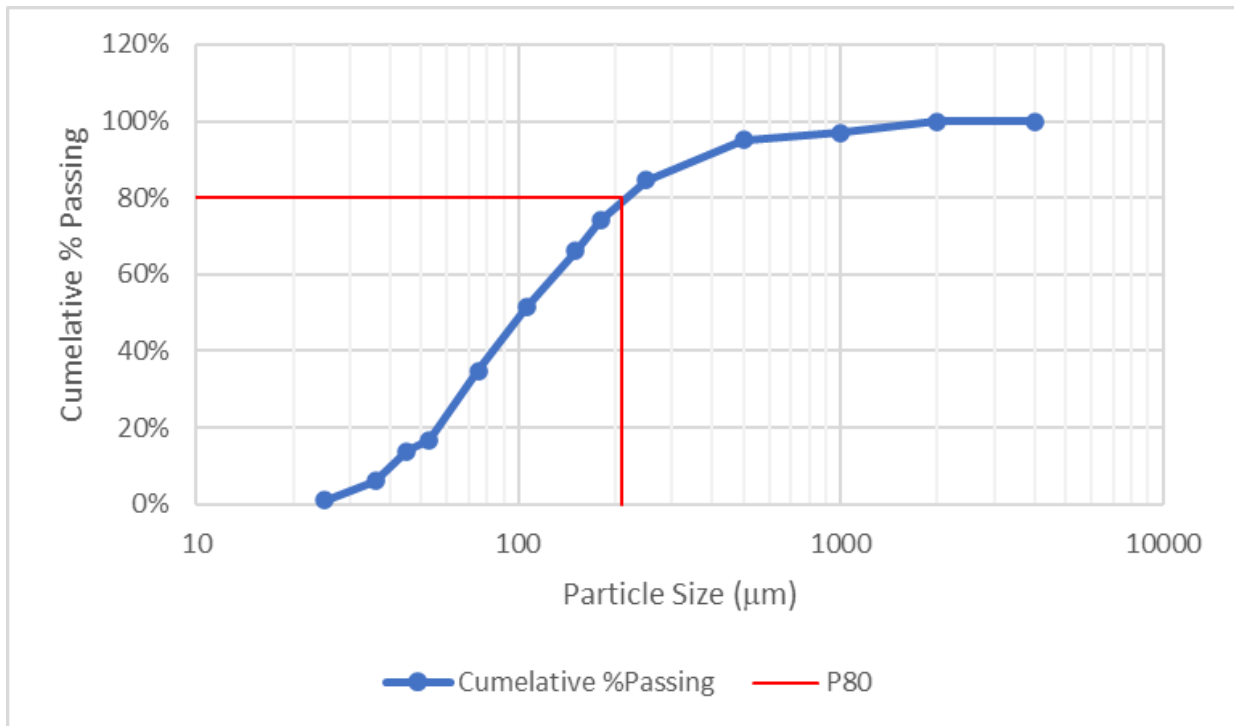


Figure 13-4: Tails Filtration Sample PSD

Particle size analysis of the tailings samples indicated a P80 of 210µm, which is in line with expected tailings particle size distribution from the Okanjande processing plant. The tailings samples were shown to be amenable to plate and frame filtration with no pre-thickening. A slurry of 22% solids content was filtered through the test unit in 11.17 minutes and achieved a final cake moisture of 16.1% solids after a cake blow of 5 minutes. With additional drying on the drying pads, the required 15% moisture content for dry stacking of tailings is achievable.

13.4 SAMPLE REPRESENTATIVITY

Sampling for the SGS flotation test work program was completed by representatives of MSA, Worley, Halyard as well as Imerys.

The Weathered samples were collected as grab samples from crushed mineralized material (-13mm) stockpiles in May 2021. The Fresh samples were collected in July 2021, and was prepared from a blend of 6 different drillholes that intersected the fresh mineralized material zone. (Table 13-22).

Table 13-22: Fresh Core Samples for SGS Flowsheet Confirmation Test program

BHID	SAMPLE ID	FROM (m)	TO (m)	LENGTH (m)	S %	C %
OKJ2	OKJ2-25	55.00	56.00	1.00	2.36	4.20
OKJ3	OKJ2-26	56.00	57.00	1.00	4.57	7.80
OKJ4	OKJ4-33	41.00	42.00	1.00	2.52	3.34
OKJ4	OKJ4-62	69.00	70.00	1.00	3.01	5.53
OKJ4	OKJ4-75	81.00	82.00	1.00	3.02	6.60
OKJ6	OKJ6-25	26.00	27.00	1.00	3.80	3.77
OKJ8	OKJ8-65	69.00	70.00	1.00	2.64	8.08
OKJ9	OKJ9-58	73.00	74.00	1.00	2.37	4.31

Samples for the filtration tests were taken from 2 locations from the existing Okorusu tails flotation facility. Sample points were spaced 60m apart and sampled to a depth of 1m. The top 10cm of material was discarded to prevent skewing of particle size due to wind action. Location of the sample points are indicated in Figure 13-5.



Figure 13-5: Tailings Sample Locations

13.5 MINERAL AND METALLURGICAL TEST WORK CONCLUSIONS & RECCOMENDATIONS

The mineral processing test work program completed at SGS in 2021 indicated that the new flow sheet could improve on the existing flow sheet. The SGS test work outcomes corresponded well with early test work and pilot campaigns conducted in the 1990's.

Based on the latest test work conducted improved recovery and flake distribution was achieved when compared to the test work conducted by MINTEK in the 1990's and 2015.

Based on the completed SGS test work program the design criteria for the flotation plant modifications were developed.

Filtration test work completed by Latham Filtration Technology indicated that a simple plate and frame filtration plant, with no pre-thickening, could be applied to successfully dewater flotation tailings to allow disposal on an integrated waste dump.

Table 13-23: Expected Concentrate Grade and Flake Size Distribution from new process flow sheet

Product	Average		Range	
	%C(t)	% Distribution	%C(t)	% Distribution
Composite				
Composite	96.0	92.0	95.0 – 96.5	91.0 – 93.0
Concentrate Composition				
>300µm	96.0	9	95.0 – 97.0	7 – 11
300 – 180µm	96.0	40	95.0 – 97.0	38 – 42
180 – 106µm	96.0	30	95.0 – 97.0	28 – 32
<106µm	96.0	21	94.0 – 96.0	19 – 23

While the mineral processing test work leading up to this PEA Report supported the expectation for much better performance in terms of graphite flake purity and flake size distribution, it is recommended to conduct further test work to investigate:

- **Concentrate upgrading with gravity separation.** This would be beneficial to possible difficulties with graphite / mica separation by flotation as the only separation method, assuming that contamination is not due to interstitial mica mineralization. If required, the newly redundant Okorusu gravity spirals can be brought back into operation with relative ease and low cost.
- **Concentrate upgrading by chemical means.** As per the early test work discussion a measure of success was achieved with various schemes of roasting followed by acidic leaching. This could potentially be a method to upgrade marginal concentrates or concentrates failing to achieve grade.

14 MINERAL RESOURCE ESTIMATES

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

The Mineral Resource estimate was completed by Ms Ipelo Gasela, Senior Mineral Resource Consultant, who is also the QP for the Mineral Resource. The Mineral Resource estimation was completed using Datamine Studio RM software, augmented by Datamine Supervisor software for statistical and geostatistical analysis. The data used for the Mineral Resource include drilling data collected by RUL, Gecko and Imerys-Gecko. TGC and S grades were estimated in the block model. The S grades were estimated to inform assessments of the impact sulphur may have on the Project.

14.1 MINERAL RESOURCE DATA

MSA received all the drillhole data from Imerys-Gecko in Excel spreadsheets including data pertaining to:

- Collars;
- Sample assays;
- Hole inclination and direction (based on set up orientation);
- Geology logging; and
- Density.

The drilling data are from percussion and diamond drillholes completed at the Okanjande graphite deposit and the surrounding, satellite deposits (Table 14-1). The only data used for Mineral Resource estimation is for the Okanjande Main deposit that includes the North Mineralized Extension and South Mineralized Extension.

A high-level validation was completed that includes the following processes:

- Examining the sample assay, collar survey, downhole survey and geology data to ensure that the data were complete for all of the drillholes;
- Examining the de-surveyed data in three dimensions to check for spatial errors;
- Examination of the assay data in order to ascertain whether they are within expected ranges;
- Checks for “FROM-TO” errors, to ensure that the sample data did not overlap one another or that there were no unexplained gaps between samples;
- Checks for excessive mineralised sample lengths; and
- Checks for unsampled drillholes.

Table 14-1: Summary of the drillhole data (MSA, 2021)

Deposit	Drillhole Series	Campaign	Type	Number of drillholes
Okanjande Main"	OAD03 – OAD13; OAD15 – OAD27; OAP10 – OAP75; OAP81; OAW01 - OAW14	RUL	Percussion and Diamond	105
Northern Mineralized Band	OAP76 – OAP78	RUL	Percussion	4
Okanjande Main (including North Mineralized Extension and Southern Mineralized Extension)	OAD19DT - OAD75D; OKD005 - OKD050; OPH001 - OPH009; SZH002 - SZH004; SZH006	Gecko	Diamond	72
Northern Mineralized Band	OKD051 – OKD063	Gecko	Diamond	13
North-east of Okanjande Main	OKD001 – OKD004	Gecko	Diamond	4
Okanjande Main (including North Mineralized Extension and Southern Mineralized Extension)	OKJ1 – OKJ12; OkaPBH_020 – OkaPBH_026	Imerys-Gecko	Diamond	19
Northern Mineralized Band	OkaPBH_027 – OkaPBH_034	Imerys-Gecko	Diamond	6
SW Target	OkaPBH_013 – OkaPBH_019	Imerys-Gecko	Diamond	7
Highlands Target	OkaPBH_009 – OkaPBH_012	Imerys-Gecko	Diamond	4
Old Mine	OkaPBH_007 – OkaPBH_008	Imerys-Gecko	Diamond	2
Leopardskloof Target	OkaPBH_001, OkaPBH_002, OkaPBH_005	Imerys-Gecko	Diamond	3
Roobult Target	OkaPBH_035 – OkaPBH_037	Imerys-Gecko	Diamond	3

The results of the validation and corrections made to the data are as follows:

- Geology logging for the Imerys-Gecko holes was not captured electronically. MSA captured weathering logging where necessary to facilitate modelling;
- Most of the collar positions plot between -2m and +1.5m relative to the topography. Only two collars are located below -2m, to -2.8m, and another two collars above +1.5m, to +3.6m, relative to the topography. In addition, numerous collars in the pit area are in excess of 2m above the topography. The collars were accepted for the Mineral Resource estimation;
- 28 drillholes located on the peripheries of the deposit were not sampled and assayed. Though these were identified to be mineralized, the thickness and magnitude of the mineralization in these drillholes is not known, therefore the areas in which they are drilled were excluded from the Mineral Resource;
- A total of 78 gaps were observed in the Gecko and Imerys-Gecko data due to un-mineralised intervals that were not sampled. The gaps are between 0.07m and 28.5m in length;
- Sample lengths are from 0.15m to 4m, with only six samples that are longer than 2m;
- The TGC assays are from 0% to 16.6%, and S assays are from 0.01% to 16.38%. These assay ranges are expected for the deposit and no adjustments were made to the data;

- A total of 144 density measurements are available, which are between 2.14 t/m³ and 2.91 t/m³, which is considered an acceptable range for the deposit;
- No downhole surveys were completed. Drillholes are up to 98m long at Okanjande and no major deviations from the collared orientation are expected; and
- A total of 12 overlaps were observed in the lithology file. These are not considered to be material and the overlapping intervals were removed from the Mineral Resource estimation database.

The final dataset used to estimate the Mineral Resource comprises 196 drillholes.

14.2 EXPLORATORY DATA ANALYSIS

The data for the Okanjande Main deposit includes drilling data collected by RUL, Gecko and Imerys-Gecko. The drillholes were drilled on northwest oriented sections. Generally, the drillholes are spaced 50m apart with drill lines at the central area of the deposit being 25m apart. Drillhole spacing widens towards the edges up to approximately 100m (Figure 14-1). The data comprises 81 percussion holes and 115 diamond holes. 23 of the diamond drillholes are from twin drilling, 21 of them twinning percussion holes and two are twin diamond drillholes. Holes were drilled either vertically or inclined, with most of the inclined holes dipping at 60° at an azimuth of 320° to intersect the south-east dipping mineralization at a near to perpendicular angle. The maximum drilling depth is to 90m below surface.

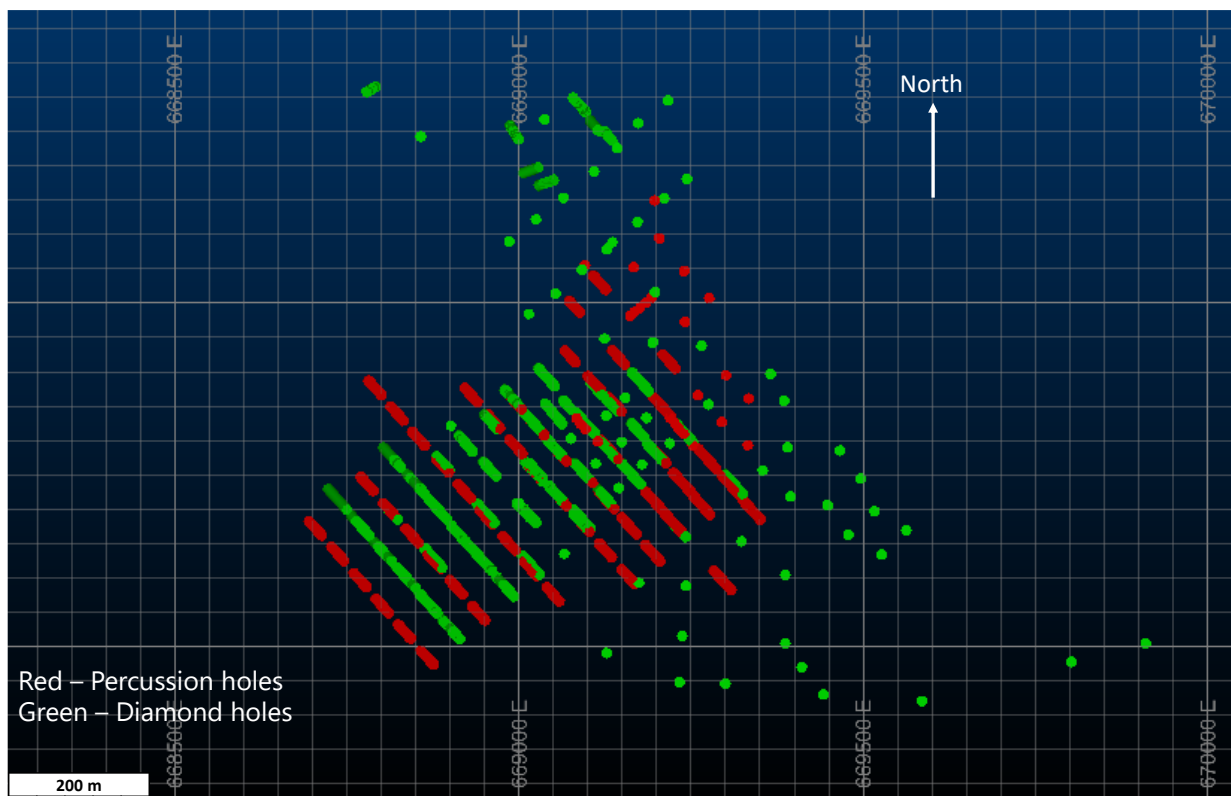


Figure 14-1: Plan view of drillhole traces, coloured by drilling type (MSA, 2021)

The TGC grade distribution is mixed due to the interbedded mineralised and non-mineralised layers in the graphitic gneiss/quartzite (Figure 14-2). The sulphur grade histogram shows a bimodal distribution (Figure 14-2). This is related to weathering, where the lower sulphur grades represent the weathered material and the higher sulphur grades are within the fresh rock.

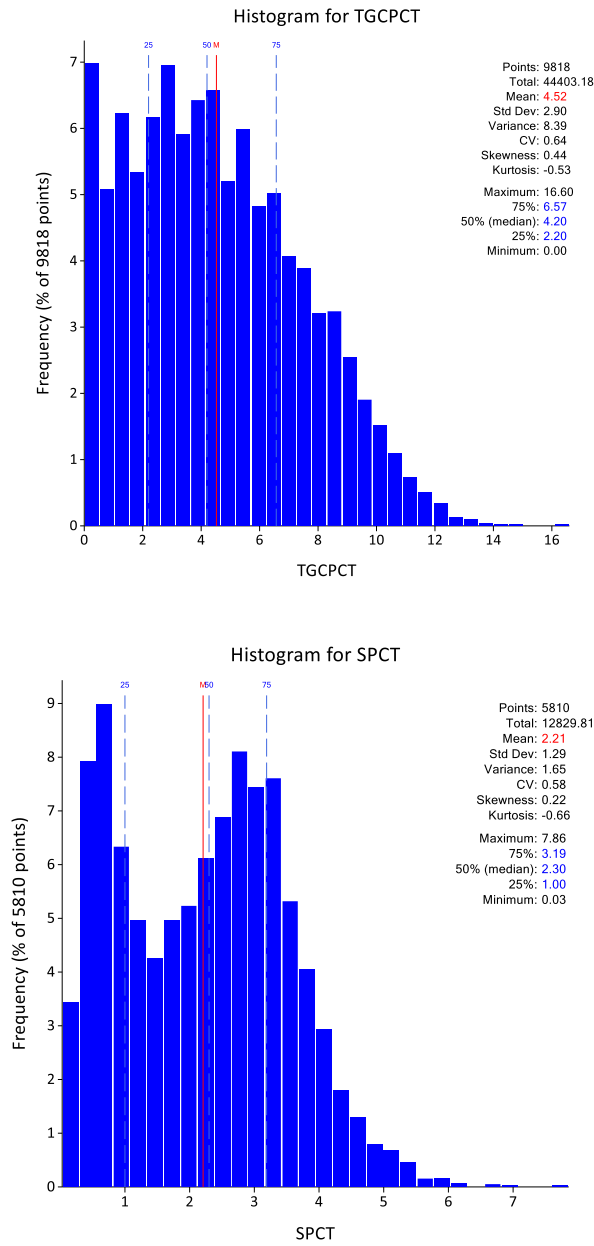


Figure 14-2: Histogram of un-composited lengths for TGC (top) and S (bottom)

14.3 BIVARIATE ANALYSIS

The bivariate plot between TGC and S assays shows two groups of weakly positive correlations; one for weathered and the other for fresh rock (Figure 14-3).

Intersections with no assays were assumed to be un-mineralized and therefore a default value of 0% was applied to the TGC grades for unsampled core. Intervals with TGC grade but missing S grade were left absent for S grade. Based on the correlation between the two variables in the fresh rock, a default S grade of 0.5% was applied to core that was not sampled.

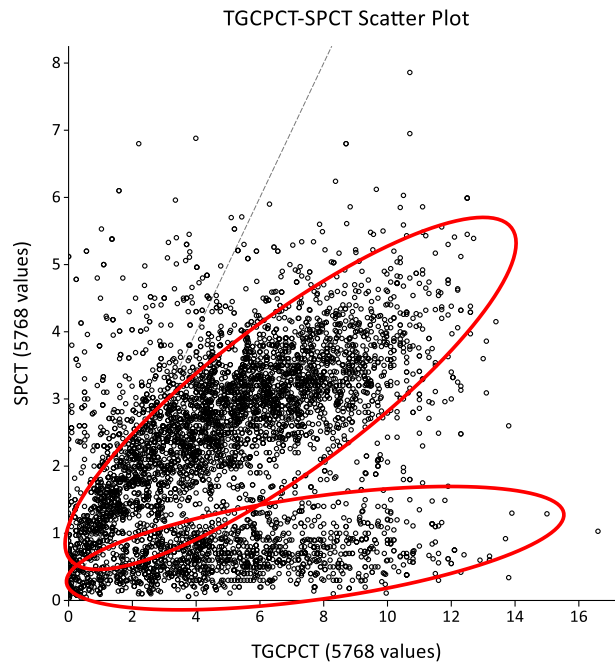


Figure 14-3: Bivariate plot between TGC and S (MSA, 2021)

14.4 GEOLOGICAL MODELLING

A topographic surface, completed in September 2018, included the profiles of the stockpiles and pits. The stockpiles are located outside the Mineral Resource area. The topographic surface was used to limit the vertical extent of the block model and depleted the mined-out areas.

The graphite at Okanjande Main is mainly contained within graphitic gneisses and quartzite while the un-mineralised rocks are pegmatite veins or interbedded un-mineralized quartzite. The logging indicates small scale local geological complexity that could not be modelled at a deposit scale. The deposit was modelled as a simple volume with the lateral extent constrained by un-mineralised drillholes. The lateral extent was modelled halfway between mineralised and un-mineralised drillholes or extrapolated up to a 100m from a mineralised, sampled and assayed drillhole (Figure 14-4).

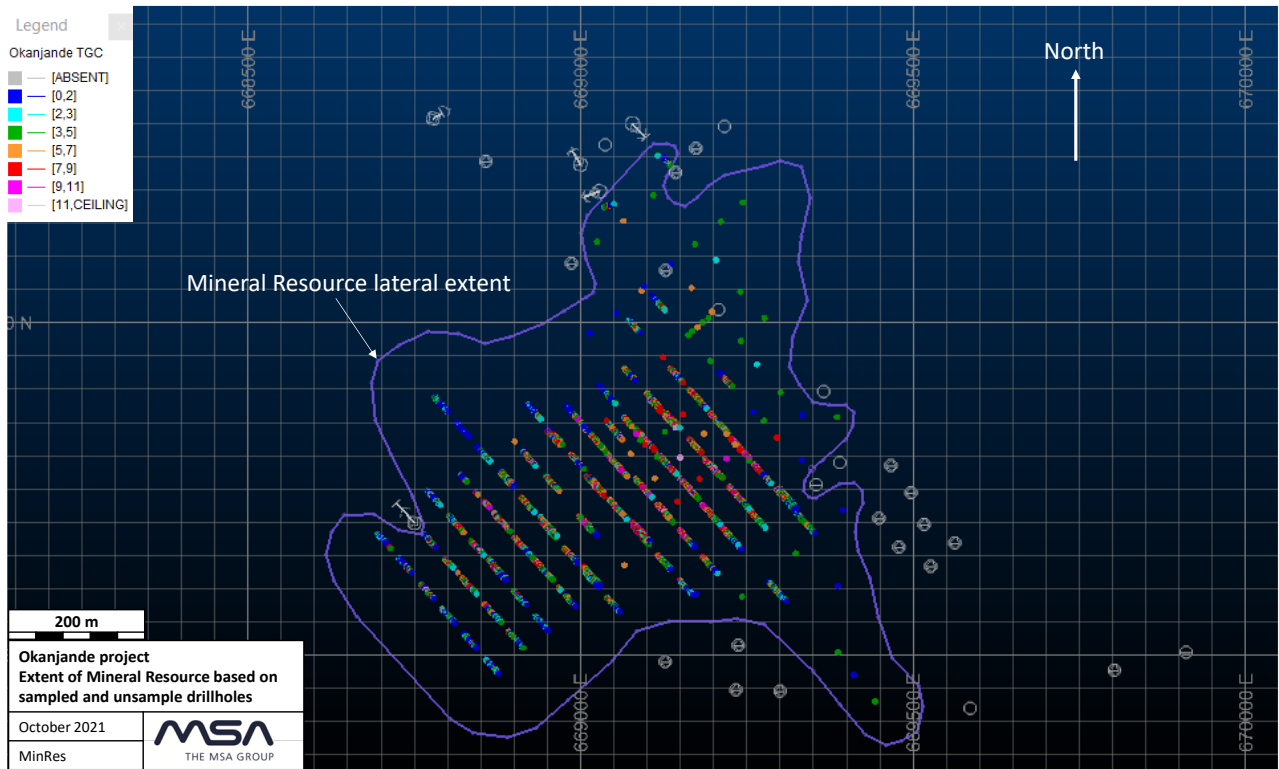


Figure 14-4: Plan view of Mineral Resource extent (MSA, 2021)

The sulphur distribution shows inflection points on the log probability plot at approximately 1% and 2%, which correspond with the base of the weathered and transitional zones respectively (Figure 14-5). However, numerous intersections with sulphur grades less than the 2% S threshold were logged as fresh rock, especially towards the west. The weathered and transitional surfaces were modelled using weathering logging, rather than using S assays. The weathered zone comprises zones of logged weathering intensity of 2 or more and the base of the transitional zone correspond to an interface between a weathering intensity of 1 and 0.

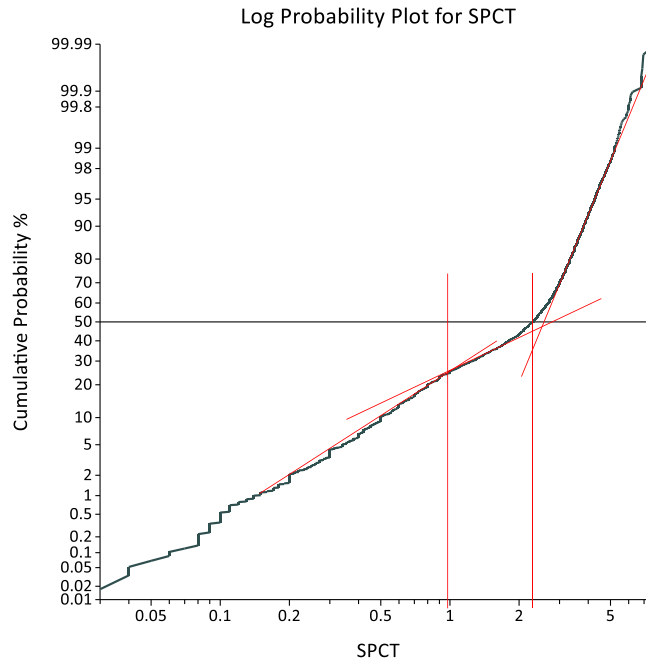


Figure 14-5: Log probability plot for sulphur grade (MSA, 2021)

Generally, the weathered zone is between 15m and 25m below surface. The transitional zone is between 1m and 5m thick, although it can reach thicknesses of 15m locally in the east.

14.5 ESTIMATION DOMAINS

Weathering domains form the basis for sulphur grade estimation; therefore, three estimation domains were used for the sulphur grade estimation (Table 14-2).

Table 14-2: Sulphur estimation domains (MSA, 2021)

Weathering Domain	Explanation
1	Weathered
2	Transitional
3	Fresh

The TGC grades have a similar grade distribution for both the weathered and transitional domains. The TGC grades in the fresh domain are higher than in the weathered zones (Figure 14-6). The higher average grade in the fresh domain is because of a high-grade zone at depth. Visual assessment of grades in the contact zone revealed that there are no grade changes attributed to the change of weathering state and that TGC can appropriately be estimated as one domain.

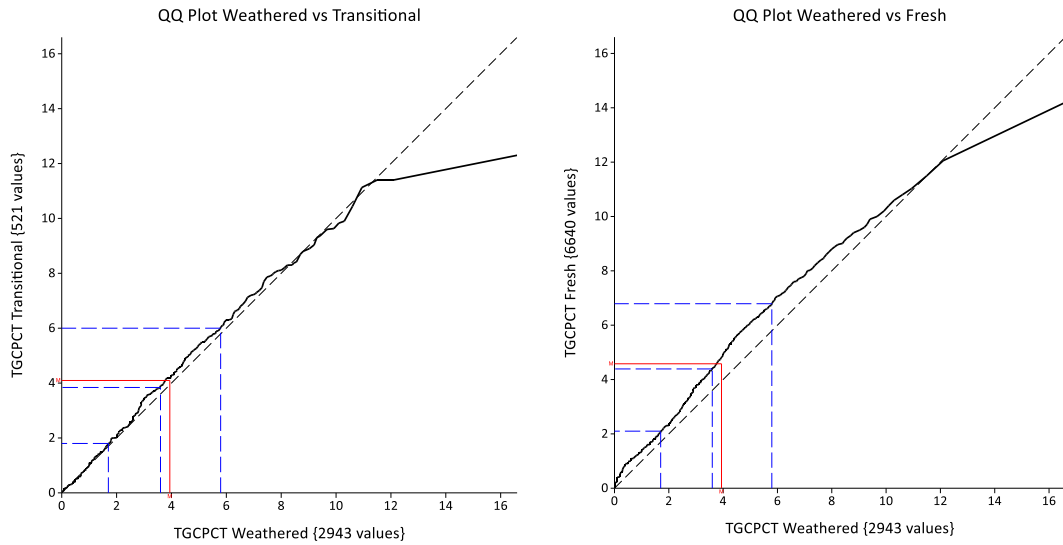


Figure 14-6: QQ plot comparing TGC grades in different weathering domains (MSA, 2021)

14.6 COMPOSITE DATA

The drillhole samples were composited to 1m because this is the most occurring sample length accounting for the majority of the data (Figure 14-7). The compositing method allowed for slight adjustments in the composite length in order to avoid discarding any sample where the sampled portion of the drillhole is not an exact multiple of the composite length.

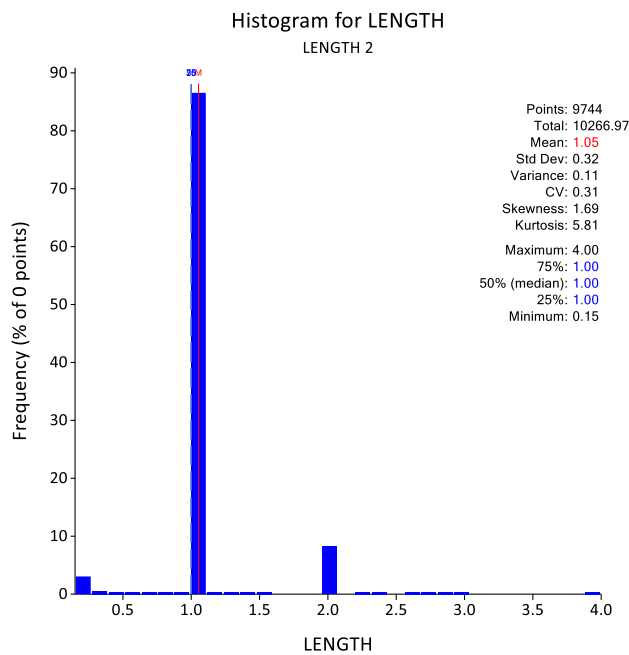


Figure 14-7: Un-composited sample length histogram (MSA, 2021)

The TGC sample grade population shows a mixed distribution, with a high frequency of composite samples with a grade of 0% TGC due to the zero values applied where samples were not taken (Figure 14-8). The mixed distribution is due to the deposit being made up of alternating mineralised and un-mineralised layers, which could not be modelled as separate domains. The deposit also has a high-grade core, which transitions into lower grades towards the peripheries.

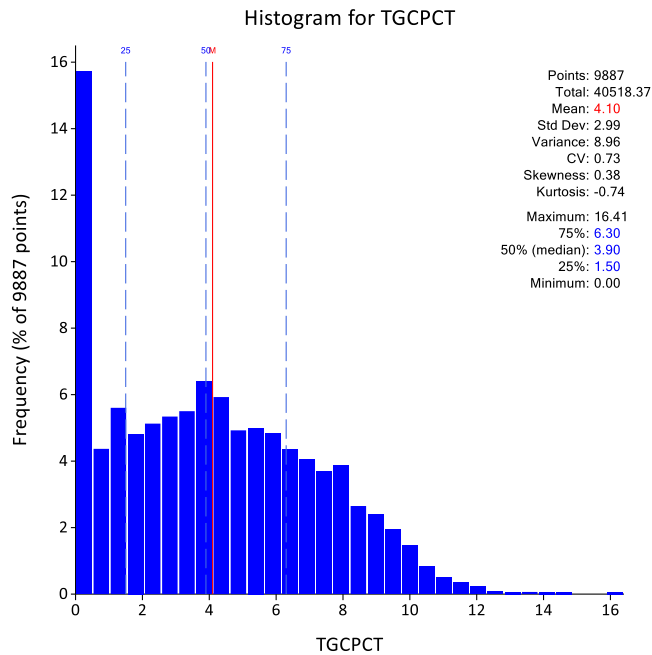
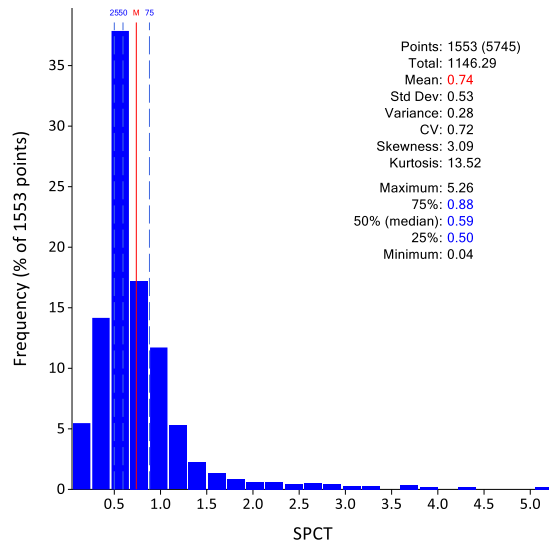


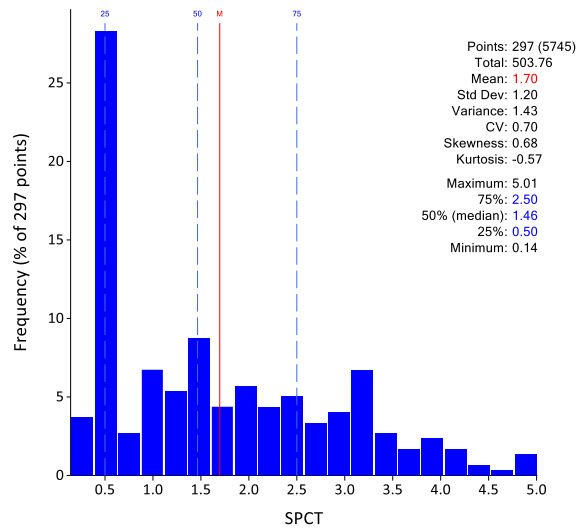
Figure 14-8: Composite TGC grade histogram (MSA, 2021)

The sulphur data was assigned domains based on weathering. The sulphur grades show a progressive increase from weathered through transitional to fresh. The weathered domain has a positively skewed distribution, the transitional has a mixed distribution and the fresh domain distribution tends towards a normal distribution. All the domains have a high frequency of 0.5% S values (Figure 14-9), which was the default value applied where there were no samples collected.

Histogram for SPCT
1



Histogram for SPCT
2



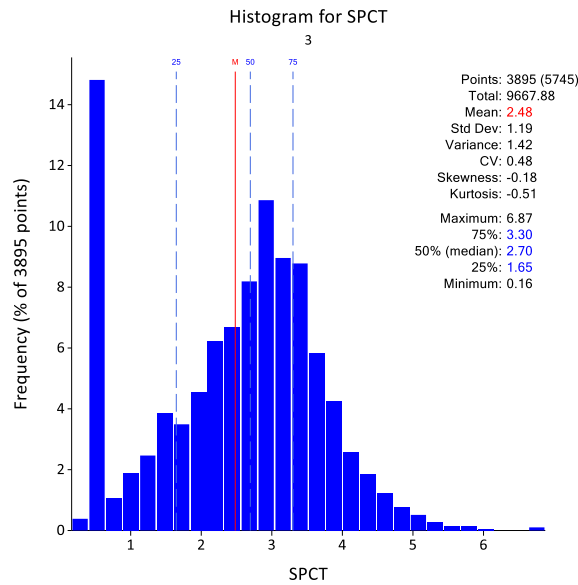


Figure 14-9: Composite sulphur grade histograms per weathering domains (MSA,2021)

14.7 GEOSTATISTICAL ANALYSIS

A single variogram was modelled for TGC and two variograms were modelled for S grade; one for weathered and the other for the fresh domain. A variogram for the transitional domain could not be modelled reliably and the variogram for the fresh domain was used in estimating the transitional domain.

The experimental variograms for sulphur were calculated from composite data without default values since the high frequency of the default values artificially changes the spatial variance.

The variograms were modelled at a lag distance of 50m, which is the general drillhole spacing. The nugget effect of the variograms was established using the downhole variograms. The variograms were modelled with two spherical structures and the sill variance was scaled to 1.

The TGC variogram and S variogram for the fresh domain were orientated in line with the interlayering observed in the granitic gneiss at an azimuth of 50° and a dip of 20° to the southeast. The S variogram model for the weathered domain was orientated horizontally with the east-west major direction being based on the variogram fan. The S data were transformed to normal scores for the weathered domain because the grade distribution is positively skewed. The final parameters for the variograms were back-transformed (Table 14-3).

The variograms demonstrate strong continuity in excess of the drillhole spacing.

Table 14-3: TGC and S variogram parameters (MSA, 2021)

Variable	Domain	*Rotation angles			Nugget Effect	Sill 1	Range 1 (m)			Sill 2	Range 2 (m)		
		X	Y	Z			X	Y	Z		X	Y	Z
TGC	-	140	20	180	0.11	0.56	40	40	5	0.33	230	195	25
Sulphur	Weathered	0	0	0	0.13	0.63	80	100	6	0.24	525	220	13
	Fresh	140	20	180	0.10	0.63	50	30	7	0.27	440	195	60

Note: *Rotation angles were applied to XZX.
Ranges are rounded off to the nearest 5 m.

14.8 BLOCK MODELLING

A kriging neighbourhood analysis was completed using the TGC variogram to optimise the block size as well as other kriging parameters.

A block model was created with a block size of 20mX by 20mY by 5mZ, which is two fifths of the average drillhole spacing. The block model was sub-celled to the minimum dimensions of 5mX by 5mY by 2.5mZ. The block model was coded by weathering.

14.9 ESTIMATION

The TGC and S grades were estimated into the block model using ordinary kriging into parent cells. The TGC grades and the sulphur grades in the transitional and fresh domains were estimated in line with the interlayering of the granitic gneiss at an azimuth of 050° and dip of 20° to the southeast. Sulphur grade in the weathered zone was estimated using dynamic anisotropy that aligns the estimates to the weathering surfaces, which are sub-horizontal.

TGC grades were estimated using a search ellipse of 100m by 80m by 20m. The transitional and fresh S grades were estimated using the same search ellipse and the weathered S grades were estimated using a search ellipse of 100m by 50m by 5m in line with the variogram range proportions. The estimates were completed using a three-search strategy. The second search distances were set at one and a half times the first search distances and the third search distances were ten times the first search distances. This is to ensure that all block model cells were filled with estimates. The grades were estimated with a minimum number of 12 composites and a maximum number of 18 composites.

14.10 DENSITY

A total of 144 density measurements were completed in October 2021 at the Okanjande deposit using the remaining core from previous drilling. The data were collected from 11 holes located in the central and southern parts of the deposit (Figure 14-10).

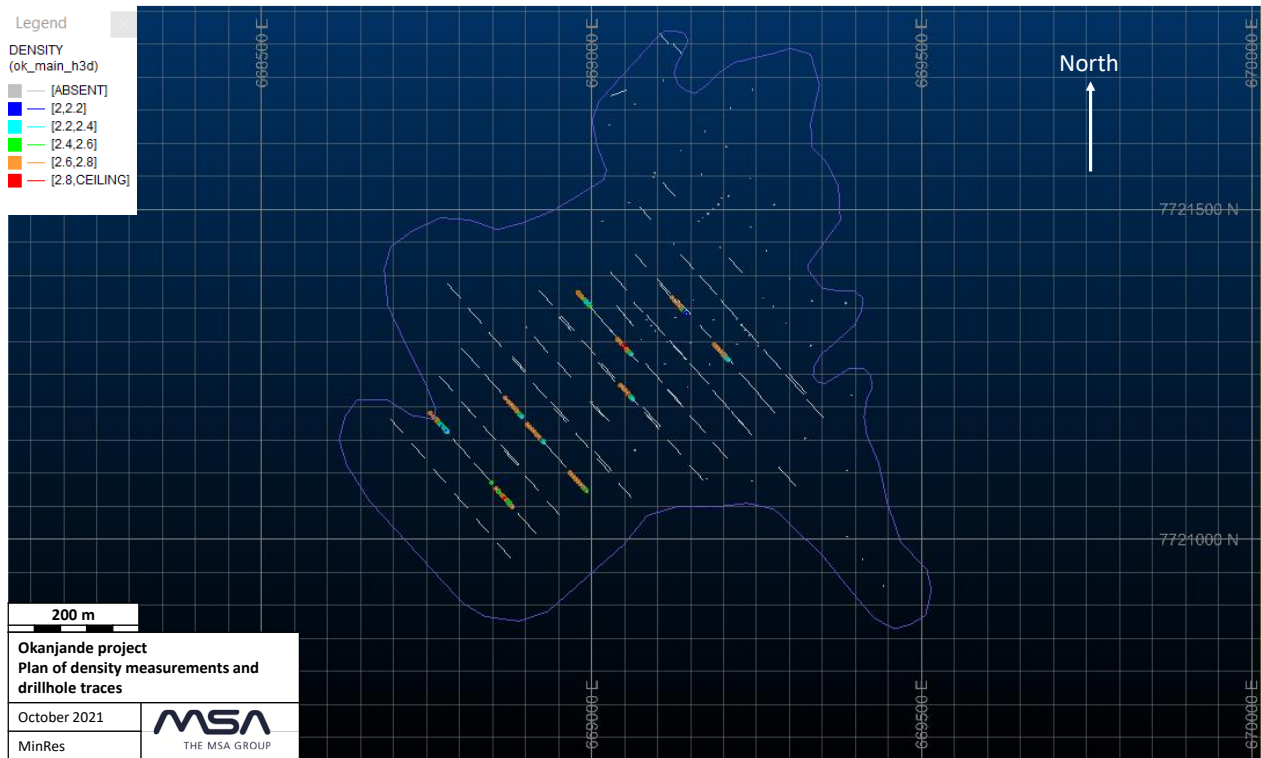
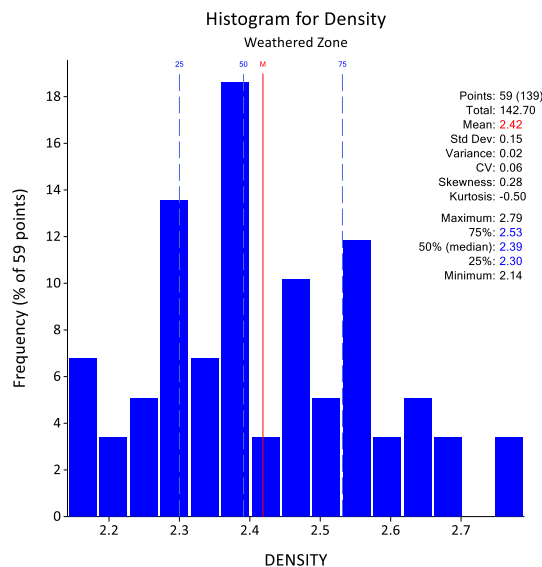


Figure 14-10: Location of drillholes selected for density measurements (MSA, 2021)

The density distributions for the weathered and fresh domains tend towards a normal distribution. Only four measurements were taken in the transitional domain due to the thin nature of this zone and difficulty in finding suitable samples.

Average density values were applied to the block model according to weathering. The amount of density measurements and their low variability were sufficient to derive a reliable average density value for the weathered and fresh domains (Figure 14-11). The mean of the transitional zone density measurements was considered acceptable as it falls between the average densities of the weathered and fresh domains.



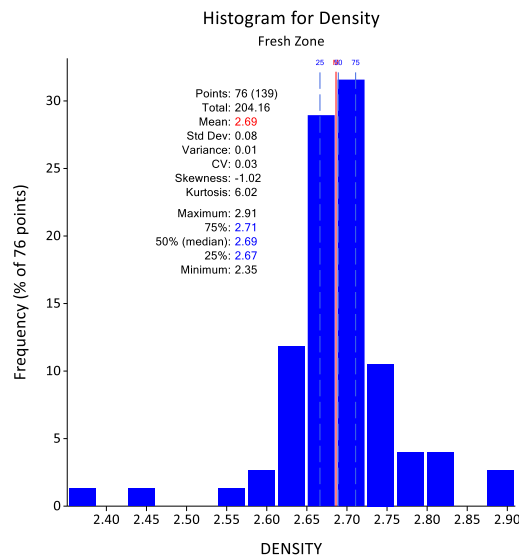
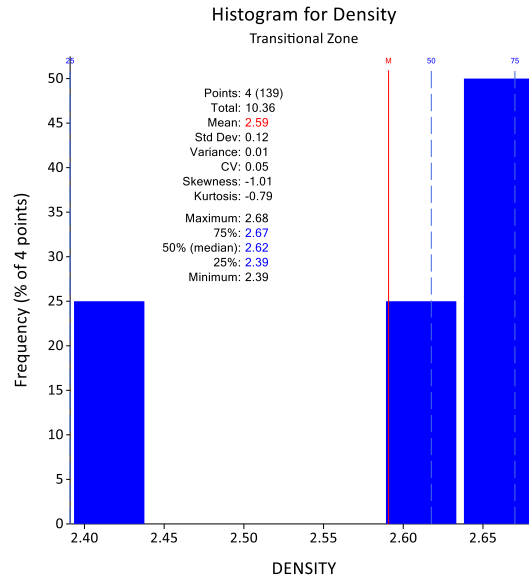


Figure 14-11: Density histograms per weathering domain (MSA, 2021)

14.11 VALIDATION OF THE ESTIMATES

The grade estimates were validated by:

- Visual inspection;
- Global mean comparison; and
- Swath plots.

Visual inspection of the block model shows that the estimated grades are representative of the input composite grades. The TGC and S grade trends in the transitional and the fresh domains follow the orientation of the graphitic gneiss interlayering, while the sulphur grade trend in the weathered domain is sub-horizontal following the orientation of the weathering surface (Figure 14-12 to Figure 14-15 (TGC) and Figure 14-16 and Figure 14-19 (S)).

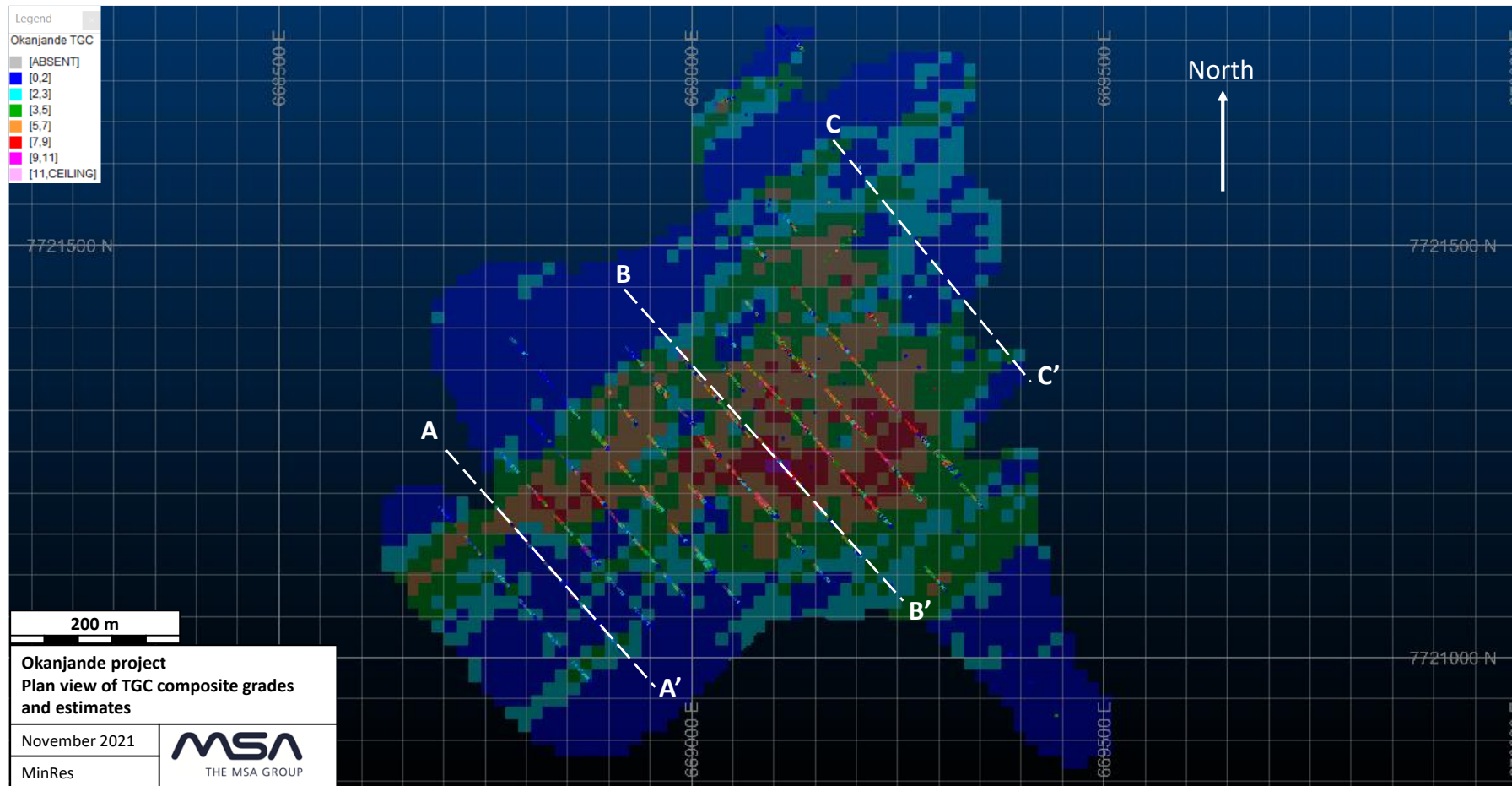


Figure 14-12: Plan view showing block model grades and composite drillhole sample grades for TGC, and section lines for the sections (MSA, 2021)

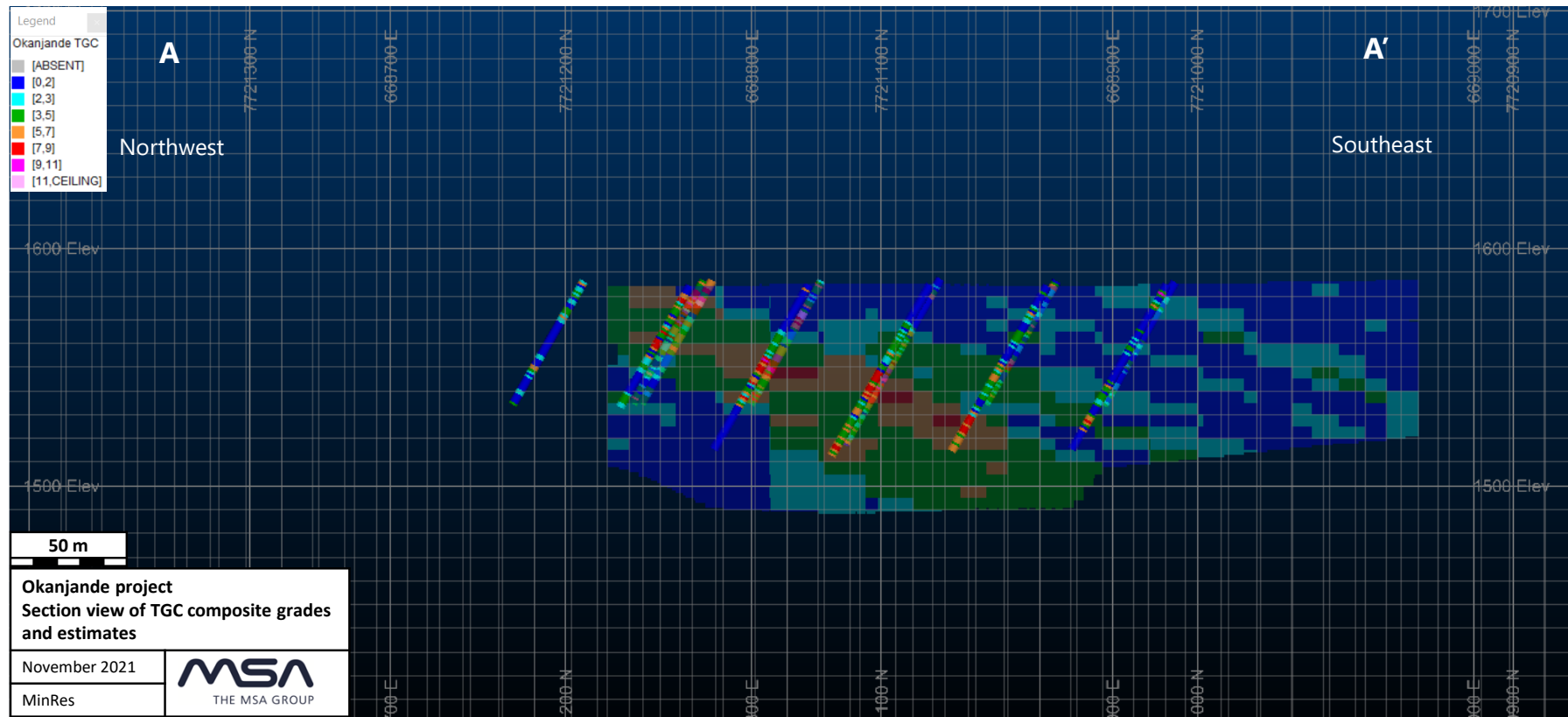


Figure 14-13: Section showing block model grades and composite drillhole sample grades for TGC for section line A – A' (MSA, 2021)

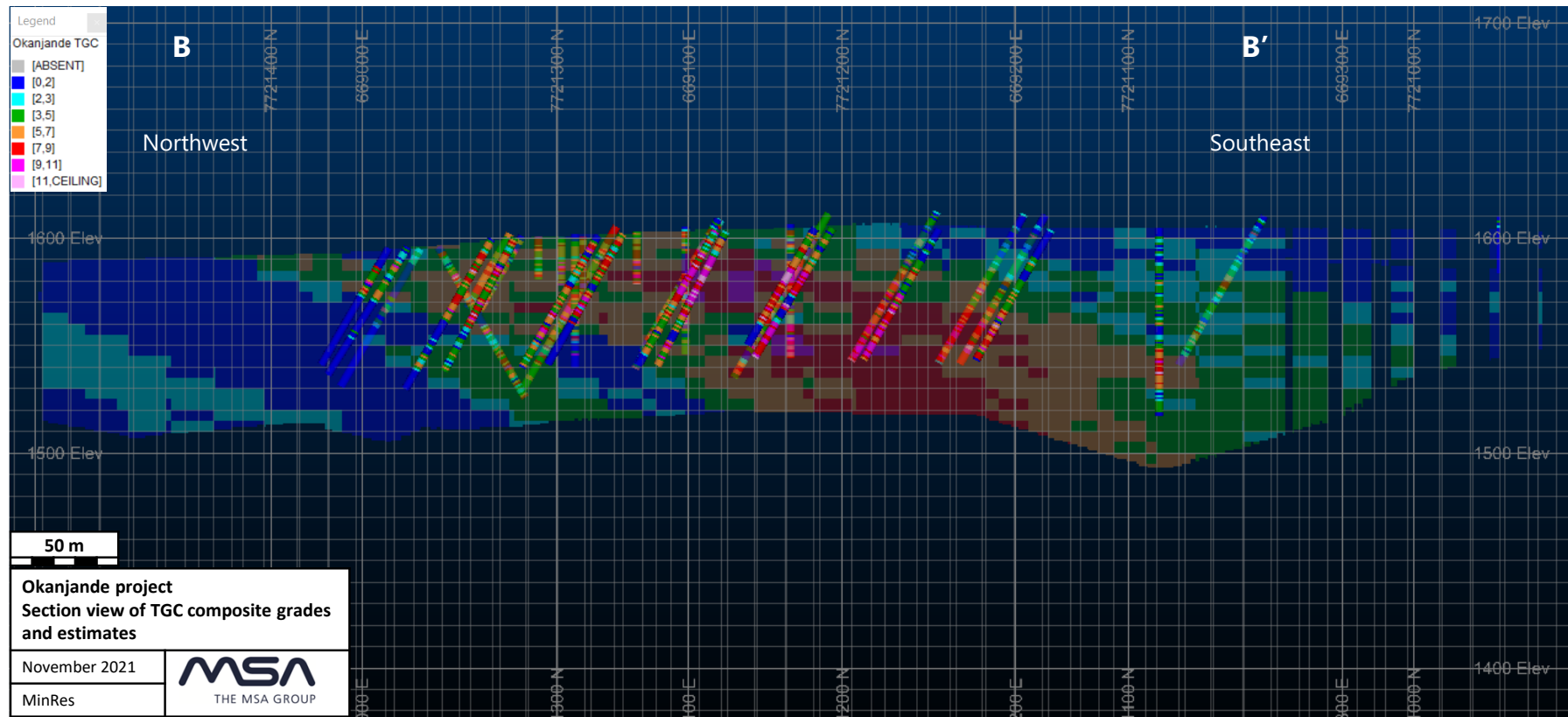


Figure 14-14: Section showing block model grades and composite drillhole sample grades for TGC for section line B – B' (MSA, 2021)

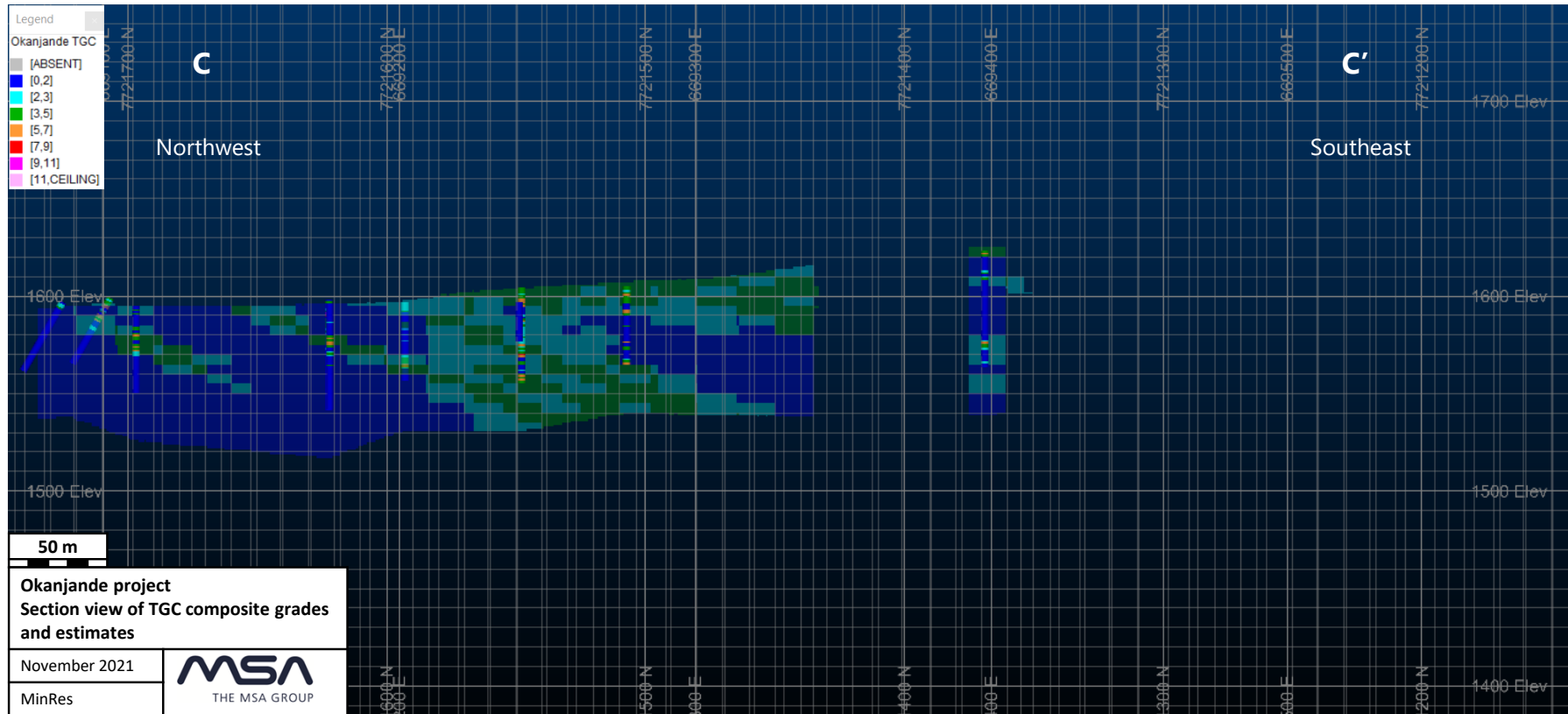


Figure 14-15: Section showing block model grades and composite drillhole sample grades for TGC for section line C- C' (MSA, 2021)



Figure 14-16: Plan view showing block model grades and composite drillhole sample grades for S (MSA, 2021)

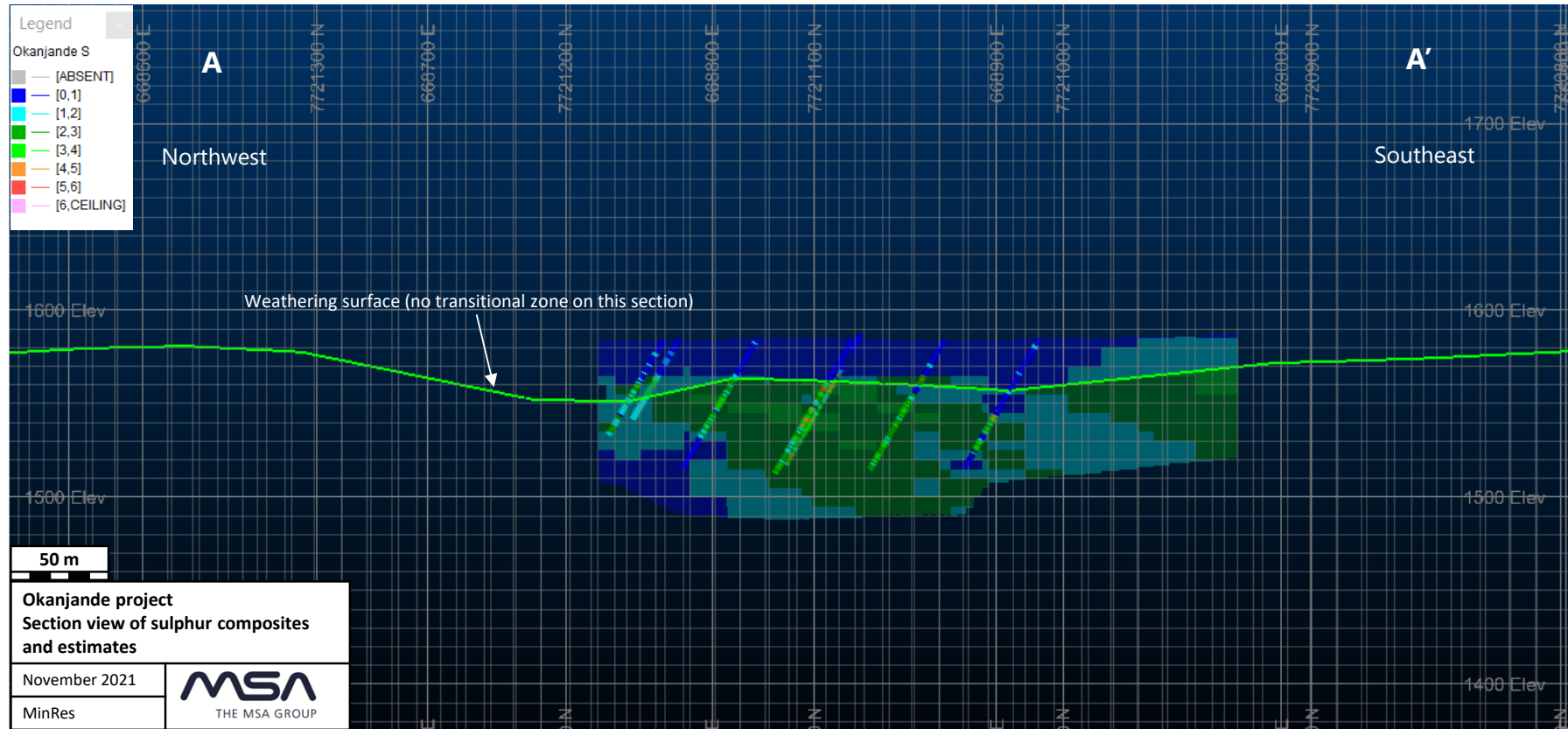


Figure 14-17: Section showing block model grades and composite drillhole sample grades for S for section line A – A' (MSA, 2021)

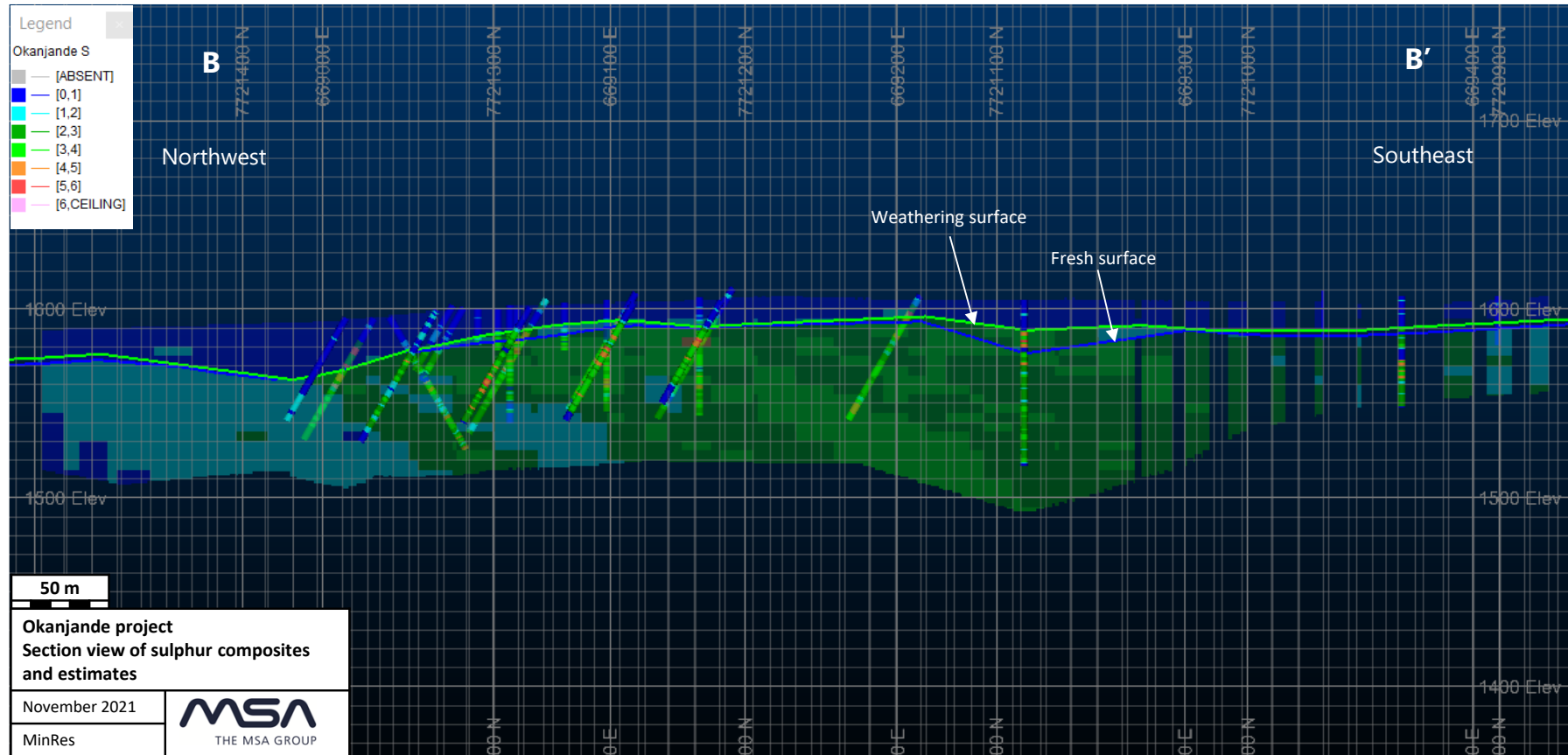


Figure 14-18: Section showing block model grades and composite drillhole sample grades for S for section line B – B' (MSA, 2021)

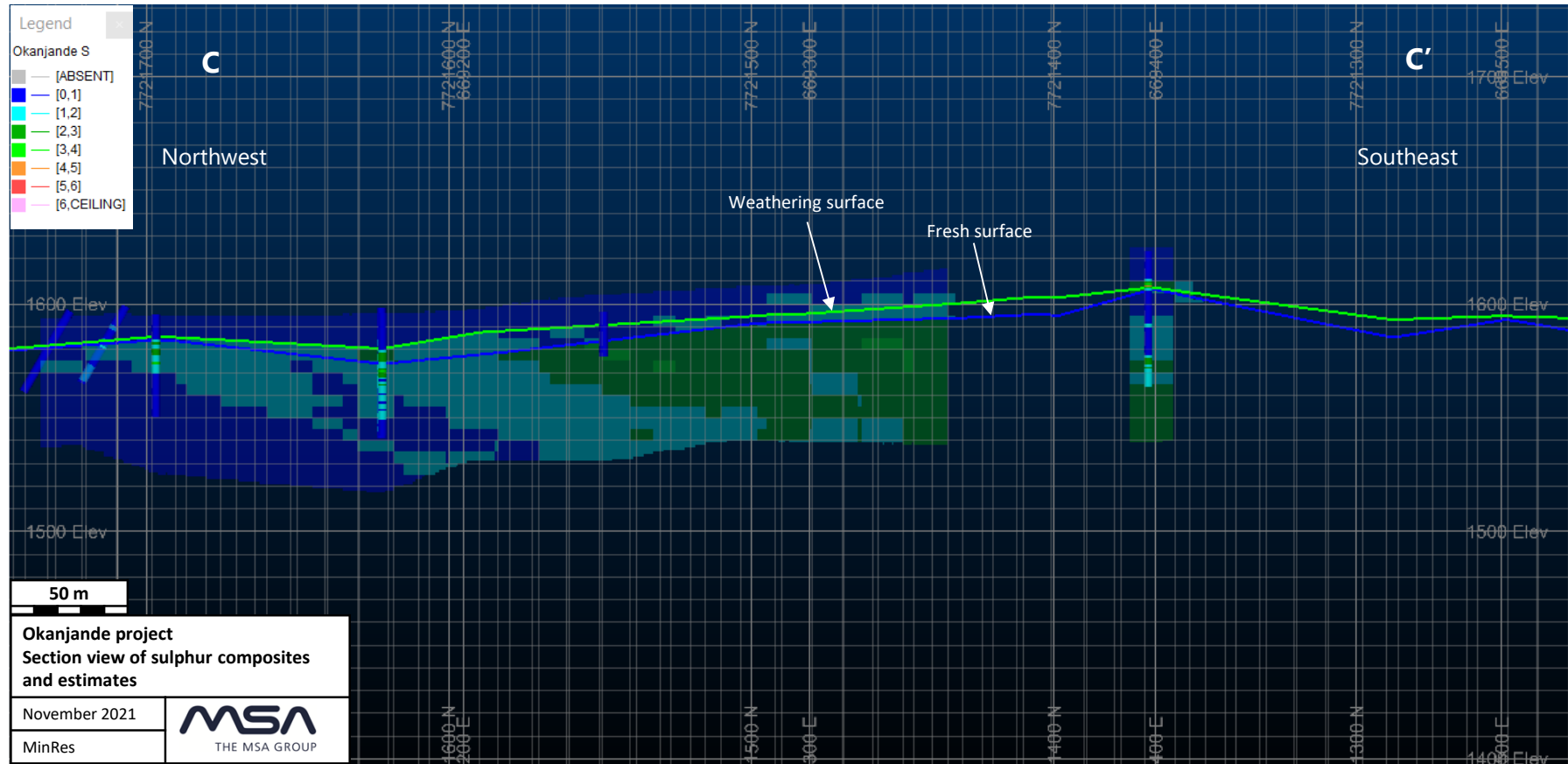


Figure 14-19: Section showing block model grades and composite drillhole sample grades for S for section line C – C' (MSA, 2021)

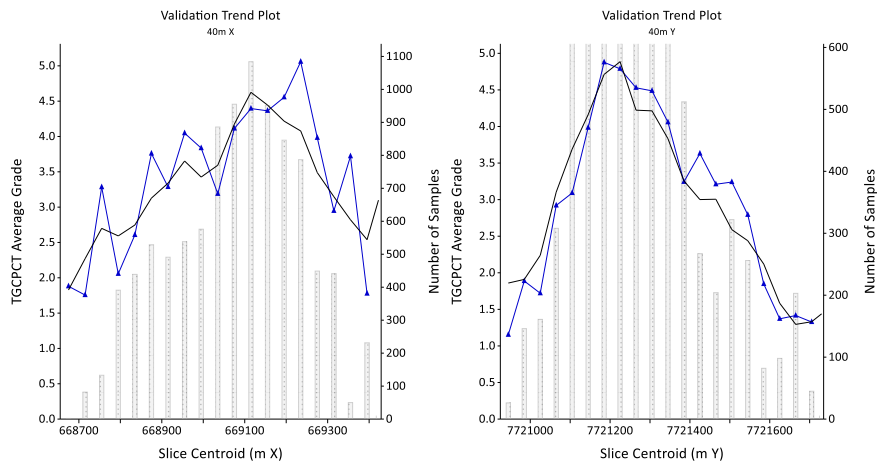
a

The composite grades were declustered to 50mX by 50mY by 5mZ and the global mean declustered grades of the TGC and sulphur composites were compared to the block model estimated average grades. The relative differences are less than 10% for all domains except sulphur estimates in the transitional zone (Table 14-4). The validation demonstrated that the estimated grades are globally representative of the composite data and the discrepancy in the transitional zone sulphur can be explained by wide data spacing in localised higher grade areas.

Table 14-4: Global mean grade comparisons

Variable	Weathering domain	Declustered mean (%)	Block model mean (%)	Percentage difference
TGC	N/A	3.65	3.61	-1%
S	Weathered	0.71	0.77	8%
	Transitional	1.59	1.87	18%
	Fresh	2.31	2.35	2%

Swath plots demonstrate that the TGC and S estimates follow the trend of the input data, with some level of smoothing that is expected through estimation. This shows that the estimated grades are locally representative of the input composite grades (Figure 14-20).



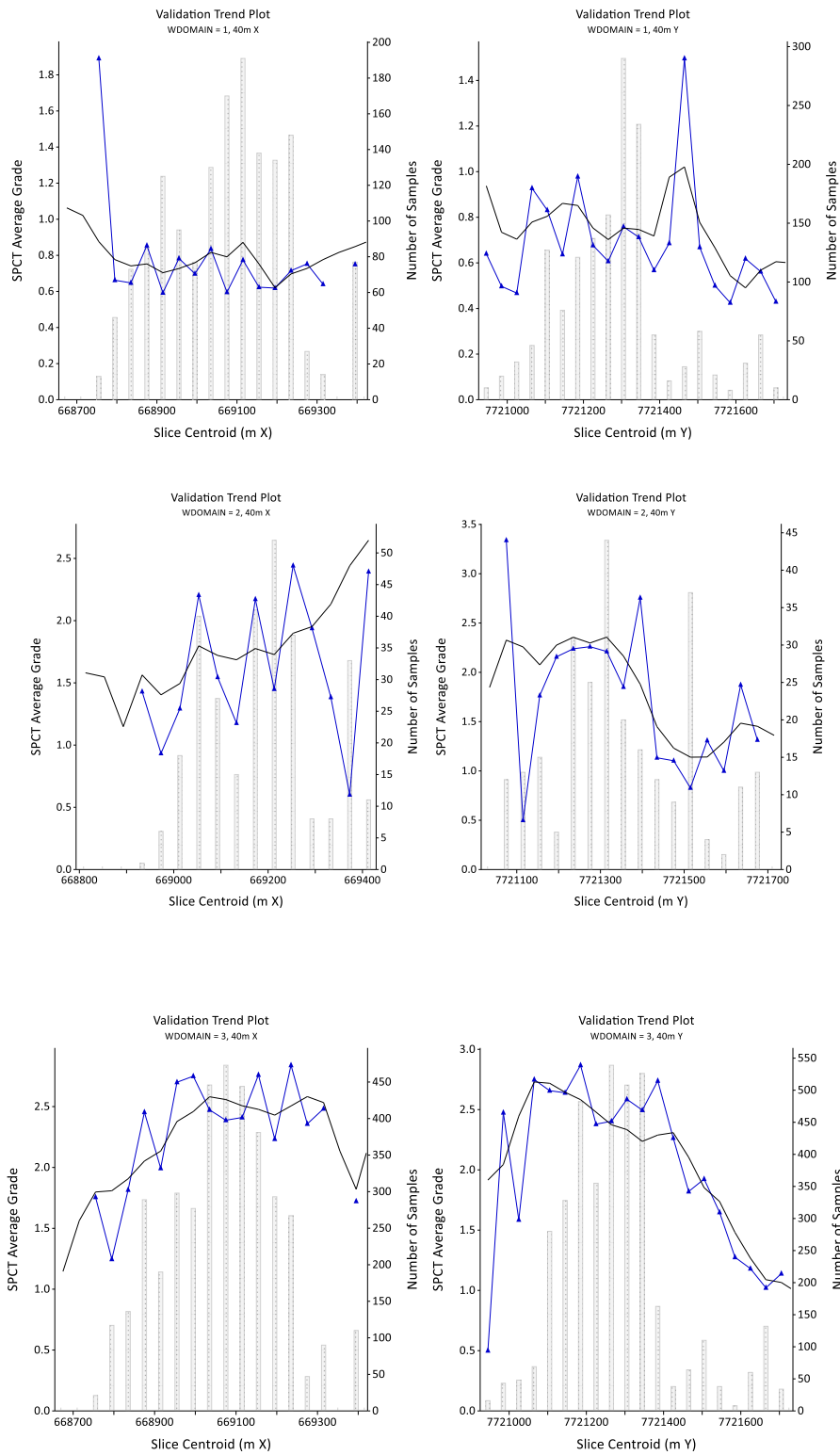


Figure 14-20: Swath Plots for TGC and S estimates (MSA, 2021)

Note: blue line = declustered data; black line = estimates

14.12 CLASSIFICATION

The estimates were classified into Measured, Indicated and Inferred Mineral Resource categories as defined by the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014). The following factors were considered during the Mineral Resource classification:

- the reliability of the drilling data;
- confidence of the geological model, geological and structural interpretation; and
- grade continuity.

The historical database has been verified through examination of drillhole logs and assay certificates. Cores were randomly selected on site and verified against drillhole logs. QA/QC analyses of the data show adequate accuracy and precision in the Gecko data, while the Imerys-Gecko assay data were assessed with field duplicates only. None of the RUL duplicate data are available for assessment. Bias testing between the different drilling campaigns shows similar TGC grade distributions.

The collar positions could not be verified by the Geology QP because the areas were inaccessible or covered with thick vegetation. The QP is satisfied that the signed off reports by the surveyors who completed the collar surveys provide for a reasonable audit trail.

The depth limit of the mineralization is not clear and appears to be open at depth. Un-sampled drillholes have defined the lateral extents over much of the area in the north. The extents were limited to 100 m extrapolation from the nearest sampled and assayed mineralised drillhole intersection for the rest of the Mineral Resource.

The Mineral Resource is classified on the basis of TGC estimates as the main variable of economic interest.

- Modelled mineralisation covered by a 50m by 50m grid, that includes the highest confidence Gecko drilling, was classified as Measured Mineral Resources. The Measured Mineral Resource was extrapolated by 25m laterally and by 5m from the base of drillholes at depth;
- Areas drilled at a grid of 50m to 100m spacing were classified as Indicated Mineral Resources. The Indicated Mineral Resource was extrapolated by 50m laterally and by 10m from the base of drillholes at depth; and
- Areas covered by drilling at wider than 100m spacing, or in the north-west where the area appears to be more geologically complex, were classified as Inferred Mineral Resources. Inferred Mineral Resources were extrapolated to the lateral extent of the Mineral Resource model and by 25m vertically from the base of drillholes at depth.

A plan of the classified Mineral Resource is shown in Figure 14-21.

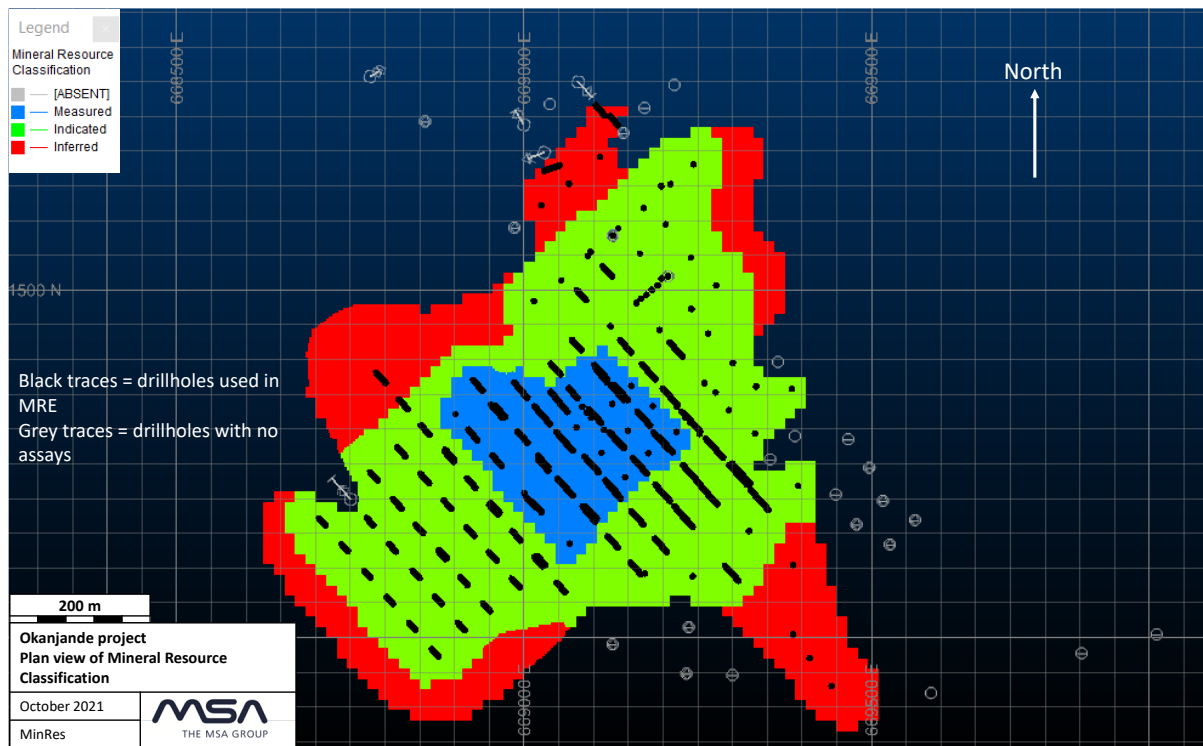


Figure 14-21: Mineral Resource Classification (MSA, 2021)

14.13 RECONCILIATION

Mining at Okanjande took place from early 2017 to October 2018. As part of grade control during mining, blast holes were drilled vertically at a spacing of 3m. The blast holes were composited according to pre-determined blocks of approximately 15m by 15m prior to assaying. The blast holes are 5 m long on average. MSA completed a reconciliation against the Mineral Resource model in the same areas that the blast holes were drilled (Figure 14-22).

On average, the blast holes and Mineral Resource model TGC grades compare well, with a relative difference of -4%. The trend of the blast hole TGC grades conform to the trend of the Mineral Resource TGC estimates (Figure 14-22).

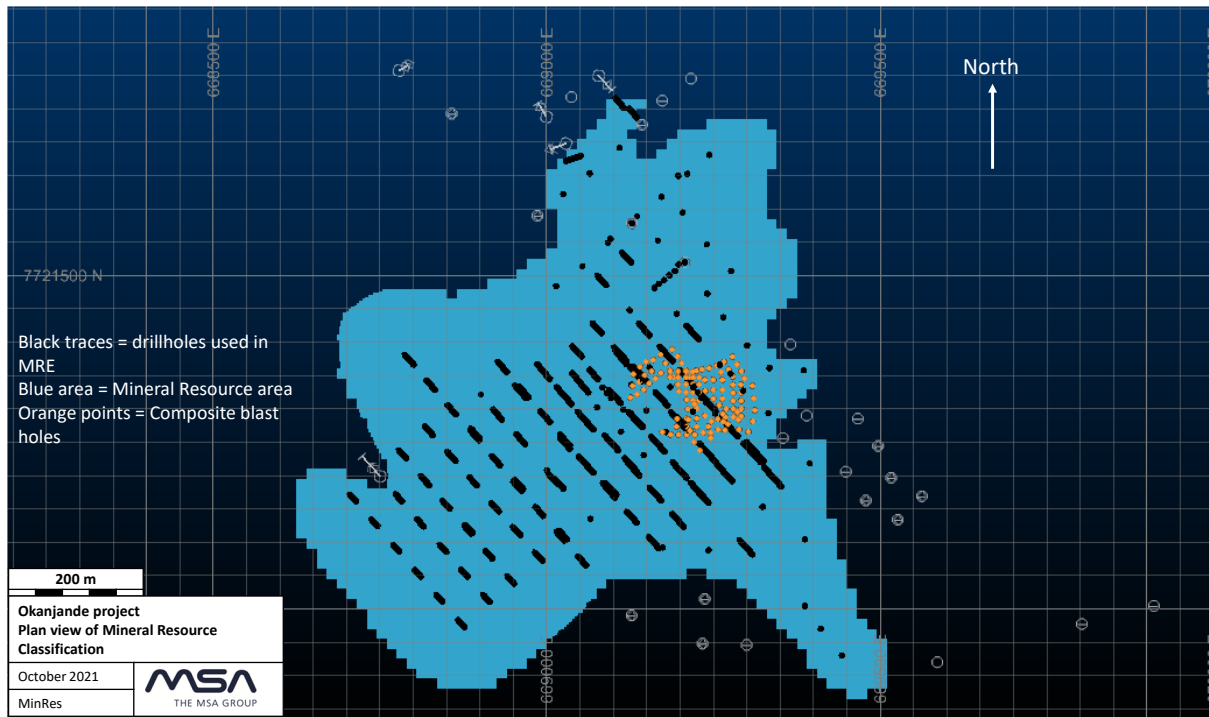


Figure 14-22: Plan view of the Mineral Resource reconciliation (MSA, 2021)

It is the QP’s opinion that the reconciliation validates the estimated TGC grades, albeit over a limited area.

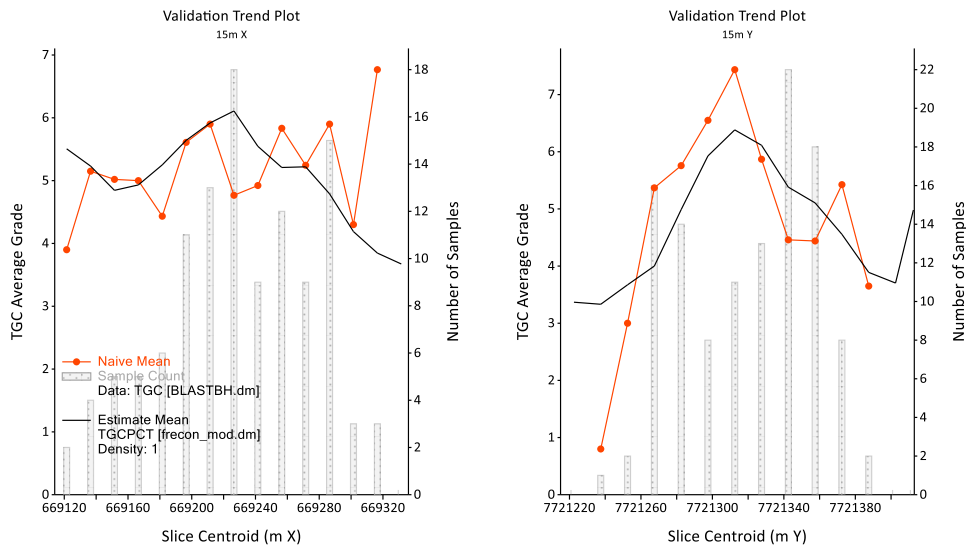


Figure 14-23: Reconciliation between Mineral Resource estimates and blast holes (MSA, 2021)

14.14 REASONABLE PROSPECTS FOR EVENTUAL ECONOMIC EXTRACTION (RPEEE)

MSA carried out a high-level economic analysis to determine “Reasonable Prospects for Eventual Economic Extraction” (“RPEEE”) as required by CIM Definition Standards for Mineral Resources and Mineral Reserves (2014).

A conceptual pit shape was manually created with a pit slope angle of 50° for the fresh rock, and a Whittle pitshell was created with pit slope angle of 45° for the weathered rock in order to determine the stripping ratio (Figure 14-24). The maximum depth of the conceptual pit is 110m, which is the depth of the model considered for Mineral Resources using the extrapolation distances described in Section 14.12. The fresh pit is at least 350m to the mining license boundary. The cost and revenue assumptions are stated in Table 14-5.

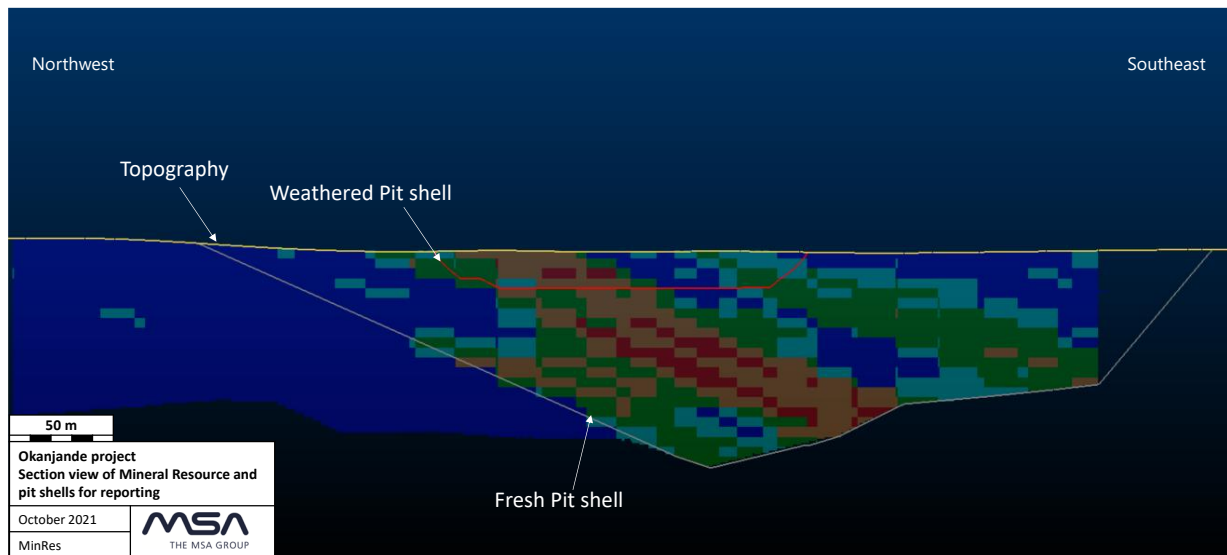


Figure 14-24: Section view of the block model and the pit shells used to report the Mineral Resource (MSA, 2021)

Table 14-5: Cost and Revenue assumptions for RPEEE (MSA, 2021)

Material type	Parameters	Value
Weathered and Transitional	Graphite Product Price	USD 1,250/tonne
	All in Mining Cost / Mineralized material t (Stripping ratio 0.54)	USD 3.75/tonne
	Processing Cost	USD 14.9/tonne RoM
	G+A Cost	USD 0.8/tonne RoM
	Royalty	2%
	Recovery	92%
Fresh	Graphite Product Price	USD 1,250/tonne
	All in Mining Cost / Mineralized material t (Stripping ratio 0.75)	USD 5.11/tonne
	Processing Cost	USD 17.9/tonne RoM
	G+A Cost	USD 0.96/tonne RoM
	Royalty	2%
	Recovery	92%

Costs and revenue inputs assume the following:

- Mineralisation will be processed at a new plant to be constructed at the Okanjande site. The plant may be further modified if/ when fresh rock is processed;
- The recovery is based on historical pilot plant performance and test work in progress (refer to section 6 History and section 13.3.2 Flowsheet Confirmation Test Work, METPRO/SGS 2021);
- Mining costs are based on approximate contractor quotations provided for the operation;
- Processing costs are based on study work done by URE Consulting Inc – by D Marsh, now of Northern Graphite;
- Product price is based on market study summarised in Section 19 of the of the Mineral Resource report dated 15 December 2021.
- Fresh mineralisation, mining and processing costs were adjusted for the increased stripping ratio and higher cost of mining and milling harder rock.

The assessment to satisfy the criteria of Reasonable Prospects for Eventual Economic Extraction (RPEEE) is not an attempt to estimate mineral reserves and does in no way whatsoever imply that the Project has been demonstrated to be economical.

14.15 MINERAL RESOURCE STATEMENT

The Weathered Mineral Resource, as at November 29, 2021, is reported in accordance with the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014), as reported in Table 14-6.

Table 14-6: Weathered Mineral Resource at a cut-off grade of 2.6% TGC as at 29 November 2021 (MSA, 2021)

Material	Classification Category	Tonnes (Mt)	TGC (%)	Graphite content (kilo tonnes)
Weathered	Measured	1.7	4.66	80
	Indicated	4.2	4.03	168
	Subtotal	5.9	4.21	248
	Inferred	0.5	3.45	17
Transitional	Measured	0.2	5.42	13
	Indicated	1.0	4.08	40
	Subtotal	1.2	4.35	53
	Inferred	0.1	3.20	2

Notes:

1. All tabulated data have been rounded and as a result minor computational errors may occur.
2. Mineral Resources which are not Mineral Reserves have no demonstrated economic viability.
3. Inferred Mineral Resources are reported separately from other categories.
4. The Mineral Resources reported are the total Mineral Resources for the Project, regardless of ownership.
5. The Mineral Resource is reported for mineralization contained within pit shells above a cut-off grade of 2.6% TGC, which is based on a product price of USD 1,250/t TGC, mining costs of USD 3.75/t RoM, processing and treatment costs of 14.9 USD/t (RoM), G&A USD 0.8/t (RoM), 2% royalty, concentrate recovery 92%.

6. MSA is not aware of any known environmental, permitting, legal, title-related, taxation, socio-political, marketing, or other relevant issue that could materially affect the Mineral Resource Estimate.
7. The Inferred Mineral Resource has a lower level of confidence than that applying to an Indicated Mineral Resource and must not be converted to a Mineral Reserve. It is reasonably expected that the majority of Inferred Mineral Resources could be upgraded to Indicated Mineral Resources with continued exploration.

Table 14-8, and Figure 14-25 and Figure 14-26 represent grade tonnage tables and plots for the Weathered Mineral Resource reported within the pitshells. The tables and plots are prepared for Measured and Indicated Mineral Resources and separately for Inferred Mineral Resources.

Table 14-7: Grade Tonnage table for Measured and Indicated Weathered Mineral Resource within the pitshells as at 29 November 2021 (MSA, 2021)

TGC Cut-off grade (%)	Tonnes above cut-off grade	Average grade above cut-off grade (%)
2.0	9.1	3.82
2.5	7.4	4.17
2.6	7.1	4.24
3.0	5.9	4.54
3.5	4.5	4.92
4.0	3.5	5.28
4.5	2.5	5.70
5.0	1.8	6.07

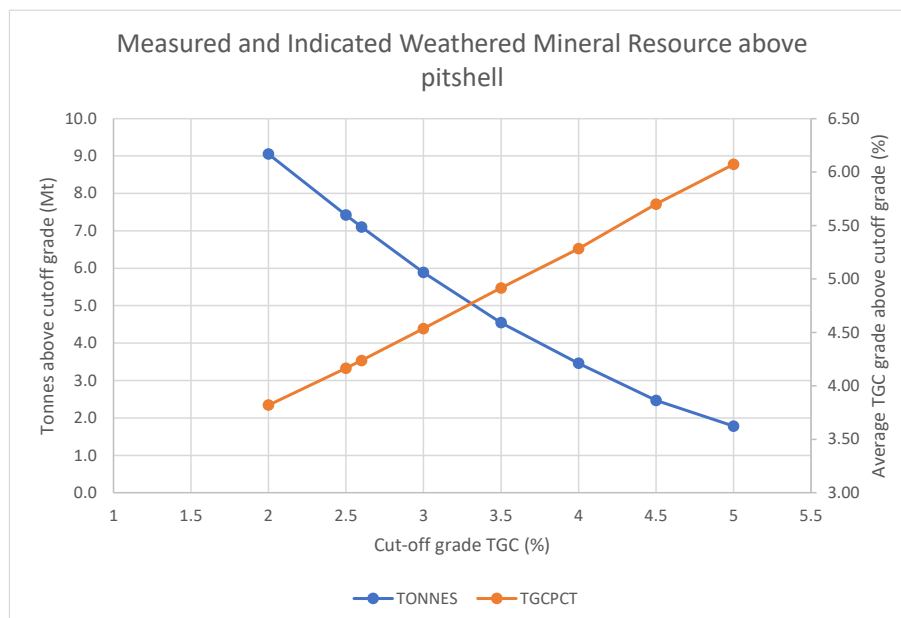


Figure 14-25: Weathered Mineral Resource Grade Tonnage plot within the pitshells as at 29 November 2021 (MSA, 2021)

Table 14-8: Grade Tonnage table for Inferred Weathered Mineral Resource within the pitshells as at 29 November 2021

TGC Cut-off grade (%)	Tonnes above cut-off grade	Average grade above cut-off grade (%)
2.0	0.8	3.04
2.5	0.6	3.36
2.6	0.6	3.42
3.0	0.4	3.69
3.5	0.2	4.18
4.0	0.1	5.09
4.5	0.0	5.51
5.0	0.0	5.87

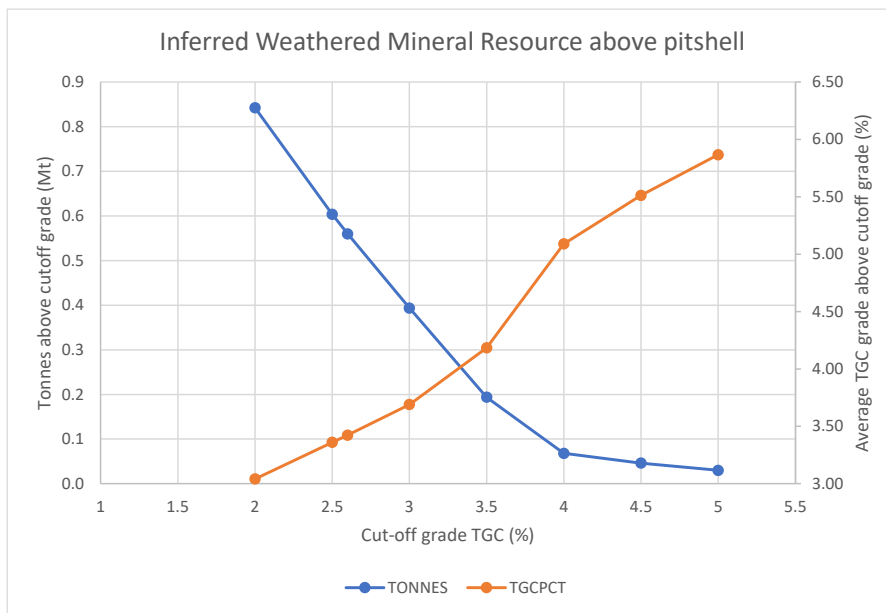


Figure 14-26: Grade Tonnage plot for Inferred Weathered Mineral Resource within the pitshells as at 29 November 2021

The Fresh Mineral Resource, as at 29 November 2021, is reported in accordance with the CIM Definition Standards for Mineral Resources and Mineral Reserves (2014), as reported in Table 14-9.

Table 14-9: Fresh Mineral Resource at a cut-off grade of 3.1% TGC as at 29 November 2021 (MSA, 2021)

Classification Category	Tonnes (Mt)	TGC (%)	Graphite content (kilo tonnes)
Measured	7.1	5.86	419
Indicated	17.0	5.10	868
Subtotal	24.2	5.33	1,287
Inferred	7.2	5.02	359

Notes:

1. All tabulated data have been rounded and as a result minor computational errors may occur.
2. Mineral Resources which are not Mineral Reserves have no demonstrated economic viability.
3. Inferred Mineral Resources are reported separately from other categories.
4. The Mineral Resources reported are the total Mineral Resources for the Project, regardless of ownership.
5. The Mineral Resource is reported for mineralization contained within a Whittle pit shell above a cut-off grade of 3.1% TGC, which is based on a product price of USD 1,250/t TGC, mining costs of USD 5.11/t RoM, processing and treatment costs of 17.88 USD/t (RoM), G&A USD 0.96/t (RoM), 2% royalty, concentrate recovery 92%.
6. MSA is not aware of any known environmental, permitting, legal, title-related, taxation, socio-political, marketing, or other relevant issue that could materially affect the Mineral Resource Estimate.
7. The Inferred Mineral Resource has a lower level of confidence than that applying to an Indicated Mineral Resource and must not be converted to a Mineral Reserve. It is reasonably expected that the majority of Inferred Mineral Resources could be upgraded to Indicated Mineral Resources with continued exploration.

Table 14-10 and Table 14-11, and Figure 14-27 and Figure 14-28 represent grade tonnage tables and plots for the Fresh Mineral Resource reported within the pitshell. The tables and plots are prepared for Measured and Indicated Mineral Resources and separately for Inferred Mineral Resources.

Table 14-10: Grade tonnage table for Measured and Indicated Fresh Mineral Resource within the pitshell as at 29 November 2021 (MSA, 2021)

TGC Cut-off grade (%)	Tonnes above cut-off grade	Average grade above cut-off grade (%)
3.0	24.8	5.27
3.1	24.2	5.33
3.5	21.7	5.55
4.0	18.8	5.83
4.5	16.0	6.11
5.0	13.0	6.43
5.5	10.4	6.73
6.0	7.9	7.04

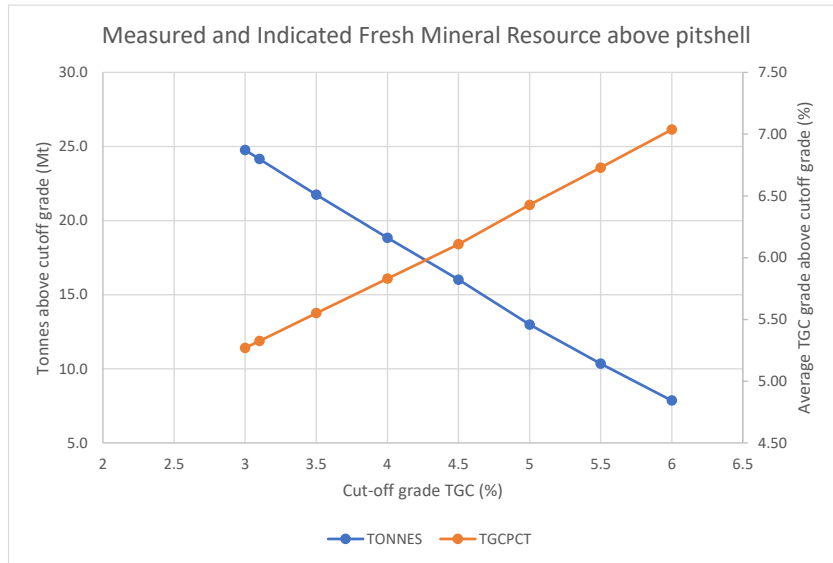


Figure 14-27: Grade Tonnage plot for Measured and Indicated Fresh Mineral Resource within the pitshell as at 29 November 2021 (MSA, 2021)

Table 14-11: Grade Tonnage table for Inferred Fresh Mineral Resource within the pitshell as at 29 November 2021 (MSA, 2021)

TGC Cut-off grade (%)	Tonnes above cut-off grade	Average grade above cut-off grade (%)
3.0	7.6	4.91
3.1	7.2	5.02
3.5	6.0	5.35
4.0	4.8	5.74
4.5	3.9	6.09
5.0	3.0	6.48
5.5	2.4	6.80
6.0	1.9	7.10

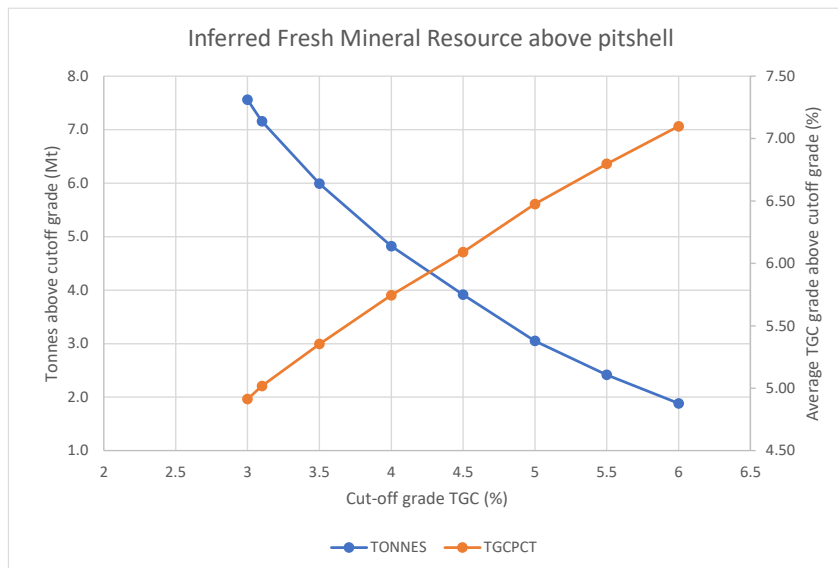


Figure 14-28: Grade Tonnage plot for Inferred Fresh Mineral Resource within the pitshell as at 29 November 2021 (MSA, 2021)

15 MINERAL RESERVE ESTIMATES

No mineral reserve estimate available at the time of writing.

16 MINING METHODS

(This section was compiled by The MSA Group Pty. Ltd., with section 16.5 by CREO Engineering Solutions Pty Ltd)

16.1 BACKGROUND

The Okanjande graphite deposit lends itself to being mined by opencast mining methods. The intention is to mine at an average rate of 631,450 tonnes of mineralised material per year which will be processed through the Okanjande processing plant.

This report considers a 10-year mining plan, commencing with the mining of the weathered graphite-bearing material and ending with a short period of mining the transitional and fresh graphite-bearing material underlying the weathered material. The Okanjande Deposit was mined from August 2017 until October 2018, when mining ceased and the Okanjande Mine was placed on care and maintenance. The run of mine (“RoM”) graphite-bearing material was trucked to Okorusu for processing.

The intention is to mine the Okanjande deposit by making use of contractors who will supply their own mining equipment to deliver the graphite-bearing material to a product stockpile and the balance to the Integrated Waste Dump (IWD). Overburden will be stockpiled on a separate overburden stockpile. It is not envisaged that any waste material will be backfilled into the pit at this stage as there is further fresh graphite-bearing material at depth; this fresh graphite-bearing material will be the subject of a future economic assessment.

The mining contractors will also be responsible for crushing the mineralized material (single stage jaw crusher).

16.2 MINE SCHEDULING

The Minex geological model, which was reviewed by the MSA Mineral Resources team, was uploaded into Deswik mine design software. A pit shell was defined incorporating only the weathered graphite-bearing material and using geotechnical parameters as defined in the previous mine design contemplated by Imerys-Gecko. This pit shell did not yield the required tonnage for a 10-year mine plan at a planned production of 631,450 tonnes per annum (“tpa”). The pit shell was then reconfigured and expanded to include some of the transitional and fresh graphite-bearing material in order to achieve a 10-year life of mine at a production rate of approximately 631,450 tpa graphite-bearing material to achieve approximately 31,103 tonnes of product per annum (Figure 16-1). A recovery of 92% was used to determine the final saleable product.

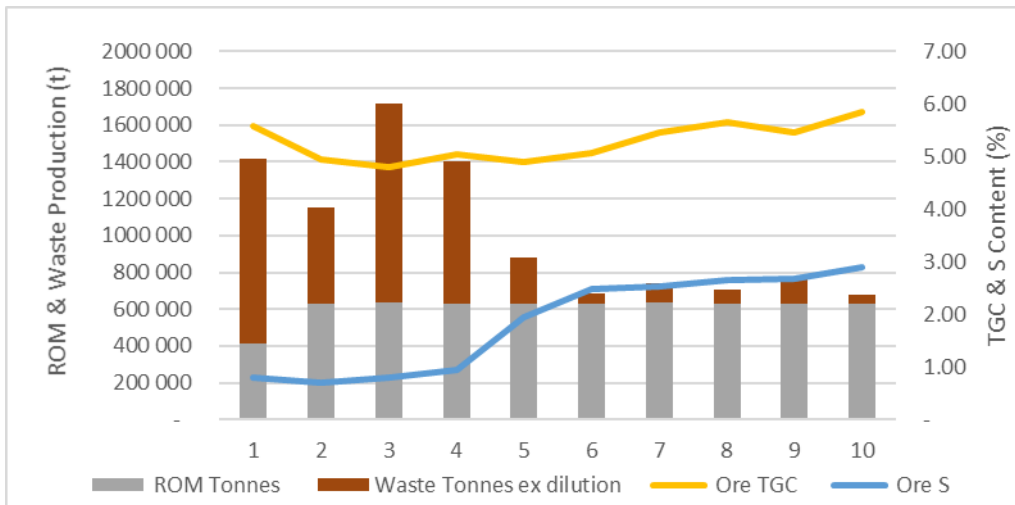


Figure 16-1: Production Profile (MSA, 2021)

16.2.1 Pit Design Parameters

The open pit design parameters considered are summarised below:

- Bench heights: 5m
- Bench widths: 10 m to 70m
- Slope angles: 45° to 55°
- Cut-off Grade: 4% TGC
- Dilution to RoM tonnes: 2%
- Recovery: 92%.

16.2.2 Schedules

16.2.2.1 Base Case 4% TGC Cut-off

Using the above design parameters, a mining schedule was developed to mine at a rate of approximately 631,450tpa at an average Stripping Ratio (“S/R”) of 0.66 (tonnes of waste to one tonne of graphite-bearing material) over the 10-year period (Table 16-1). The first four years consider only mining the weathered graphite-bearing material. In the second quarter of Year 5 the mining of Fresh graphite-bearing material begins and continues until the end of the 10-year period. The ramp up to full production is rapid, with the first year’s mining opening up the pit utilizing the current established faces (in the historic Okanjande open pit) to create pit room to sustain the maximum tonnage required.

The total Graphite production at a 4% TGC cut-off over the 10-year LOM is 295 548 tonnes at an average rate of 29 548 tonnes per annum (tpa).

Table 16-1: 10-year mining schedule @4% TGC cut-off (MSA, 2021)

	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	TOTAL
Total Mining (t)	1 419 921	1 154 011	1 715 632	1 401 837	878 519	685 038	738 987	707 697	764 570	675 780	10 141 990
Waste (t)	1 009 479	522 561	1 082 452	770 387	247 069	53 588	105 807	76 247	133 120	44 330	4 045 040
ROM (t)	410 443	631 450	633 180	631 450	631 450	631 450	633 180	631 450	631 450	631 450	6 096 950
TGC %	5.58	4.95	4.81	5.05	4.90	5.06	5.46	5.65	5.47	5.86	5.27
S %	0.79	0.71	0.79	0.95	1.94	2.48	2.54	2.65	2.69	2.90	4.85
S/R	2.46	0.83	1.71	1.22	0.39	0.08	0.17	0.12	0.21	0.07	0.66
Product TGC (t)	21 070	28 756	28 019	29 337	28 453	29 395	31 806	32 823	31 777	34 043	295 480

Mining is performed by two excavator teams operating at a rate of 3,000 tonnes per day (“tpd”) on both the waste and mineralized material (Table 16-2). Two teams will be required for the first 5 years, reducing to 1 team for the last 5 years.

Table 16-2: Production Capacity

Quantity		Production Rate 3000 tpd	
	Days per Month	22	
2	Excavator	132,000	tpm
12	Months	1,584,000	tpa

Notes: *Additional Waste stripping will be required to be stripped in year 2 to reduce the stripping requirement in year 3 where the required capacity is higher than the equipment capacity.*

The progress plots in (Figure 16-2) and schedules (Table 16-3 and Table 16-4) show the mining advance over the 10-year plan. In the progress plots, the pink represents the weathered material while the blue represents the fresh material to be mined (the orange around the edges shows the final pit shell). In Year 1, mining starts in the southwest portion of the pit and opens in all directions to create pit room. The volume of waste is high in the first year due to the expansion of the pit; this contributes to the high strip ratio in the first year of 2.46 tonnes of waste per one tonne of graphite-bearing material. The total tonnage to be mined decreases as the waste decreases, so equipment requirements will decrease after the first 5 years.

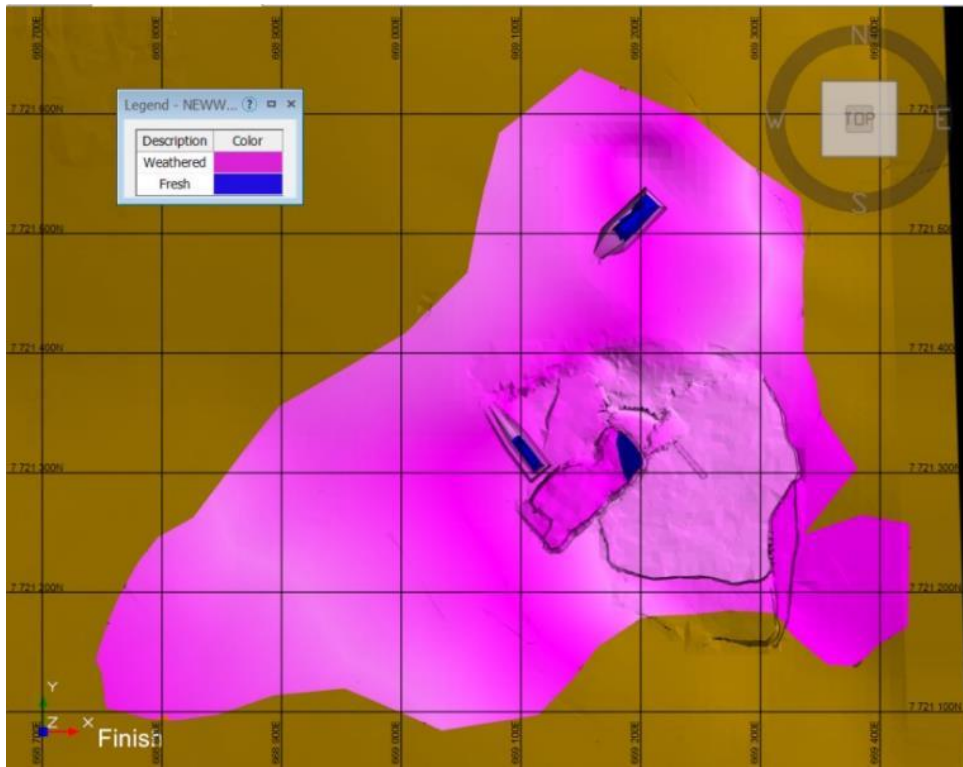
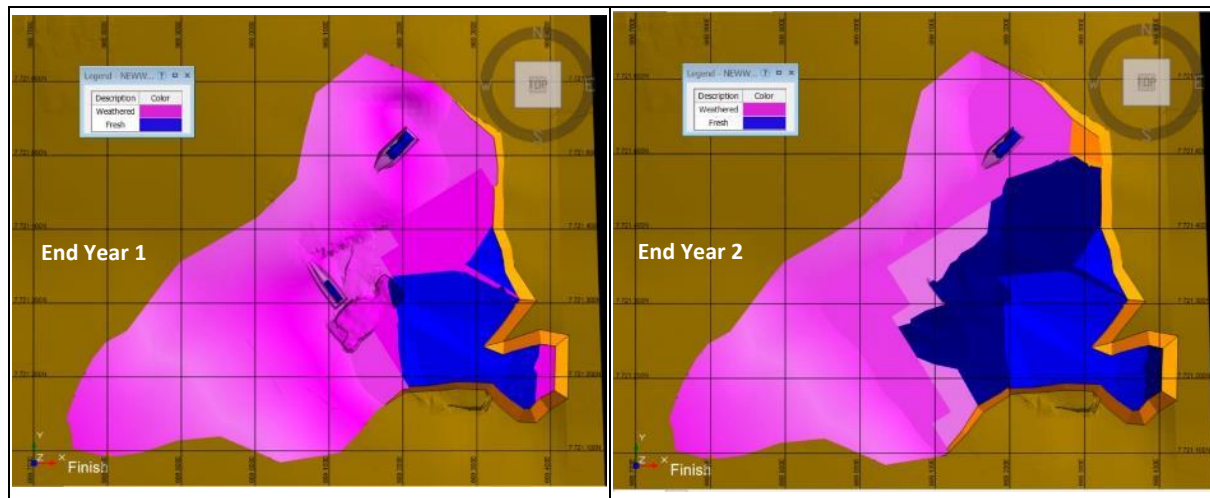
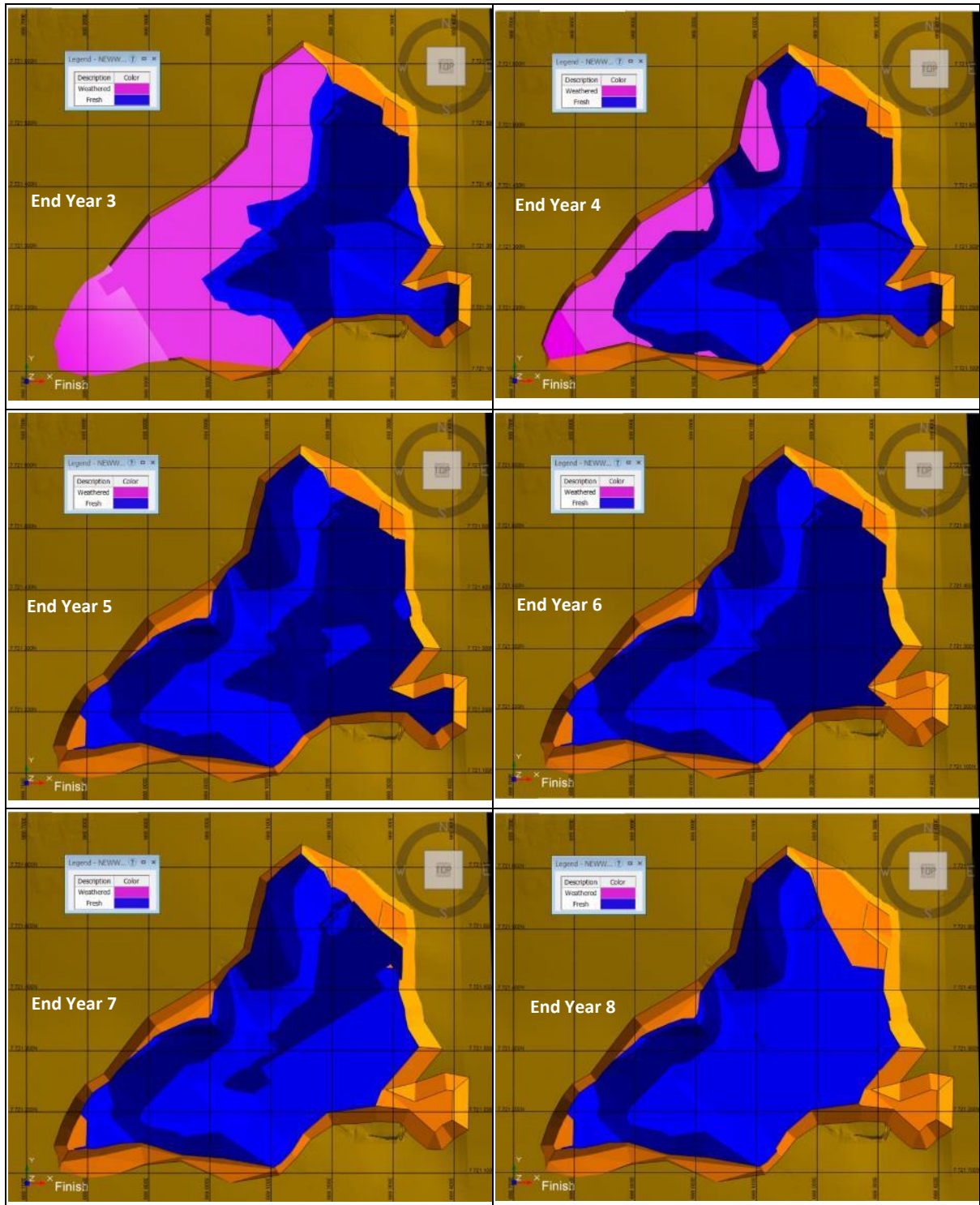


Figure 16-2: Okanjande Pit at start of Mining (MSA, 2021)





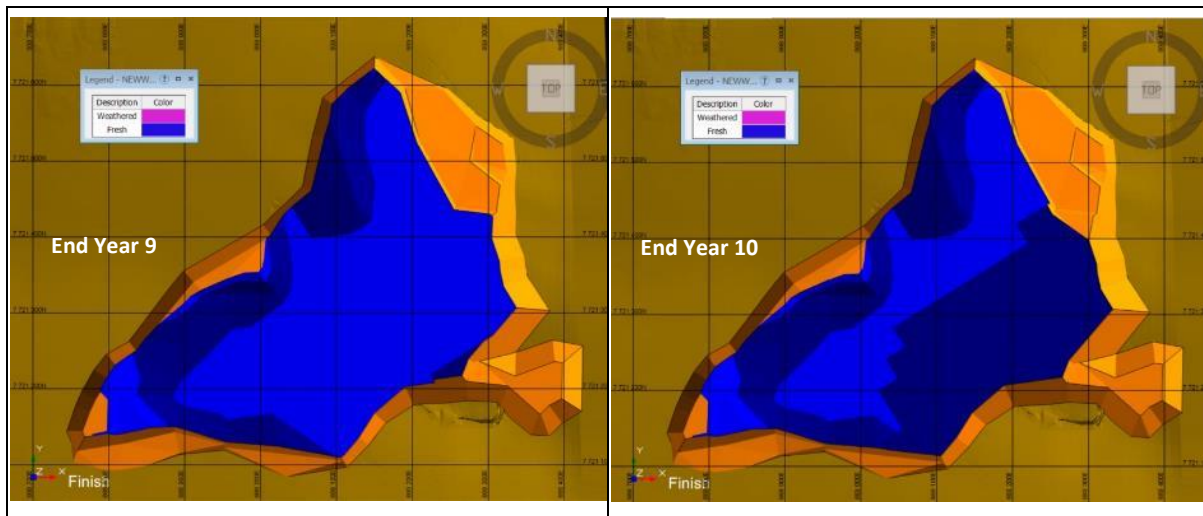


Figure 16-3: Okanjande 10-year Pit Progress Plots (MSA, 2021)

Table 16-3: Mining Schedule – Weathered graphite-bearing material (MSA, 2021)

	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	TOTAL
ROM (t)	410 442	631 450	633 179	631 450	234 102	-	-	-	-	-	2,540,623
TGC %	5.58	4.95	4.81	5.05	4.86	-	-	-	-	-	5.03
S %	0.79	0.71	0.79	0.95	1.03	-	-	-	-	-	0.83
Product TGC (t)	21 068	28 775	27 996	29 330	10 458	-	-	-	-	-	117,650

Table 16-4: Mining Schedule – Fresh graphite-bearing material (MSA, 2021)

	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	TOTAL
ROM (t)	-	-	-	-	397 347	631 450	633 180	631 450	631 450	631 450	3,556,327
TGC %	-	-	-	-	4.92	5.06	5.46	5.65	5.47	5.86	5.44
S %	-	-	-	-	2.48	2.48	2.54	2.65	2.69	2.90	2.63
Product TGC (t)	-	-	-	-	17 992	29 381	31 800	32 828	31 761	34 035	177,829

16.2.2.2 Increasing and Decreasing TGC Cut-off

An exercise was carried out to look at the effect of increasing and decreasing the cut-off grade to see the effect on the Graphite production, depletion of weathered material and total tonnes moved. Table 16-5 below shows the results for 3, 3.5, 4, 4.5 and 5% TGC cut-offs. The base case of 4%TGC is highlighted.

Table 16-5: 10-year Mining Schedule @ 3, 3.5, 4, 4.5 and 5% TGC cut-off (MSA, 2021)

	3.0%	3.5%	4.0%	4.5%	5.0%
Total Tonnes	8 464 034	9 193 913	10 141 990	11 086 863	11 938 342
Waste Tonnes	2 347 081	3 096 960	4 045 040	4 989 909	5 849 142
ROM Tonnes	6 096 953	6 096 953	6 096 950	6 096 953	6 089 199
TGC	4.75	5.00	5.27	5.41	5.50
S	1.40	1.61	1.88	2.11	2.48
S/R	0.38	0.51	0.66	0.82	0.96
Product Tonnes	266 338	280 259	295 480	303 441	308 119

The total Graphite production increases as the cut-off percentage increases with increased waste produced and the weathered material is mined more quickly.

The mining of fresh graphite material starts one year earlier for each half percent increase in cut-off grade. Starting at year 7 at the 3% cut-off it reduces to year 3 at 5% cut off.

The tables below show the mining schedules for the 3.5% and 4.5% cut-off grades, at 3.5% cut-off the weathered material is depleted in year 6 of mining and at 4.5% it is depleted in year 4.

Table 16-6: Mining Schedule @3.5% TGC cut-off

	2022	2023	2024	2025	2026	2027	2028	2029	2030	2031	TOTAL
Waste (t)	712,209	301,899	551,091	555,711	420,592	171,711	110,700	107,731	68,914	96,400	3,096,960
Weathered											
ROM (t)	410,443	631,450	633,180	631,450	631,450	457,784	-	-	-	-	3,395,756
Grade (% TGC)	4.75	4.95	4.64	4.57	4.60	4.76	-	-	-	-	4.71
Grade (% S°)	0.76	0.76	0.76	0.83	0.97	0.93	-	-	-	-	0.84
Concentrate (t)	17,931	28,779	27,055	26,542	26,742	20,065	-	-	-	-	147,114
Fresh											
ROM (t)	-	-	-	-	-	173,666	633,180	631,450	631,450	631,450	2,701,196
Grade (% TGC)	-	-	-	-	-	4.50	5.11	5.30	5.51	5.75	5.36
Grade (% S°)	-	-	-	-	-	2.35	2.51	2.42	2.65	2.86	2.59
Concentrate (t)	-	-	-	-	-	7,194	29,773	30,763	32,027	33,387	133,144
Total Mining (t)	1,122,652	933,349	1,184,271	1,187,161	1,052,042	803,161	743,880	739,181	700,364	727,850	9,193,912
Total Concentrate	17,931	28,779	27,055	26,542	26,742	27,259	29,773	30,763	32,027	33,387	280,258

Table 16-7: Mining Schedule @4.0% TGC cut-off

	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	TOTAL
Waste (t)	1,347,566	1,278,819	1,552,292	242,060	100,981	50,447	99,233	110,168	44,330	164,012	4,989,909
Weathered											
ROM (t)	410,442	631,450	633,180	87,682	-	-	-	-	-	-	2,540,626
Grade (% TGC)	5.80	5.21	5.29	5.27	-	-	-	-	-	-	5.38
Grade (% S°)	0.78	0.77	0.90	1.11	-	-	-	-	-	-	0.84
Concentrate (t)	21,885	30,271	30,791	4,255	-	-	-	-	-	-	87,203
Fresh											
ROM (t)	-	-	-	534,769	631,450	631,450	631,450	631,450	631,450	631,450	3,556,328
Grade (% TGC)	-	-	-	4.97	5.06	5.60	5.64	5.60	5.74	5.28	5.39
Grade (% S°)	-	-	-	2.40	2.43	2.73	2.69	2.65	2.93	2.56	2.69
Concentrate (t)	-	-	-	24,857	29,386	32,538	32,852	32,549	33,359	30,696	216,237
Total Mining (t)	1,758,009	1,910,269	2,185,472	873,510	732,431	681,897	732,413	741,618	675,780	795,462	11,086,863
Total Concentrate	21,885	30,271	30,791	29,112	29,386	32,538	32,851	32,549	33,359	30,696	303,441

16.3 MINING RECOVERY

Given the nature of the Okanjande deposit, 92% recovery with 2% dilution is anticipated, the dilution is based on barren rock within the Okanjande deposit which is randomly dispersed and differs in thickness and orientation throughout. The Strip ratio of the Okanjande deposit is low for the 4% TGC cut-off and the relatively small amount of waste rock is stockpiled on a dedicated rock dump.

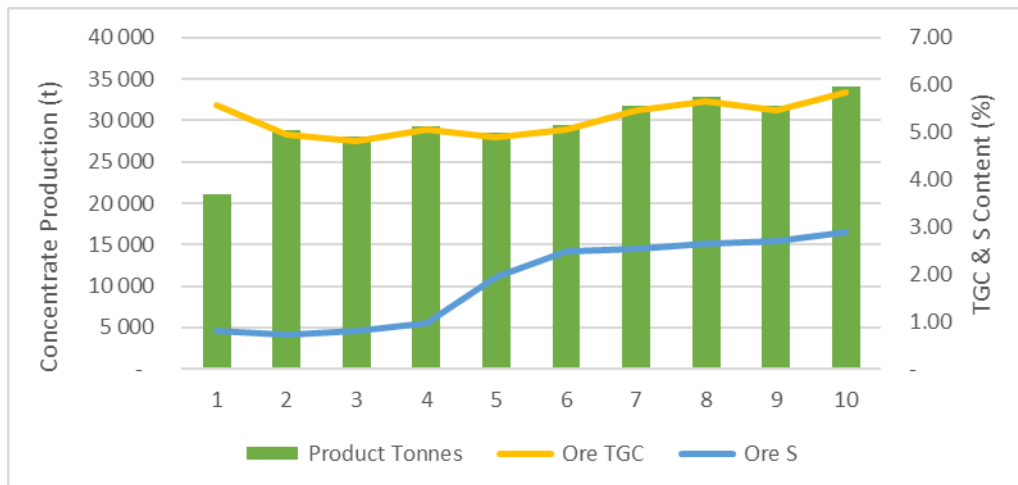


Figure 16-4: Okanjande Graphite Concentrate Product per year @ 4% TGC cut-off (MSA, 2021)

16.4 MINING EQUIPMENT

The following mining equipment is proposed to undertake the mining of the graphite-bearing material and waste at the Okanjande deposit (Table 16-8).

Table 16-8: Mining Equipment

Quantity	Description	Size / Capacity
2	Excavator 60 t	60 t
5	Articulated Dump Trucks ("ADTs")	30 t
1	Drill Rig	Top Hammer 102mm diameter
1	Track Dozer	40 t
1	Motor Grader	CAT140/Bell 770
1	Water Bowser	15,000 l
1	Lubrication Service Truck	
2	Front End Loader ("FEL")	3 -5 m ³
1	Diesel Bowser	15,000 l

In addition to the mining equipment specified above (Table 16-8), Light Delivery Vehicles (LDVs) for supervisors and support staff are also required.

If crushing of the ROM mineralized material at the Okanjande deposit is required an additional FEL and excavator to feed the crusher will be required on the site.

16.4.1 Drilling and Blasting

It is proposed that holes of 102mm diameter are drilled into mineralised material and waste to get a final powder factor of 0.7 kg/m³.

To achieve this powder factor with 102mm holes and 5m high benches, each bench is drilled on a 3m by 3m pattern and charged with an emulsion explosive agent. If one considers a charge length of 3m and an emulsion density of 1.3, the mass of explosives per hole will be approximately 32kg. The powder factor can be increased by reducing the burden and spacing if the fragmentation needs to be increased.

Drilling and blasting is planned to be done by the mining contractor or by their sub-contractor. Blocks corresponding to a week's loading are to be drilled and blasted on a weekly basis. On blasting day, the explosive supplier does a bench delivery of emulsion and accessories for that blast and the mining contractor oversees the blast.

16.4.2 Mining Cost

Mining costs for the operation were obtained from three budget quotations provided by mining contractors in Namibia. A review of the costs by URE Consulting Inc concludes that costs provided by two contractors were very similar and recommended the figures in Table 16-9 be used in the economic assessment of the Okanjande deposit and that closer to the start of mining, two of the three contractors will be asked for full and final quotes.

The mining cost considered is a combination of Preliminary and General, Load and Haul and Drill and Blast costs and is summarised in Table 16-9. These costs are wet rates and include the cost of diesel which will either be supplied by the owner or the contractor, subject to final negotiations.

Table 16-9: Mining and Transport Costs (Namibian Dollar)

Operating Costs	Units	Cost
Establishment and De-establishment	N\$, Once-off	3,064,250
P&Gs	N\$/a	12,862,591
Load, haul and dump top soil	N\$/m ³	51.45
Drill & blast waste and mineralized material	N\$/m ³	43.84
Load, haul and dump waste	N\$/m ³	51.45
Load haul and stockpile mineralized material	N\$/t	56.47

Drilling and blasting costs are estimated using a powder factor of 0.6kg/m³ and will be higher if the recommended powder factor of 0.7kg/m³ is used.

There are additional *ad-hoc* costs which have also been quoted for standing time, off-mine road maintenance and bush clearing.

The access road to the mine needs to be maintained and the estimated cost for this from the quotes received is in the order of N\$3.2 million per annum or N\$5.07 per tonne of graphite-bearing material.

16.5 INTEGRATED WASTE DUMP DESIGN (IWD)

(This section was compiled by CREO Engineering Solutions (Pty) Ltd.)

16.5.1 IWD Design Criteria

Waste Rock from the Okanjande mining operations will be co-deposited with dry (filtered) tailings from the flotation process on a storage facility located to the northeast of the process plant. A dry tailings system was selected to minimize water consumption for the project.

Table 16-10 summarises the design criteria and the assumed values of material properties for the design of the IWD. Table 16-11 provides the design quantities used for the design of the IWD.

Table 16-10: Okanjande waste rock dump design criteria

Design Basis	UOM	Value	Source
Design Life of Facility	a	10	NG
Total Waste Rock Tonnes (moist)	t	4,213,583	MSA
Total Filtered Tailings Tonnes (moist)	t	6,623,751	CREO
Total Co-Deposited Material tonnes (moist)	t	10,837,335	CREO
Waste Rock Moisture	% w/w	4.00	CREO
Filtered Tailings Moisture	% w/w	12.60	CREO Mass Balance
Co-Deposited Tailings/Waste Rock stacked density	t/m3	1.50	CREO
Co-Deposited Tailings/Waste Rock Compacted density	t/m3	1.80	CREO
Total Co-Deposited Material Volume (moist)	m3	6,020,741	CREO
IWD Final Compacted Slope	L/H	3.50	CREO
Lift Height	m	10	CREO
Number of Lifts	#	4	CREO
Total Height	m	40	CREO

Table 16-11: Okanjande TSF Design Quantities

Years	Unit	1	2	3	4	5	6	7	8	9	10	11	Totals
		2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	
Weathered WR	t	525,770	797,938	835,944	965,020	461,426	0	0	0	0	0	0	3,586,098
Fresh WR	t	0	0	0	0	68,499	156,592	83,018	94,820	109,045	92,422	23,089	627,485
Total WR	t	525,770	797,938	835,944	965,020	529,925	156,592	83,018	94,820	109,045	92,422	23,089	4,213,583
Weathered Tails	t	222,250	566,355	689,638	689,292	599,135	0	0	0	0	0	0	2,766,671
Fresh Tails	t	0	0	0	0	89,436	687,483	687,002	684,959	683,987	683,260	340,955	3,857,080
Total Tails	t	222,250	566,355	689,638	689,292	688,571	687,483	687,002	684,959	683,987	683,260	340,955	6,623,751
Total IWD	t	748,021	1,364,293	1,525,583	1,654,312	1,218,496	844,075	770,020	779,779	793,032	775,681	364,043	10,837,335
Stacked Vol	m3	498,680	909,529	1,017,055	1,102,875	812,330	562,717	513,347	519,853	528,688	517,121	242,695	7,224,890
Total Compacted Vol	m3	415,567	757,940	847,546	919,062	676,942	468,931	427,789	433,210	440,573	430,934	202,246	6,020,741

16.5.2 High Level IWD Design

The design of the IWD follows outcomes of the EIA conducted in 2-14 for the Gecko graphite process at the Okanjande site. The location of the IWD was chosen over the outcrop of Swakop Group marble band which is located in north-western corner of the mining license. The Swakop Group marble strata is characterized by carbonate rich rock with low permeability. This footing will minimize penetration to ground water and provide a measure of neutralization in the case that any potentially acid generating drainage or run-off should seep out of the IWD. See Figure 16-6.

Geochemical testing of tailings and waste rock indicated that all weathered tailings and waste rock are potentially acid generating and that all transitional and fresh tailings and waste rock are acid generating. For this reason, the IWD will be lined with a double HDPE liner with leak detection. Run-off water from the IWD is channelled to a lined collection pond, where run-off may be neutralized and recovered to the process.

The first 2 years of pad space and run-off collection will be initially constructed ahead of operation, after which the base area will be expanded regularly ahead as part of the operations.

The IWD base will be constructed with a liner system. The liner system consist of a layer of bedding sand, a double HDPE liner with leak detection, geotextile and an aggregate liner system which will provide protection for the liner as well as providing a drainage layer for seepage. See Figure 16-5 below.

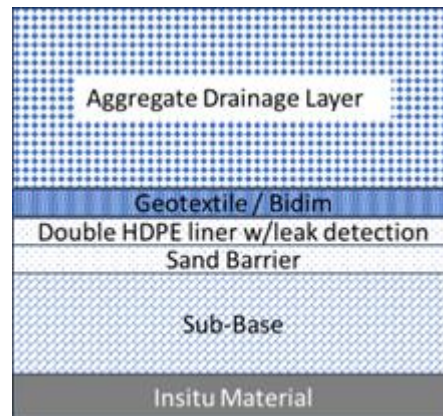


Figure 16-5: IWD Base and Liner Schedule

The base area will be cleared, levelled and sloped and compacted. Run-off will be collected in a lined trench running along the south-eastern side of the IWD and from there run-off will collected in a lined collection pond. The return water system A berm will be constructed around the perimeter of the IWD with a spacing to allow for re-sloping of the facility during closure.

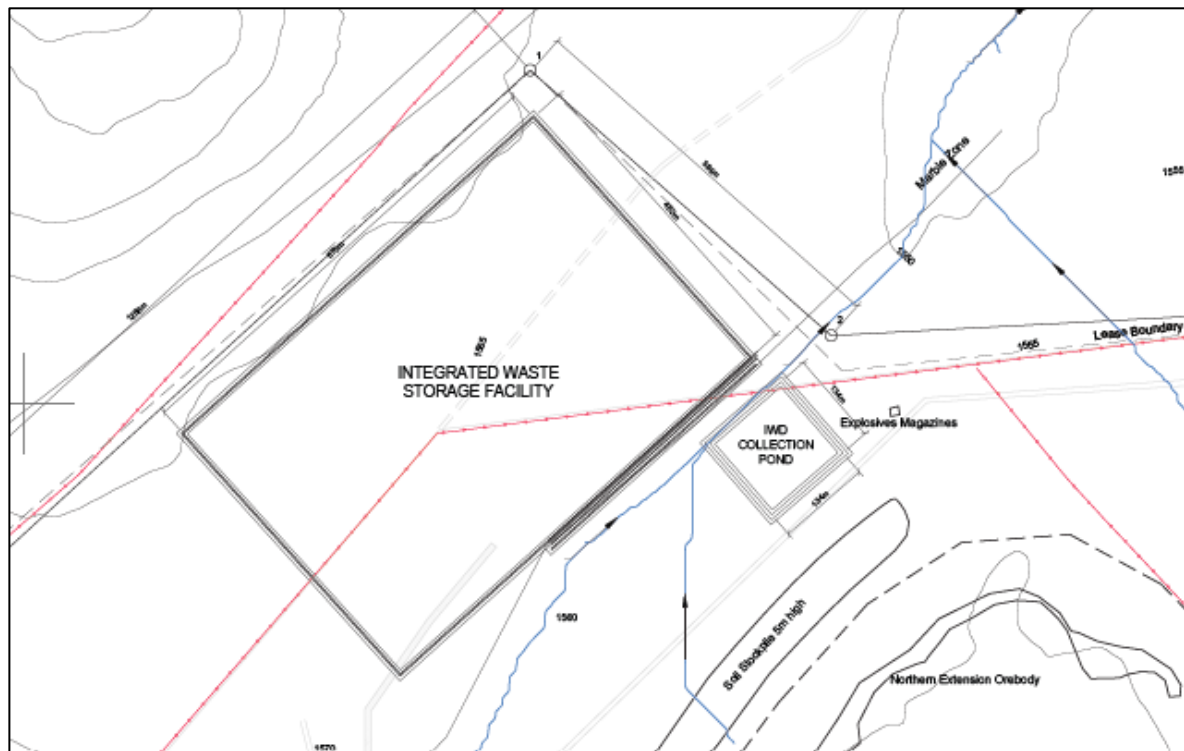


Figure 16-6: Okanjande IWD conceptual footprint

16.6 CONCLUSION

This 10-year mining plan (4%TGC cut-off) considers the mining of weathered as well as fresh graphite bearing material. The IWD design is suitable for all waste and ore rock types. All testing to date indicates that the same processing flowsheet can be used for both fresh and hard rock.

If the cut-off grade is increased, the above adjustments to process the fresh graphite bearing material will be required sooner.

17 RECOVERY METHODS

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

17.1 PREVIOUS IMERYS-GECKO PROCESS

The existing graphite processing plant at Okorusu, which was used to process Okanjande graphite mineralized material during 2017-2018 is based on the Imerys Lac de lles graphite operation in Quebec, Canada (Figure 17-1). This process relied heavily on the use of gravity spirals to up-grade flotation concentrates. While spirals can be a suitable technology to separate gangue from graphite, it is paramount that the graphite is well liberated to produce an acceptable concentrate. However, this degree of liberation is rarely observed after a primary grind and polishing and/or stirred media mills are required in the cleaning circuit to liberate the graphite flakes.

The plant struggled operationally to achieve nameplate production rates and concentrate grades. A number of design and operational issues were identified (METPRO, 2021):

- The rod mill had a high reduction ratio target of 22:1 based on a crush size of P80 = 10mm and a mill discharge size of P80 = 425µm. While reduction ratios of 2-20 are observed for rod mills, a more typical reduction ratio is approximately 8. Depending on the circulating loads, this large reduction ratio could translate into increased flake degradation.
- Graphite projects typically do not require the use of dispersants. The rougher circuit was fed from the grinding circuit at over 46% solids, which would likely have created high viscosities and gangue entrainment issues, which may have triggered the use of the dispersant Betamin. Rougher circuits are typically operated between 25% solids (clayish material) and 35% solids (fresh rock).
- Dowfroth 200 is a very strong poly glycol ether frother that produces a strong and persistent froth. This frother also tends to create entrainment issues in graphite flotation. It appears that the plant may have been switched over to Aerofroth 76A prior to shut down. Aerofroth 76A is a suitable frother based on comparative reagent tests performed on other graphite deposits. Alternatively, MIBC is a standard frother in graphite applications.
- It appeared that collector and frother were only added to the 2nd cleaner stage, which could have led to elevated graphite losses.

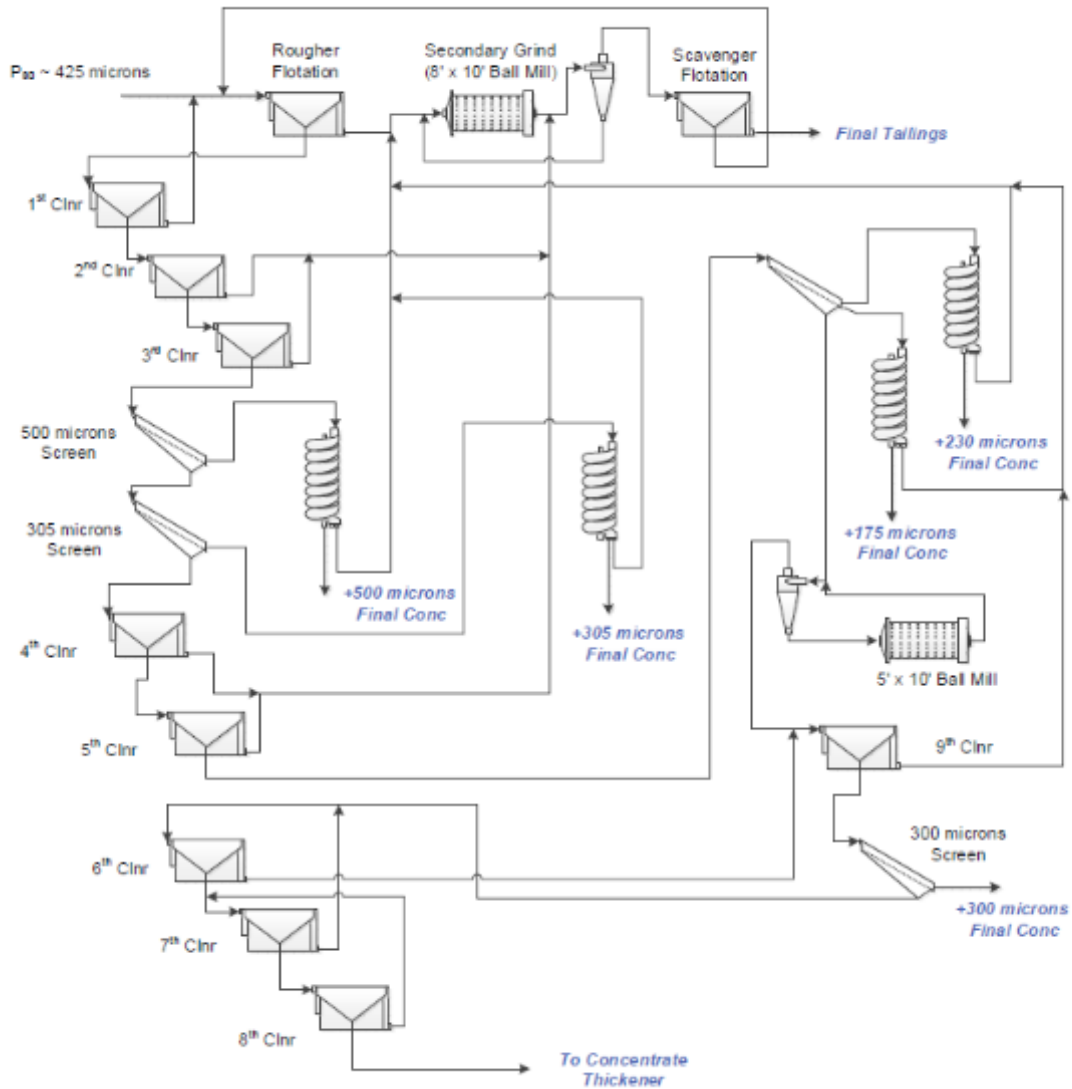


Figure 17-1: Gecko/Imerys Era Okorusu Process (2017-2018)

Due to the processing problems experienced during the Gecko/Imerys era of operation, as well as the results of early MINTEK test work (13.1 Early Process Test Work – MINTEK) and the subsequent SGS metallurgical test work program (13.3.2 Flowsheet Confirmation Test Work, METPRO/SGS 2021), a new process flow sheet was developed which is more suitable to the Okanjande mineralization (Figure 17-3).

17.2 UPDATED NORTHERN GRAPHITE PROCESS DESIGN

17.2.1 Design Basis

The process plant will be relocated to the Okanjande Mine site. The relocation of the process plant will achieve 3 main benefits:

- Reduction of operating costs through elimination of ~90km of mineralized material hauling to the Okoruso site. Mineralized material hauling costs previously contributed to 15% of the overall C1 operational costs.
- Elimination of space constraints. The new process plant location at the Okanjande mine site will allow for unhindered expansions, which are planned for the future. The Okoruso site was space constraint and did not allow for any expansion.
- Tailings Storage. Tailings storage at the Okoruso site was limited and would only allow for storage of 1.2Mt of tailings. The Okanjande site has sufficient real estate for the construction of a tailings facility to accommodate life of mine operations.
- Reduction in Carbon footprint through the removal of the road haulage requirement to the Okoruso site and implementation of solar power generation.

The addition of new comminution and tailings dewatering circuits and the changes to process flow of the existing flotation circuit are the result of incorporation of findings from the outcomes of both early and recent test work programs and are designed to improve flake preservation and improve purity of graphite flake concentrate through following changes:

- A split milling circuit that will prevent excessive recirculation and reduce the potential for flake degradation.
- The split milling circuit will make provision for potentially implementing a flash flotation stage after the Autogenous Mill (AG mill) to further improve large flake recovery.
- Generation of a flotation feed stream of the optimal solids content – 30%.
- A Flotation Feed Buffer Tank to reduce fluctuations in feed rate and solids content and improve steady operation of the flotation circuit.
- The removal of gravity spirals early in the process flow where incomplete liberation would be expected.
- The removal of excessive recirculating streams which hamper effective capacity of process equipment and complicates process control and plant water balance.
- The reduced flotation solids content will remove the need for dispersants.
- The addition of more suitable frother in the form of MIBC.
- Allowance for reagent addition at all flotation banks.
- Reduction of raw water consumption due to the disposal of dry tailings.

The process design/development for the upcoming Northern Graphite processing plant may be separated into 4 distinct areas, each with its own level of engineering changes required:

1. Front-End Comminution: Due to the relative unsuitability of existing comminution equipment, restrictions in terms of accessibility and the cost of relocating this equipment, a completely new front-end comminution circuit is required. This allows for fit-for-purpose process equipment selection to optimize liberation and graphite flake preservation.
2. Flotation: The majority of the existing flotation processing equipment is suitable in terms of throughput capacity, type and condition; and suitable for repurposing in the new graphite process. The only new processing equipment to be included in the flotation circuit will be a polishing scrubber and 2 polishing stirred medial mills. The remainder of processing equipment will be refurbished and piping re-routed to match the updated flow sheet.
3. Concentrate de-Watering, Classification and Bagging: This circuit was designed and constructed by Imerys-Gecko to be fit-for-purpose for graphite concentrate de-watering, drying, classification and bagging. No changes or modifications are required for the upcoming operation and minor maintenance and refurbishment is required.
4. Tailings Dewatering and dry Tailings and Waste rock storage: Flotation tailings will be filtered directly through 3 plate and frame filter presses. Test work indicated that no pre-thickening is required to achieve the targeted throughput and filter cake moisture content. The filter cake will be of sufficient low humidity to allow for dry tailings disposal. The selected tailings disposal site is of sufficient size and located close enough to the mine to allow for co-disposal of dry tailings and mining waste rock. The IWD will be lined with a HDPE liner system. In addition it will be constructed of the Swakop marble band which has low permeability and has acid neutralizing capacity.

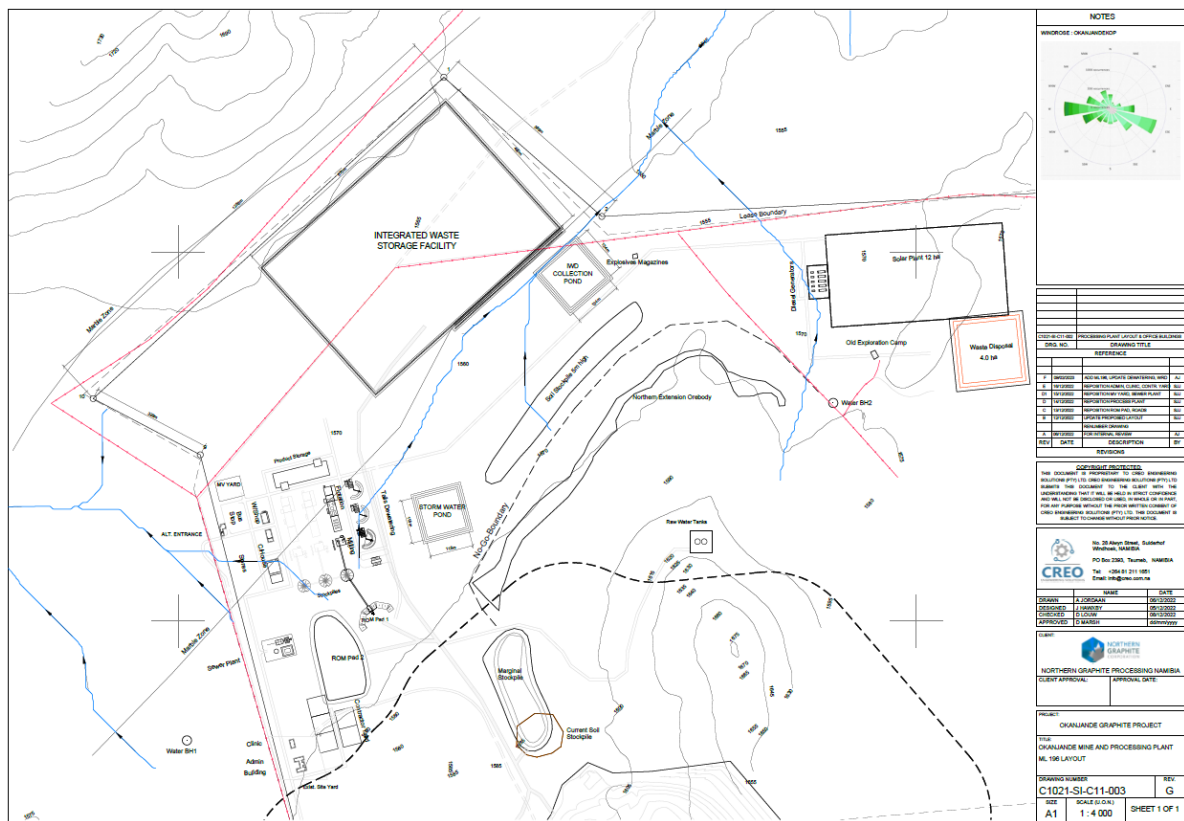


Figure 17-2: General Layout - Okanjande mine and processing plant

Table 17-1: Process Design Criteria

Project Fundamentals		
Design Throughput Factor	%	15%
Mean Annual Throughput	t/a	631,450
Operating Schedule Crushing		
Hours per Shift	h	12
Shifts per day	s/d	1
Days per Week	d/w	7
Operating Days per Year	d/a	365
Availability	%	92%
Utilization	%	90%
Effective Utilization	%	82.8%
Operating Hours per Year	h/a	3627
Nominal Feed Rate	t/h	174.1
Design Feed Rate	t/h	200

Operating Schedule: Mill and Concentrator		
Hours per Shift	h	8
Shifts per day	s/d	3
Days per Week	d/w	7
Operating Days per Year	d/a	365
Availability	%	92%
Utilization	%	90%
Effective Utilization	%	83%
Operating Hours per Year	h/a	7,253
Nominal Feed Rate	t/h	87.1
Design Feed Rate	t/h	100

ROM Physical Characterization		
Moisture		
Min	%	2%
Ave	%	4%
Max	%	7%
Material Handling		
Rock SG	t/m3	2.7
Bulk Density	t/m3	1.6
Angle of Repose	deg	35
Hardness		
JK Tech Dropweight Index	a x b	85.5
Uni-Axial Compressive Strength	Mpa	46.7
Abrasion Index	Ai	0.2
BCWi		12
BBMi	kWh/t	11.6
BRMi	kWh/t	12.1
Particle Size		
ROM Top size	mm	600
ROM F80	mm	300-400
Head Grade		
ROM Head Grade	%TGC	5%
Primary Crushing		
Primary Crusher P100 – Max	mm	180
Primary Crusher P100 – Nominal	mm	150
Primary Crusher P80 – Nominal	mm	120
SAG Mill		
Nr of Mills	#	1
Drive		Variable Speed
Fresh Mill Feed – Nominal	t/h	87.1
Fresh Mill Feed – Design	t/h	100.1
Design Mill Feed Top size	mm	180
Design Mill feed F80	mm	120
Design Mill Product P80 (estimate)	µm	3000
Mill Discharge Solids Concentration (estimate)	% m/m	65%

Rod Mill		
Nr of Mills	#	1
Drive		Soft Start
Fresh Mill Feed – Nominal	t/h	87.1
Fresh Mill Feed – Design	t/h	100.1
Design Mill Feed Top size	µm	4000
Design Mill feed F80 (estimate)	µm	3000
Design Mill Product P80 (estimate)	µm	325
Mill Discharge Solids Concentration (estimate)	% m/m	65%
Mill Circulation Load (estimate)	%	250%
Flotation		
Mean Combined Recovery of TGC	%	92%
Mean Combined Concentrate Grade	% C	96%
Mean Combined Concentrate Production Rate (dry)	t/a	30,257
Flake Distribution		
+300µm	% m/m	9%
+180µm	% m/m	40%
+106µm	% m/m	30%
-106µm	% m/m	21%

17.2.2 Process Description

17.2.2.1 Front-End Comminution

The Comminution circuit will consist of a single crushing stage followed by an open circuit SAG mill and closed-circuit Rod Mill with hydro-cyclone classification.

The primary crushing stage will be fulfilled by contractor crushing at the Okanjande mine site. The crushing circuit will consist of a modular jaw crusher with a hopper feeder and stockpile feed conveyor fitted with a tramp magnet. Crushed mineralized material will be conveyed to a crushed mineralized material stockpile of 2400t total capacity. Crushed mineralized material will be fed onto the mill feed conveyor via 2 drawdown feeders. The mill feed conveyor will be fitted with a tramp magnet and belt scale.

The SAG mill will be of the grate discharge type and be fitted with a trommel screen to discharge mill scats and oversize. The trommel screen oversize will be discharged into a bunker for removal by skid-steer/front-end loader. The trommel screen undersize will feed into a common mill sump which will feed the hydro-cyclone classifier.

The Hydro-cyclone underflow will feed directly into the Rod Mill. The Rod mill will discharge into the common mill sump. The hydro-cyclone overflow will feed the Low Intensity Magnetic Drum Separator (LIMS). Magnetics from the LIMS will be discharged into a sump for removal by skid-steer for disposal on the integrated waste dump. Non-magnetics from the LIMS will feed the Flotation Feed buffer tank.

17.2.2.2 Flotation

The Flotation circuit consists of the existing flotation banks transferred to the new Okanjande processing site. Ground ROM mineralized material at 30% solids will be fed to conditioning tanks where Diesel (collector) and MIBC (frothing agent) will be dosed. Conditioning will be followed by a Rougher stage with rougher tails pumped to a rougher scavenger stage. Tails from all downstream cleaner stages will be treated through a bank of cleaner scavenger cells.

Rougher concentrate will be polished through a newly installed 2.6'x7.5' wet scrubber (Polishing #1) with smooth rubber liners and ceramic grinding media. Rougher concentrate polishing will be followed by a single cleaning stage which will be followed by a polishing in repurposed 6.5'x9.0' ball mill (Polishing #2) fitted with smooth rubber liners and ceramic grinding media. The repurposed 6.5'x9.0' polishing mill will be sufficient for the current processing capacity, however it will need to be replaced with a fit-for-purpose polishing unit for any future capacity increases. Polished concentrate will be cleaned again through another two stages of cleaners with internal recycle.

Cleaned concentrate will be classified into a coarse (>180µm) and fine (<180µm) fraction of two stages of SWECO screens. Both fractions will be polished through vertical stirred media mills followed by two and three further cleaning stages with internal recycles for the coarse and fine concentrate fractions respectively.

Polished and cleaned coarse and fine concentrate streams will be combined and pumped forward to the Concentrate Thickener.

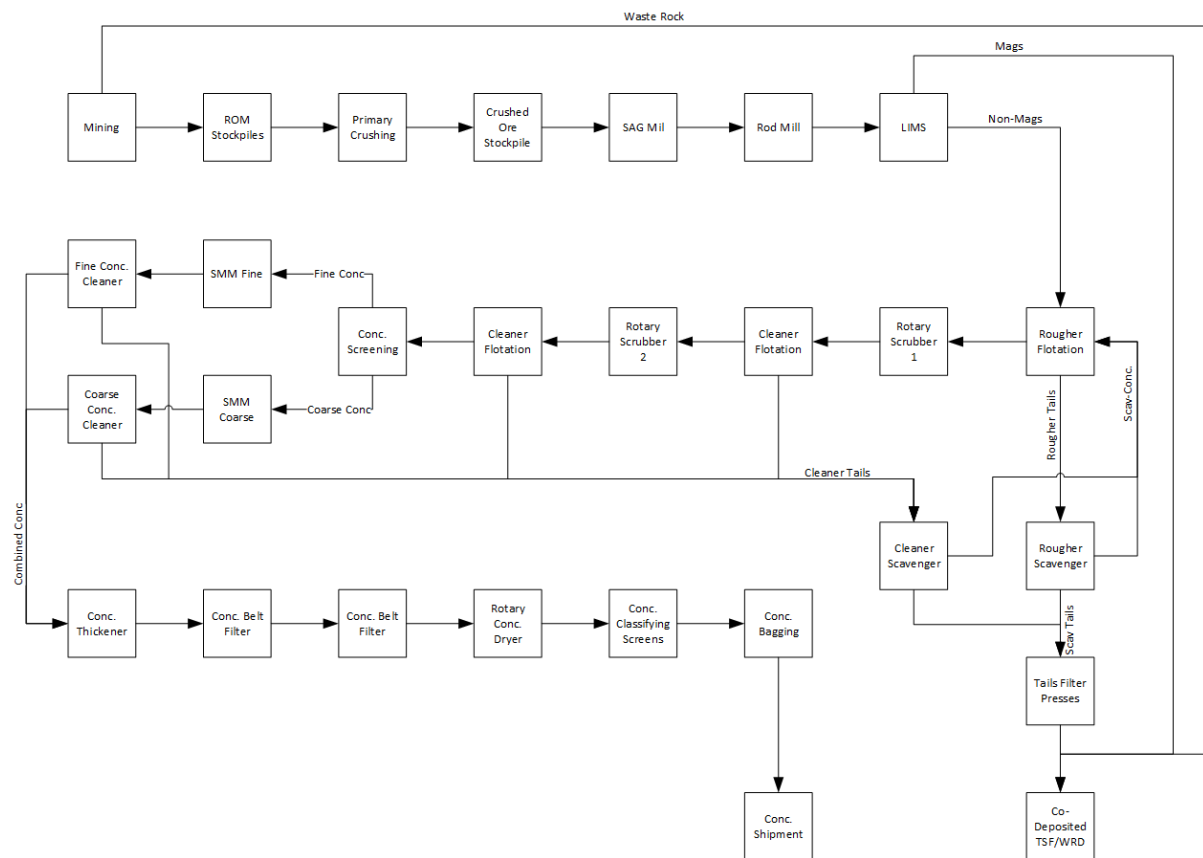


Figure 17-3: Okanjande Process Block Flow Diagram

17.2.2.3 Concentrate Dewatering, Classification and Bagging

The Concentrate Dewatering, Classification and Bagging section of the Okorusu plant was constructed by Imerys-Gecko and is fit-for-purpose for the described duties. No modifications are planned for this section of the plant, other than the relocation to the Okanjande site and only minor maintenance will be required since it was only constructed in 2016/17.

Combined concentrate size fractions will be received from the flotation circuit into the concentrate thickener, where the concentrate will be thickened before being pumped to the vacuum belt filter for further dewatering. The vacuum belt filter is fitted with dust extraction to recover any graphite dust generated.

Graphite filter cake is discharged into a rotary drum dryer which is diesel fired. The rotary drum dryer is fitted with dust extraction to recover any generated graphite dust.

Dried graphite concentrate is discharged from the drum drier and pneumatically transported to storage bins from where it is passed over classifying screens for separation into the following fractions:

- +300 μ m (ultra-coarse)
- -300 to +180 μ m (coarse)
- -180 to +106 μ m (middling's)
- -106 μ m (fines)

The different size fractions are stored in product bins each with dedicated bagging stations for bagging the various concentrate size fractions into 1t bulk bags for shipment.

17.2.2.4 Integrated Waste Dump (IWD)

Rougher and Cleaner Scavenger tails will be pumped to the tailings dewatering section where the tails will be filtered through plate and frame filter presses before being discharged on a dry-tailings stockpile. Filtered tailings will be loaded, hauled and dumped by the contract mining fleet, together with waste rock from mining operations, and co-deposited on the IWD facility.

The Okanjande integrated waste dump will be located on the north-western edge of the mining license above the band of the Swakop marble group. The integrated waste dump will be of the dry deposition type. Filtered tailings will be co-deposited with waste rock on the facility.

Geochemical testing of tailings and waste rock indicated that all weathered tailings and waste rock are potentially acid generating and that all transitional and fresh tailings and waste rock are acid generating. For this reason, the IWD will be lined with a double HDPE liner with leak detection. Run-off water from the IWD is channelled to a lined collection pond, where run-off may be neutralized and recovered to the process.

The IWD will overlie the underlying bank of Swakop group marble, which has very low permeability and in addition has acid neutralizing capacity. This adds additional security and will prevent any contamination of ground water and is in line with the 2014 EIA findings.

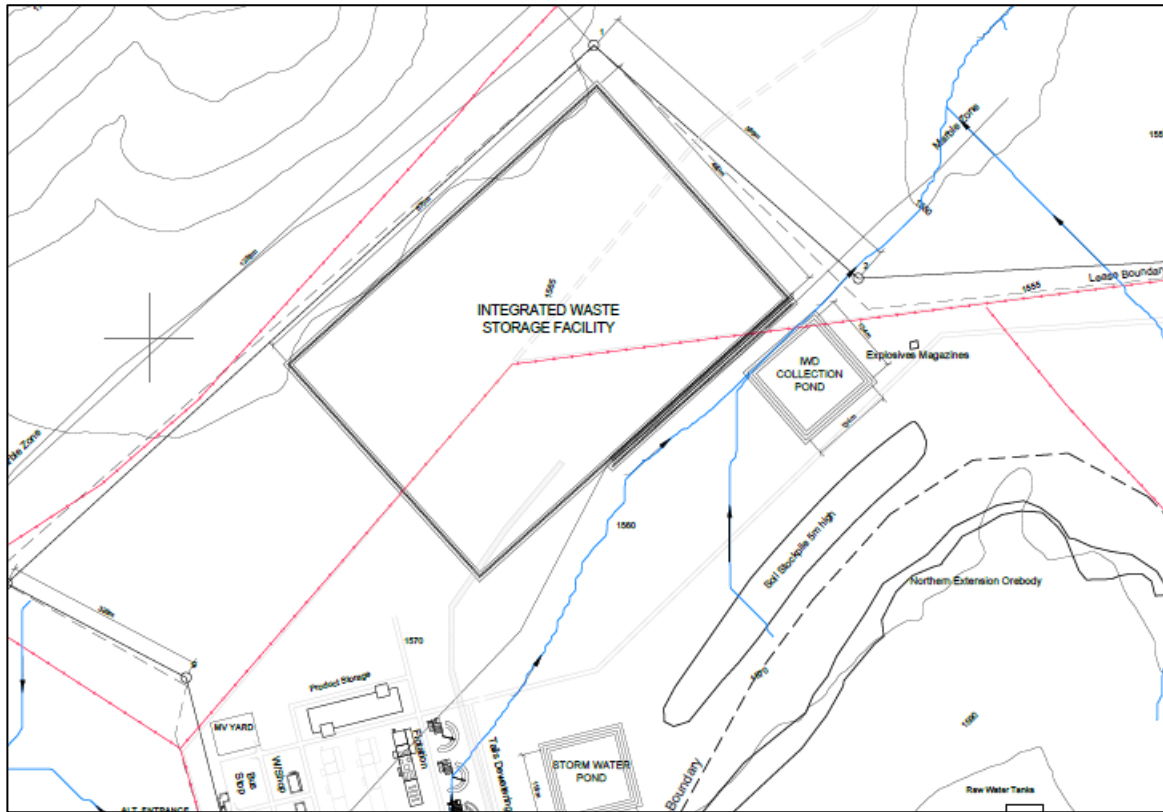


Figure 17-4: Okanjande Co-Deposition IWD

17.2.3 Mass Balance

17.2.3.1 Mineralized material, Graphite, Concentrate, Tails

The high-level mass balance over the graphite flotation process is given in Table 17-2. A comprehensive mass is provided in the Appendices.

Table 17-2: High Level Mass Balance

PROJECT KEY PERFORMANCE INDICATORS	UOM	Value
Throughput	t/h	87.1
Mean Head Grade	% TGC	5.3%
Total Graphite Concentrate produced	t/h	4.4
Concentrate Grade	%	95.0%
Total Contained Graphite in Concentrate	t/h	4.2
Total Tailings (dry) deposited	t/h	82.7
Recovery	%	92.0%

Table 17-3: Graphite Mass Balance (5.3% Head grade)

Product	Mass	Grade	Distribution
	%	% C(g)	% C(g)
+80 mesh 2 nd Clnr Conc	3.19	95.1	57.1
+80 mesh 1 st Clnr Tails	0.15	15.0	0.4
-80 mesh 3 rd Clnr Conc	2.00	95.1	35.8
-80 mesh 1 st Clnr Tails	0.25	30.0	1.4
4 th Clnr Tails	0.06	12.5	0.1
3 rd Clnr Tails	0.15	10.0	0.3
2 nd Clnr Tails	0.40	5.59	0.4
1 st Clnr Tails	3.50	2.10	1.4
Rougher Tails	90.3	0.18	3.1
Feed	100.0	5.3	100.0
Combined Products	Mass	Grade	Distribution
	%	% C(g)	% C(g)
Combined Concentrate	5.19	95.1	92.9
+80 mesh 2 nd Clnr Conc	3.19	95.1	57.1
-80 mesh 3 rd Clnr Conc	2.00	95.1	35.8
4 th Clnr Conc	5.59	90.0	94.7
3 rd Clnr Conc	5.65	89.2	94.9
2 nd Clnr Conc	5.80	87.2	95.1
1 st Clnr Conc	6.20	81.9	95.6
Rougher Conc	9.7	53.1	96.9
Feed	100.0	5.3	100
Combined Products	Mass	Grade	Distribution
	%	% C(g)	% C(g)
Combined Concentrate	5.19	95.1	92.9
+80 mesh 1 st Clnr Tails	0.15	15.0	0.4
-80 mesh 1 st Clnr Tails	0.25	30.0	1.4
2 nd Clnr Tails	0.61	7.35	0.8
1 st Clnr Tails	3.50	2.10	1.4
Rougher Tails	90.30	0.18	3.1
Feed	100.0	5.3	100

17.2.3.2 Water

Water will be abstracted from a bore field on the farm Doornlaagte situated some 24km from site. Run-off from the IWD will be collected, neutralized and recycled to the process. The water consumption rate was obtained from the site mass and water balance. The average anticipated annual water consumption is 105,390m³/a.

17.2.3.3 Reagents

Estimate process annual reagent consumption rates are given in Table 17-4. Dosage rates are based on values obtained from historical graphite processing at the Okorusu plant and the recent SGS test work program.

Table 17-4: Process Reagent Consumption

Reagents and Consumables	UOM	Annual Consumption
Diesel - Flotation	kg/a	57,461.95
Flocculent- Concentrate	kg/a	1,610.65
Water Treatment	kg/a	3,157.25
MIBC	kg/a	31,572.50
Diesel - Drying	kg/a	418,110.03
Filter Cloths	nr/a	85.00
Heat Shrink Bags	nr/a	31,103.13
Pallets	nr/a	31,103.13
Bulk Bags	nr/a	31,103.13

17.3 PRODUCTION PLAN

Production from the Okanjande mine will ramp up rapidly due to being a brown fields operation with all initial required overburden stripping completed during the previous operational period 2017-2018.

The run of mine mineralized material head grade increase with depth with the average grade exceeding 5% from year 7. A flat concentrate graphite grade was assigned to the production plan. However, the increased grade in the last 3 years of operation could likely lead to improved achieved concentrate grade and/or overall graphite recovery.

Table 17-5: Process Production Plan

Production Plan	Units	Totals	2023	2024	2025	2026	2027	2028	2029	2030	2031	2032	2033	2034	2035
Mining															
Mineralized Material Mined (dry)	t/a	6,096,950	0	205,221	520,946	632,315	632,315	631,450	631,450	632,315	632,315	631,450	631,450	315,725	
Grade Mined	%	5.27%	0.00%	5.58%	5.20%	4.88%	4.93%	4.90%	5.06%	5.26%	5.55%	5.56%	5.67%	5.86%	
Waste Mined (dry)	t/a	4,045,040	0	504,740	766,020	802,507	926,420	508,728	150,329	79,698	91,027	104,684	88,725	22,165	
Total Mined (dry)	t/a	10,141,990	0	709,961	1,286,966	1,434,821	1,558,734	1,140,178	781,778	712,013	723,342	736,134	720,175	337,890	
Ore Stockpiles															
Opening															
Mineralized Material Tons	t		0	0	0	0	865	1,729	1,729	1,728	2,593	3,458	3,458	3,458	0
Incoming															
Mineralized Material Tons	t	6,096,950	0	205,221	520,946	632,315	632,315	631,450	631,450	632,315	632,315	631,450	631,450	315,725	0
Outgoing															
Mineralized Material Tons	t	6,096,950	0	205,221	520,946	631,450	631,450	631,450	631,450	631,450	631,450	631,450	631,450	319,183	0
Closing															
Mineralized Material Tons	t		0	0	0	865	1,729	1,729	1,728	2,593	3,458	3,458	3,458	0	0
Processing															
Mineralized Material Processed	t	6,096,950	0	205,221	520,946	631,450	631,450	631,450	631,450	631,450	631,450	631,450	631,450	319,183	0
Grade Processed	%	5.3%	0.0%	5.6%	5.2%	4.9%	4.9%	4.9%	5.1%	5.3%	5.6%	5.6%	5.7%	5.9%	0.0%
Graphite Processed	t	321,174	0	11,451	27,080	30,814	31,129	30,927	31,917	33,212	35,069	35,108	35,768	18,697	0
Concentrate Produced	t	311,031	0	11,090	26,225	29,841	30,146	29,951	30,909	32,164	33,961	34,000	34,639	18,107	0
Tailings															
IWD Deposition (Moist)	t	10,837,335	0	748,021	1,364,293	1,525,583	1,654,312	1,218,496	844,075	770,020	779,779	793,032	775,681	364,043	0

17.4 MAJOR EQUIPMENT LIST

Table 17-6: Okanjande Process Plant Major Equipment List

Equipment	Description	Dimensions	Installed Power (kW)	Condition
Comminution				
SAG Mill		5.5x1.8m	1250	New
Rod Mill		3.0x4.0m	630	New
Hydro-Cyclone		TBD	NA	New
LIMS		900x900		New
Buffer Tank		50m ³	7.5	New
Flotation				
Rougher	Rougher #1&2	12x2.8m ³	266	Refurbished
Polishing Scrubber #1	Scrubber	2.6x7.5m	132	New
1 st Cleaner	Cleaner #1	6x1.4m ³	45.5	Refurbished
Polishing Scrubber #2	Mill	1.9x2.7m	228	Refurbished
2 nd Cleaner	Cleaner #6	6x1.4m ³	45.5	Refurbished
3 rd Cleaner	Cleaner #2	4x1.4m ³	30.5	Refurbished
180mm Classification Screen	SWECO 31, 32, 33		2.4	Refurbished
Dewatering Flotation	Scavenger #3	3x1.4m ³	22.5	Refurbished
Coarse Polishing Mill	Stirred Media Mill		55	New
1 st Coarse Cleaner	Cleaner #3	2x1.4m ³	15	Refurbished
2 nd Coarse Cleaner	Cleaner #4	2x1.4m ³	15	Refurbished
Fine polishing Mill	Stirred Media Mill		55	New
1 st Fine Cleaner	Cleaner #7	4x1.4m ³	30.5	Refurbished
2 nd Fine Cleaner	Cleaner #5	2x1.4m ³	15	Refurbished
3 rd Fine Cleaner	Cleaner #8	2x1.4m ³	15	Refurbished
Dewatering, Classification and Bagging				
Thickener			2.2	Existing
Belt Filter	Vacuum		7.5	Existing
Dryer	Direct Fired		70	Existing
Vacuum Transfer			18.5	Existing
Filter Extraction Fan			75	Existing
Sizing Screens			11	Existing
Product Silos	4 classifications			Existing
Screens Extraction Fan			37	Existing
Bulk Bagging Stations	4 classifications		48	Existing
Tailings Dewatering				
Filterpress Feed Tank		150m ³	45	New
Filterpress feed pumps (x3)			110	New
Filtrate pumps			150	New
Filter presses (x3)				New

Equipment	Description	Dimensions	Installed Power (kW)	Condition
Filtrate Tank		100m3		New
Filterpress transfer conveyors (x3)			11	New
Filterpress discharge conveyor			45	New
Dry Tailings stockpile floor				New
Dry tailings stockpile sump				New
Compressor			500	New

17.5 PROCESS PLANT LAYOUT

The existing Okoruso processing plant including flotation, concentrate dewatering, drying and bagging will be relocated to the Okanjande site. The original layout will be maintained at the new Okanjande site as far as reasonably possible. The overall process plant is aligned in a south to north orientation.

ROM pad 1 will be located north of the pit and the crushed mineralized material stockpile north of that. The milling circuit is followed by the flotation, concentrate dewatering, drying and bagging areas, all in a south to north orientation. The product storage area is on the far north side of the process plant. The tailings dewatering plant is located on the eastern side of the plant, allowing free access to the mining fleet to load and haul dry filtered tailing to the IWD. The IWD is located north-east of the processing facility.

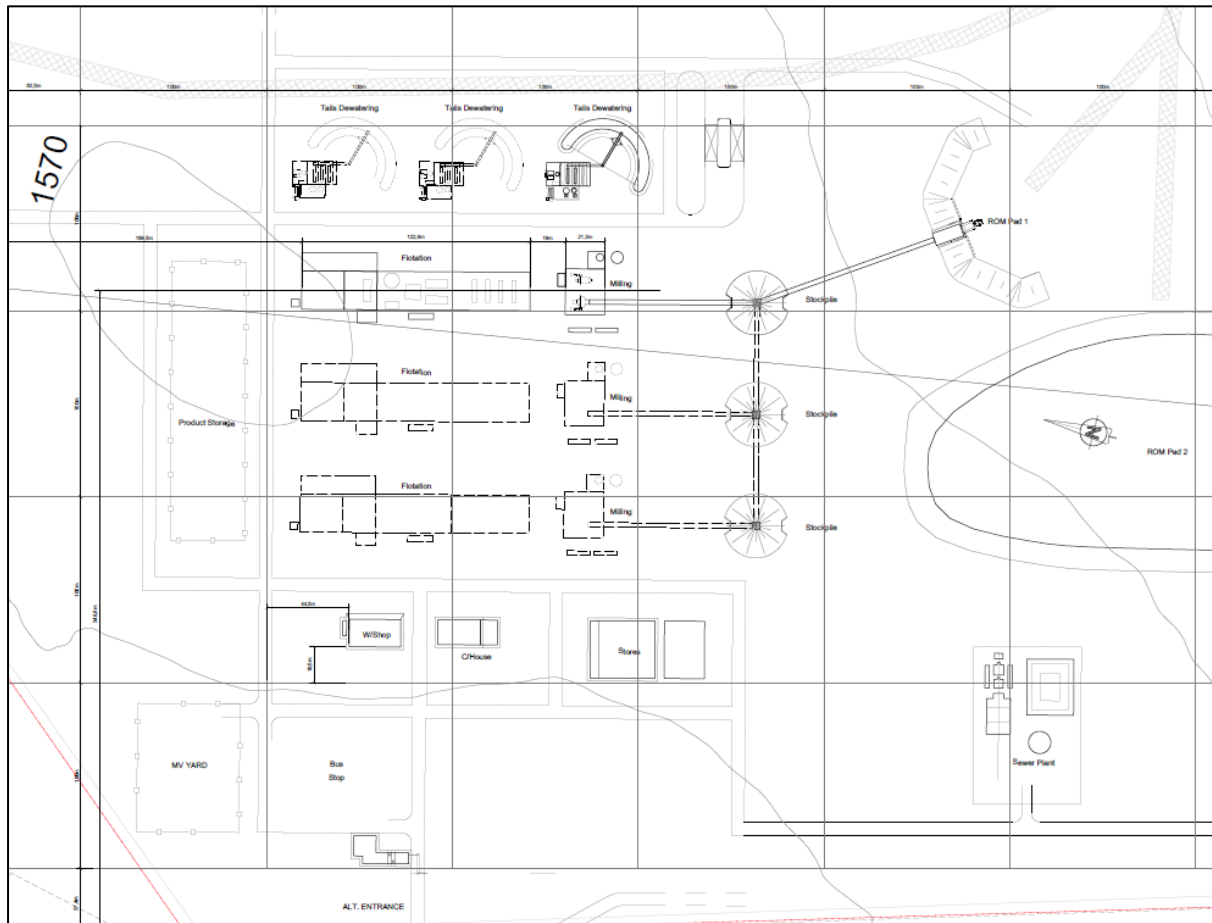


Figure 17-5: Okanjande Process Plant Layout

18 PROJECT INFRASTRUCTURE

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

18.1 ACCESS ROADS AND HAUL ROADS

The Okanjande deposit is accessed from the B1, 8km south of Otjiwarongo, onto the D2512 (a well-maintained gravel road) for some 24 km to the entrance of the property. Access for operational and logistical purposes is possible with ease along this route if road maintenance is executed during operations. Previous mining activities at the Okanjande deposit are evident and the established haul roads can be utilised as soon as operations continue. Figure 18-1 illustrates the access to the Okanjande project site from the B1 main road.

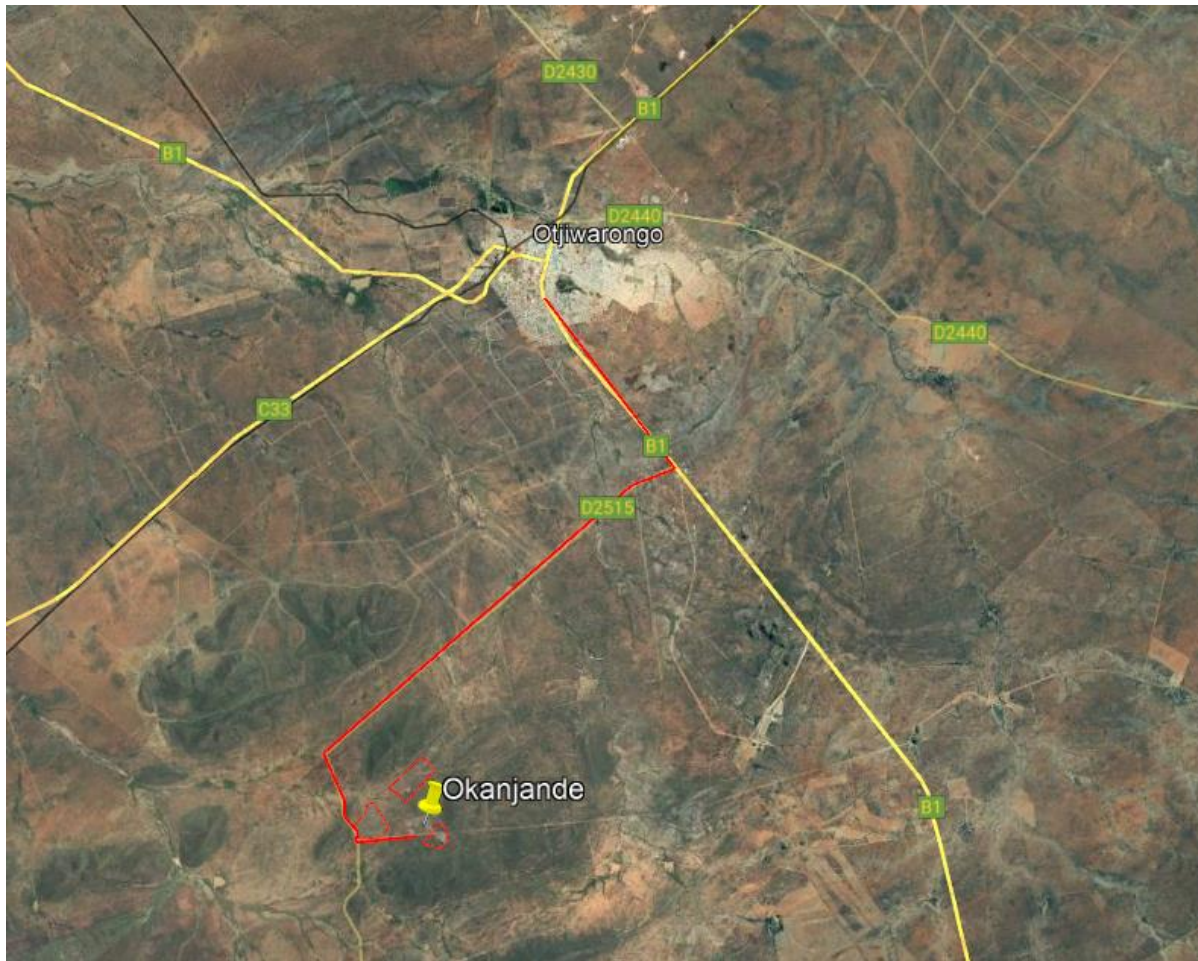


Figure 18-1: Okanjande access route

18.2 LOCAL RESOURCES

Local resources required for labour, supplies and equipment will be insufficient from the nearby community to meet all the project requirements. Some local resources can be utilised, but skilled mining personal, consultants and suppliers will be available throughout Namibia to support the project and its operations.

During operations, Okanjande will have to source some of its key personnel from outside of the immediate area, Otjiwarongo and possibly from the neighbouring country, South Africa.

18.3 BUILDINGS

The mining operations infrastructure located at the Okanjande deposit has adequate and well-maintained operating facilities sufficient for supporting operational restart.

The following new building structures will be erected at the Okanjande site:

- Administration offices
- Operational office
- Warehouse
- Change house
- Site ablution facilities

- Maintenance Workshop
- Mobile equipment workshop

18.4 SURFACE WATER DRAINAGE AND COLLECTION

The Okanjande project is located between two hills on a relatively flat plain. The process plant is located on a local saddle structure with natural drainage away from the process plant via 2 streams draining towards the north-west and north-east. See Figure 18-2. Where possible all surface run-off will be directed to the two natural drainage systems with berms and channels. The dimensions for each channel will vary based on the peak flow estimates. Where required geo-textile and stone pitching / riprap and gabions will be used to minimize soil erosion.

Surface run-off water from the process plant area will be directed to the return water dam located directly east of the process plant.

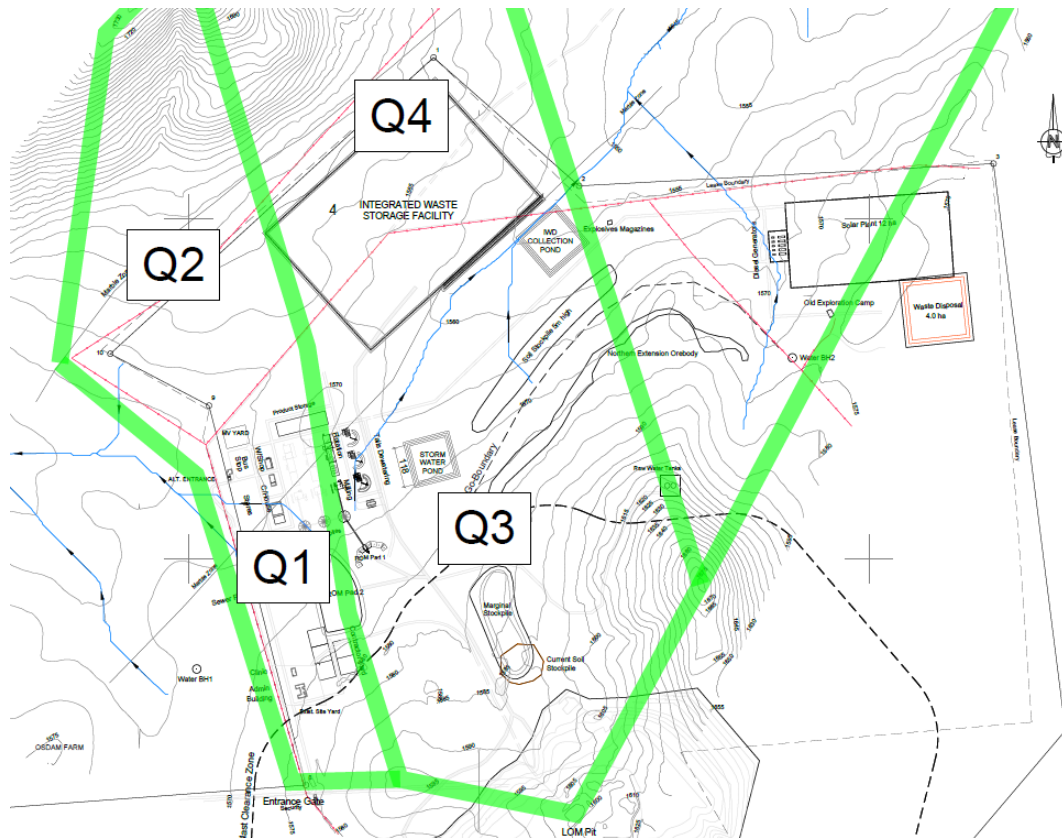


Figure 18-2: Okanjande Site Drainage

18.5 SEWERAGE DISPOSAL – OKANJANDE

A modular small trickling filter sewerage treatment plant (18.75m³/d) will be installed to accommodate for 50 persons and assuming a peak factor of 2.5.

18.6 SECURITY

The Okanjande property is adequately fenced off. The existing access gate will be upgraded to accommodate increased traffic and to implement electronic access control.

18.7 WATER SUPPLY

The water supply will be obtained from two boreholes on Farm Doornlaagte. The boreholes will be installed with two new submersible borehole pumps which will pump water to a new 150m³ reservoir. From the reservoir a base pump station (two centrifugal pumps in a duty plus one standby configuration) will pump the water to Okanjande via a new underground OD 110mm O-PVC pipeline. The water will be stored in newly established reservoirs.

The reservoirs will be installed at such an elevation to provide sufficient static pressure to ensure all areas of the site can be fed from the reservoirs.

A potable water treatment plant has been allowed for with a capacity to supply 50 people on site at. The plant will be provided with a booster pump/s to ensure sufficient pressure is available.

18.8 COMMUNICATIONS

The communications infrastructure will consist of a communications tower connecting the local network to the site as well as housing repeater infrastructure for radio communications on site. The incoming communications network will be distributed internally to the different users for an established office network providing internet connectivity and telephone communication. The internal distribution will contain a combination of fibre optic and copper networks.

18.9 POWER SUPPLY

No utility power is currently available at the Okanjande property. The power infrastructure will consist of relocating the existing 22kV distribution network in service at the Okoruso processing facility. This includes transformers and medium voltage switchgear. New internal distribution reticulation will be constructed at Okanjande to the buildings and motor control centres.

Power supply will be supplied in two phases, initially from an independent power producer combining renewable (Solar and Battery) and thermal power; followed by a combination of grid power and solar power after 4 years when the grid connection is completed. The plant will be located on the Okanjande site within the existing mining licence boundary.

Grid power from the local supply authority, Nampower, is anticipated to be connected approximately 4 years thereafter which will result in the power supply mix to be changed to a combination between grid and renewables, with thermal power only as a backup solution. Provision has been included for the Grid Supply CAPEX which includes upgrading the existing Nampower owned Otjiwarongo substation and the construction of a new 66kV transmission line, approximately 17km to the Okanjande plant.

18.10 FUEL

There are limited fuel storage facilities available on the Okanjande mine site. Diesel storage is available for the site generator powering the administration buildings and workshop. Mining fleet fuel infrastructure will be provided by the Mining Contractor.

18.11 CAMP & CANTEEN

No room or board facilities will be built for this project and no site located facilities are available to be utilised for operational personnel or construction crew. No canteen facility will be provided for operational or administrative staff.

It is recommended to accommodate staff in Otjiwarongo and set up transport to and from site for the Okanjande site.

18.12 FIRE PROTECTION

The Okanjande fire protection system consists of the following elements:

- Fire Water Header Tank
- Diesel Driven Booster Pump Station
- Firefighting Water Reticulation
- Firefighting Equipment (Fire Hydrants, Hose Reels, Fire Extinguishers)
- Smoke Ventilation (Flotation Plant, Workshops, Warehouse)

19 MARKET STUDIES AND CONTRACTS

(Prepared by CREO Engineering Solutions (Pty) Ltd, based on information provided by Northern Graphite Corporation)

Graphite is one of only two naturally occurring forms of pure carbon, the other being diamonds. Diamonds have a three-dimensional crystal structure whereas graphite consists of a “two dimensional”, planar molecular structure. For this reason, graphite generally occurs as flakes, which are multiple layers of “graphene” held together by weak bonds. Graphene is a single, one atom thick layer of carbon atoms arranged in a “honeycomb” or “chicken wire” pattern. It has been estimated that there are three million layers of graphene in a one-millimetre thickness of graphite. The delamination or exfoliation of graphite flakes is therefore one method of producing graphene.

Graphite is formed by the metamorphism of carbon rich materials which leads to the formation of either crystalline flake graphite, fine grained amorphous graphite, or crystalline vein or lump graphite. Graphite is a non-metal but has many properties of metals and is desirable as a strong, light weight, heat and corrosion resistant reinforcement material with a high aspect ratio. It also has high thermal and electrical conductivity, chemical inertness, and a natural lubricity.

Because of supply concerns relating to the fact that China is the world’s dominant producer, and to potential demand growth from new applications such as lithium-ion batteries, the European Union considers graphite to be one of 14 “critical mineral raw materials” with high supply risk. The United States has also included graphite on a list of mineral resources whose loss could critically impact the public health, economic security and/or national and homeland security of the United States.

There is very little recycling of, or substitution for, graphite.

19.1 GRAPHITE SUPPLY

Benchmark Mineral Intelligence (“Benchmark”) estimates that worldwide production of natural flake graphite in 2021 will be just over 1,000,000t of which China will produce approximately 75%. Brazil (8%) and Mozambique (6%) make up most of the balance with the rest largely scattered among Madagascar, Tanzania and Europe.

Historically, most Chinese flake graphite production came from Shandong province. However, production has declined as mines get older and deeper, costs rise, resources are depleted, and environmental regulations have become increasingly strict. The majority of Chinese production now comes from Heilongjiang province where deposits are generally higher grade but smaller flake and lower quality. As a result, Chinese production of the larger flake sizes is declining. Heilongjiang has substantial small flake resources and excess production capacity, most of which is targeting the lithium-ion battery market.

At the 14th ICCSINO Anode Material and Feedstock Market Conference held in Qingdao China in May, 2021 it was reported that production from the two main areas in China was 731,000t in 2020. Including output from other provinces that would put Chinese flake graphite production at 850,000t which indicates that it is being under-estimated by western commentators including Benchmark, Roskill and Fastmarkets who put it at about 750,000tpa.

Surplus Chinese supply had largely related to a large mine in Loubei which is owned by Minmetals, a state-owned company that is one of the largest metals and minerals trading companies in the world. The mine had only been operating about four months per year and producing about 300,000tpa of flake graphite. Based on the ICCSINO

figures it is likely the Loubei mine is now near full production. Minmetals, which is the largest Chinese natural graphite producer, had forecast a supply deficit in 2020, which did not happen due to Covid, and a very large supply deficit by 2025.

A large new graphite mine was built in Mozambique in 2018 and it substantially increased world natural flake graphite supply. The mine was significantly over budget, has only operated at about 50 percent of nameplate capacity and has been cash flow negative since start up. Its future is uncertain as western buyers do not want to switch to a supplier that may not be able to continue operations. Most of its production has been sold into the Chinese steel market as the Chinese do not find its production suitable for the battery market.

Because of potential demand from the rapidly growing EV/LiB markets, there are over 20 advanced stage graphite projects outside of China that have completed feasibility studies. Almost all are looking for financing and almost none are under construction. Most are in Tanzania, Mozambique, Australia and Canada. They represent a significant source of future supply, but each will take up to two years to bring into production after they obtain financing. Also, not all are battery grade and many come with substantial political risk.

19.2 GRAPHITE DEMAND

Historically the single biggest use of graphite has been for the manufacture of refractories in the steel industry which accounts for over 40% of demand or 400,000t. These are essentially fire bricks which line and protect blast furnaces. They contain 10-25% graphite which acts as a lightweight reinforcement that is resistant to heat and corrosion. Accordingly, graphite is a consumable in the steel making process, not an alloy, although graphite is sometimes added to steel to increase the carbon content.

BMI reported that the battery anode market experienced significant growth in 2022, with supply of natural graphite under pressure. This supply-demand imbalance caused graphite prices to rise, and this trend is predicted to continue through 2023.

The demand for graphite anodes grew by 46% in 2022 while flake graphite supply only grew by 14% during the same period. The biggest consumer of natural flake graphite was the lithium-ion battery anode market during 2022. With this increase of consumption by the lithium-ion battery industry it has overtaken the more traditional markets such as refractory and foundry industries. The lithium-ion battery market will increasingly guide the graphite market fundamentals and support price increases.

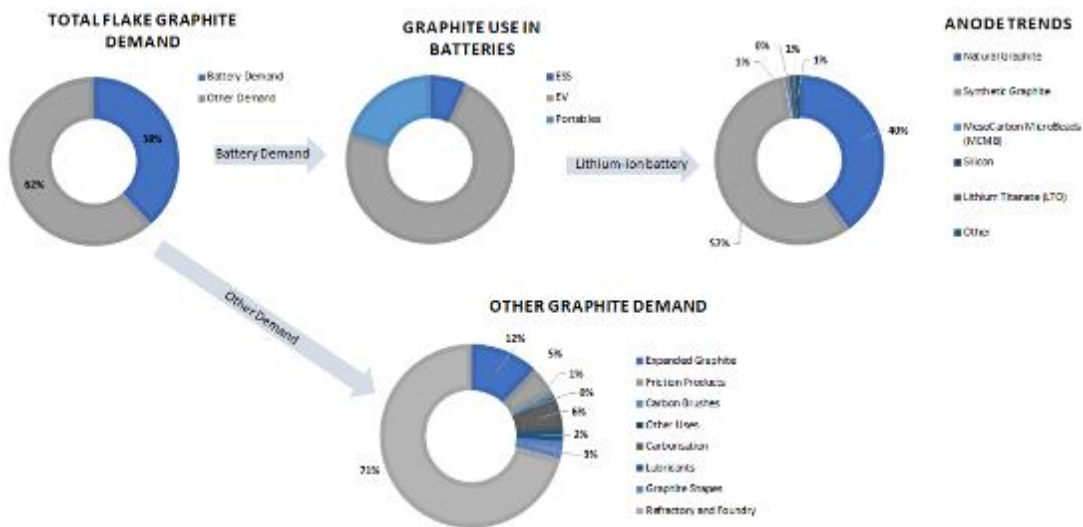
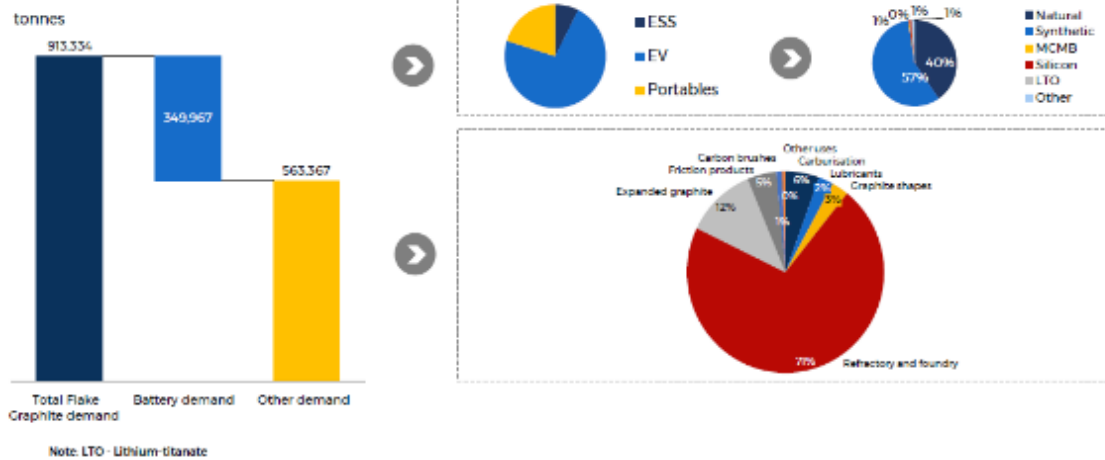


Figure 19-1: Flake Graphite Demand Breakdown, by end-use 2021 (BMI, 2021)

Graphite is the anode material in lithium-ion batteries and is their largest single component. There are no substitutes in this application although the anode material can be either synthetic or natural graphite. Natural graphite is less expensive and has a higher capacity whereas synthetic graphite charges faster and has a longer cycle life. Natural graphite-based anode material is called spherical graphite (“SPG”) and is manufactured from graphite mine concentrate through a process that involves micronization, rounding, purification, high temperature heat treatment and coating. Micronizing and rounding flake graphite concentrate provides an SPG yield of 35 to 40%. Rejects can be used in other applications.

In order to be used in the manufacture of SPG, mine concentrates should have a purity of 94% or higher, have a high bulk density, high crystallinity and have the ability to be rounded and purified. Not all mine concentrates, or even all concentrates from any one mine, will be “battery grade”.

In a relatively short period of time, lithium-ion batteries have gone from a very small market to the largest. The initial growth was largely due to small devices such as cell phones, cameras, laptops, power tools, etc. Hybrid and electric vehicles and grid storage are potentially huge markets that are still in their infancy and provide ample opportunity for continued strong growth in lithium-ion battery production and therefore the demand for

raw materials, including flake graphite. Continued annual growth of 20% means a big new 100,000tpa mine is required every year and none are currently under construction. Benchmark estimates that flake graphite production needs to more than double by 2025 to meet demand.

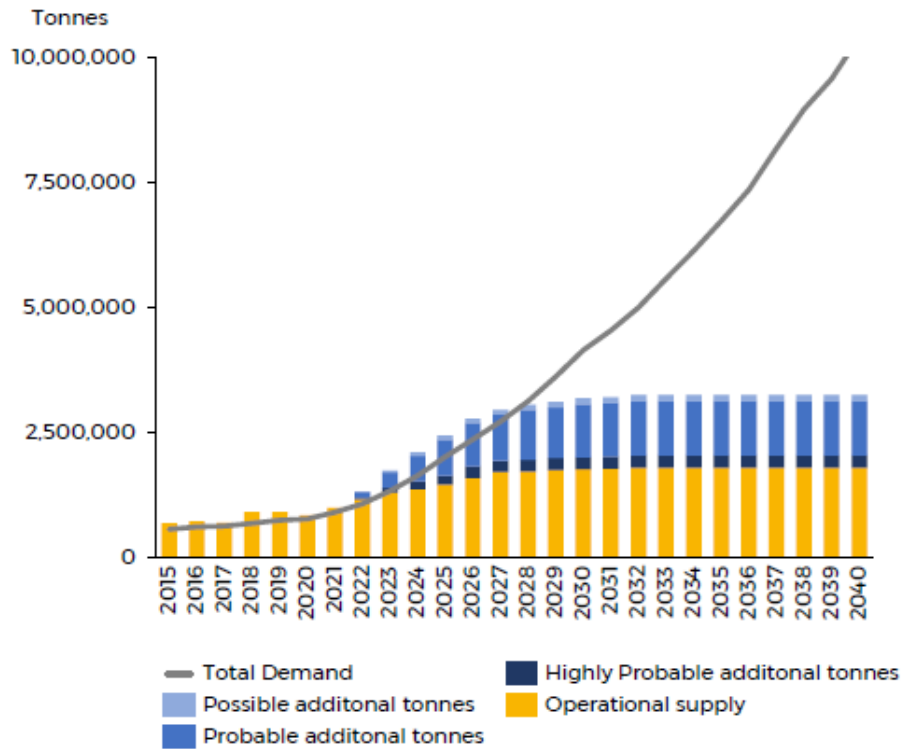


Figure 19-2: Graphite Demand, 2015 to F2040 (BMI, 2021)

Graphite also has a myriad of other uses including in the automobile industry (gaskets, brake linings and clutch parts), thermal management in consumer electronics, insulation products, electric motors (carbon brushes), heat and corrosion resistant gaskets, fuel cells, fire retardants, lubricants, pencils and many others. The “graphite” commonly used in golf clubs, hockey sticks, tennis rackets and composite materials is actually carbon fiber, a synthetic form of graphite made from petroleum coke.

These other uses make up about 160,000t of demand of which 100,000t are expandable graphite products. Expandable graphite is made by treating graphite concentrate, predominantly +50 mesh extra-large flake, with a dilute acid solution which intercalates between the many layers in each flake. When heated, the solution expands forcing the layers apart and increasing their volume by hundreds of times. The expanded graphite is then pressed into self-binding sheets and foils which are used in many of the products listed above. Expandable graphite is one of the fastest growing graphite markets along with lithium-ion batteries expandable graphite is also the only market segment that has reported price increases over the last few years.

19.3 GRAPHITE PRICES

There is no posted spot price or futures market for graphite. Sales are negotiated between producers and consumers of which there are many. Some have long standing relationships and negotiate prices and volumes annually but there are no long-term contracts or offtake agreements. A number of industry sources (including Industrial Minerals magazine and Benchmark) publish prices for the most popular grades and they provide a conservative indication of pricing for large, high-volume buyers. Generally, concentrates have to be above 150 mesh in size (which is considered small flake), have a carbon content higher than 94 percent, and not have any undesirable impurities in order to consistently attract buyers and achieve the best price.

Prices increase with flake size and carbon content. The premiums are relatively small going from +150 mesh (small) flake to +100 mesh (medium) and +80 mesh large flake. Prices are much higher for +50 mesh (XL) and +32 mesh (XXL) flake sizes as they are rarer and more valuable. Purity is also a factor but not a big one if the concentrates are above 94% carbon. Most mine concentrates, produced by flotation alone, range between 94 and 98% carbon. The difference in pricing is currently about 2.8% per 1% increase in the absolute level of purity. Other more technical factors can affect pricing including bulk density, expansion ratio, ash composition, crystallinity, volatile content and oxidation resistance.

Most mines also produce a substantial amount of -150 mesh material, which is micro flake and fines, and it usually has a carbon content less than 94%. This material presents a marketing challenge as it is not suitable for many of the major markets.

Graphite prices peaked in the \$1,300/t range for the premium grade (large flake +80 mesh, 94-97%C) in the late 1980s and then declined sharply as Chinese producers entered the market. This caused western mines to close and development projects to be shelved. Prices did not begin to recover until 2005 and peaked in a range of US\$2,500 to \$3,000/t in 2012 when some shortages were reported. This was due to the growth in China and the commodity super cycle and the resultant demand from the steel market. The subsequent slowdown in the Chinese economy combined with a lack of growth in the US/Japan/Europe caused prices to fall back approximately 50% from 2012 levels. In the second half of 2017 prices rebounded 30% to 40% due to a

recovering steel industry, continued strong growth in lithium-ion battery demand and environmentally related capacity shutdowns in China.

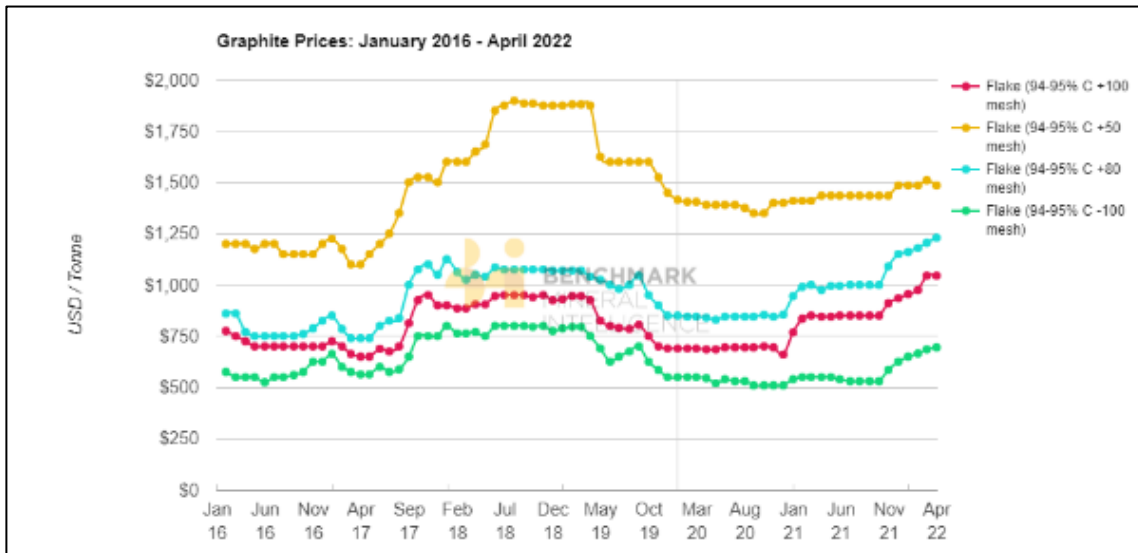


Figure 19-3: Recent Graphite Pricing Trends (BMI, 2022)

Northern’s sales from the second and third quarters of 2022 (See Table 19-1) from their Lac des Isles plant achieved ex-gate pricing of \$1,643 and \$1,572 respectively (Corporation, MANAGEMENT’S DISCUSSION AND ANALYSIS For the Three and Nine Month Periods Ended September 30, 2022 , 2022), (Corporation, MANAGEMENT’S DISCUSSION AND ANALYSIS For the Three and Six Month Periods Ended June 30, 2022, 2022). The concentrate produced from the Okanjande project will have a superior flake size distribution than that of the Lac Des Isles process and it would be reasonable to expect similar, if not higher, realised prices for this concentrate.

Table 19-1: Realized Graphite Pricing, Lac des Isles 2022

Graphite Pricing	UOM	2022 Q2	2022 Q3	Total
Sales	-			
Tonnes	mt	1,773.00	2,184.00	3,957.00
Average realized selling price per tonne	CAD	2,083.00	2,053.00	
FOREX	CAD:USD	1.2832	1.3056	
Average realized selling price per tonne	USD	1,623.29	1,572.46	1,595.23

Northern’s experience with the producing Lac des Isles mine in Quebec demonstrates that high quality concentrates from a secure, politically stable, non-Chinese source of supply can achieve substantial premiums

over published prices. Accordingly, a moderately conservative average price of US\$1,550/t has been used for the purposes of this PEA.

In the near to medium term graphite prices are expected to increase as lithium-ion battery demand is growing rapidly, supply from China is getting tighter, political and ESG concerns relating to China are increasing and freight markets are getting expensive and uncertain. Local supply chains are desired.

19.4 POTENTIAL PRODUCTS AND MARKETS

As indicated in section 13 Mineral Processing and Metallurgical Testing, Northern estimates that in practice it is possible to achieve a 20% increase in the coarser flake size due to market requirements of only 80% of the product being at spec, the balance being smaller flake. The expected product split figures provided by Northern are 11% +50 mesh XL flake; 48% +80 mesh large flake; 24% +100 mesh medium flake; and 17% -100 mesh small flake.

Northern expects the Okanjande deposit concentrates may be able to attract a premium by selling smaller volumes to multiple customers rather than selling larger volumes to a small number of traders and/or distributors. It may also realise a premium based on its high-quality ex-mine product, providing a secure, stable, non-Chinese source of supply and inventory management.

19.5 GRAPHITE MARKET OUTLOOK

In the near- to medium-term, graphite prices are expected to increase as lithium-ion battery demand is growing rapidly, supply from China is getting tighter, and political, environmental and social governance (“ESG”) concerns are increasing. The importance of a local and/or shorter supply chain has been highlighted by recent disruptions including the Suez Canal blockage, COVID-related shortages of labour resulting in long delivery times at ports, and a shortage of containers and container ships. All have combined to substantially increase transportation costs and make projects such as the Okanjande Project more attractive.

19.6 MATERIAL CONTRACTS

The Company has not yet entered into any material contracts. However, the Company has opened a sales office in Italy to initiate market development in Europe and is also planning on developing a battery anodes material plant in Canada. Any excess product from Namibia would be delivered to this plant for further processing and “value add”.

19.7 QUALIFIED PERSON STATEMENT

CREO has reviewed the market studies and pricing statements from Northern Graphite and confirms that the results support the use of the \$1,550/tonne used in this PEA Report.

20 ENVIRONMENTAL STUDIES, PERMITTING AND SOCIAL OR COMMUNITY IMPACT

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

20.1 PROJECT ENVIRONMENTAL PERMITTING REQUIREMENTS

20.1.1 Authorisation Status

A review of available documentation was undertaken to determine the authorisation status of the Okanjande site. Table 20-1 provides the authorisation history of the two sites respectively. Okanjande has received an Environmental Clearance Certificates (ECC) in terms of the Environmental Management Act (No. 7 of 2007) (EMA). The Okanjande Mine ECC allows for the construction, infrastructure development and operation of the graphite mine and the graphite processing plant comprising of crushing, milling, heavy mineral separation, flotation, filtration, drying, screening, bagging and tailings disposal. The Okanjande ECC for the development and graphite mining activities was first issued to the project on 23 April 2015.

Table 20-1: Environmental Authorisation Status Okanjande

Year	Consultant	Type of application	Status	Comments
2014	Enviro Dynamics	EIA		
2020	Imerys	Renewal application (with EAR)	EAR Submitted, ECC renewed and Valid until 2024-11-11	
2021	Geokey	Environmental Audit Report	Bi-Annual reporting up to date. EMP up to date	Requirement for Bi-annual internal audit and report

20.2 REGULATORY REQUIREMENTS

This section details the regulatory requirements, proposed public participation processes as well as recommended specialist studies for the project.

20.2.1 Environmental clearance certificate (ECC)

Although Okanjande currently has an ECC, it is recommended that any potential changes to the process be reconciled with the current ECC to determine whether amendments are required. This will first and foremost entail engagement with the Ministry of Environment, Forestry and Tourism (MEFT) as the regulating agency. The process to follow will entail a Scoping Study with impact assessments of the new and revised activities, and an updated EMP.

Table 20-2: Environmental Clearance Status

Project	License /Project	ECC expiry	ECC/MET online #
Okanjande	EPL4717 / OKA	20-Feb-25	ECC – 001996
Okanjande	ML196 / OKA	11-Nov-24	ECC – 01730

20.2.2 Water Abstraction Permits

20.2.2.1 Okanjande Mine Site

The water abstraction permit nr 11163 on the remaining portion of Farm Okanjande was issued 5 July 2021 to abstract 25,000m³/a for the use in mining and accessory works purposes. The permit is applicable to borehole nr. WW203557 and is valid for a period of 2 years.

Application for permit renewal was submitted to the Department of Water Affairs for future water supply to the Okanjande project.



Figure 20-1: Permitted Farm Okanjande Borehole siting

The water abstraction permit nr 10946 for boreholes on the Farm Doornlaagte was issued 13 July 2021 to abstract 90,000m³/a for the use in mining purposes. The permit is applicable to borehole nrs. WW201922 and WW201923 and is valid for a period of 2 years.

An application for renewal of the permit was submitted to the Department of Water Affairs on 10 February 2023 for future use of the bore for water supply to the Okanjande project.

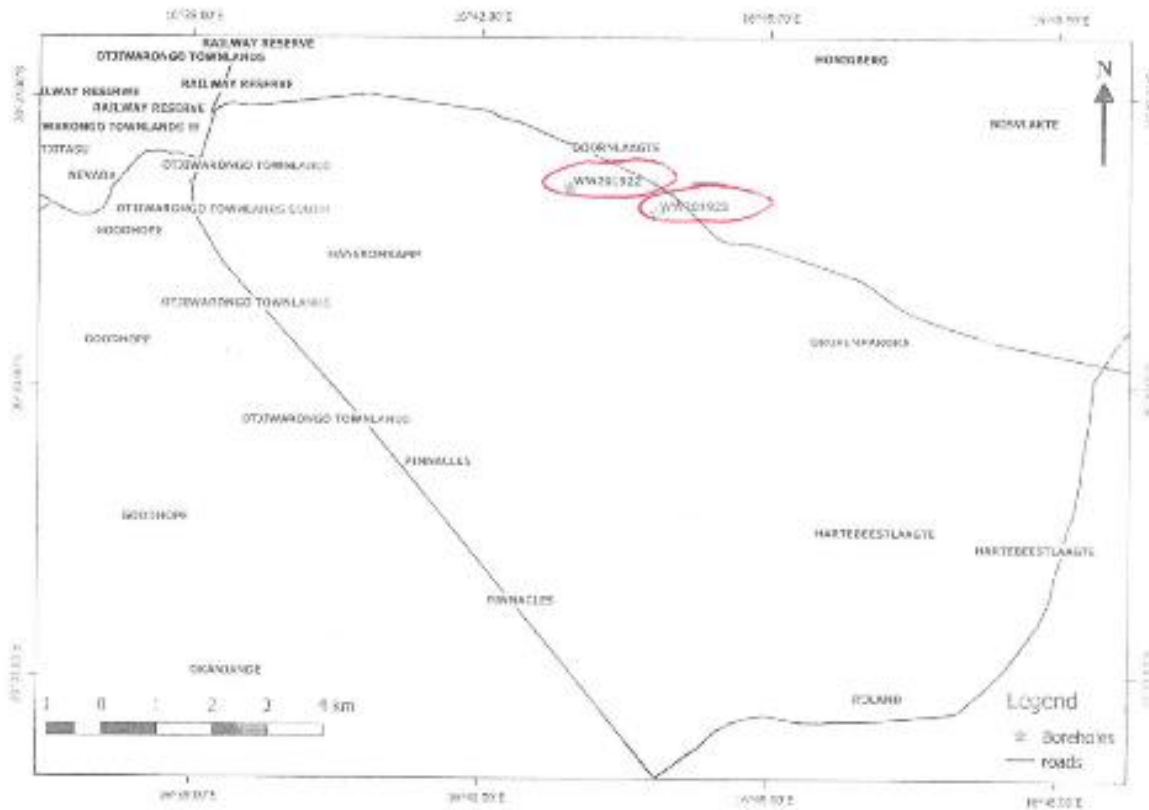


Figure 20-2: Permitted Farm Doornlaagte Borehole Siting

20.3 SOCIAL/COMMUNITY REQUIREMENTS

20.3.1 Public Participation

The company engages in public participation through the Environmental Assessment (EA) process.

Community consultation and stakeholder engagement was undertaken as parts of an ongoing EA process. Detailed in the Community Consultation (2013) and EIA reports (2014, 2022). In March 2023, the company Environam Consultants Trading (ECT) as part of an amendment study, submitted the latest draft of the revised Environmental Management Plan (EMP) for review in the public domain. The draft was also distributed to the Interested and Affected Parties as stakeholders in the process.

Since this component of the EA process (i.e. public consultation) is indispensable for the building of strong, constructive and responsive relationships, stakeholder engagement forms a continuing process. This is essential for the successful management of the project's social and environmental impacts. An ongoing process of public participation shall therefore be maintained to ensure that stakeholders are not overlooked and that that effective methods for communication are used. Through continued involvement of the communities and stakeholders in a meaningful way, the company strives to meet these objectives:

- To facilitate two-way engagement with stakeholders whereby relevant information on the project are provided in an accurate and timely manner;
- To identify and address issues that stakeholders may have with the project;

- To explore areas within the community where the project proponent can make a positive contribution that would lead to the upliftment of the community.

20.3.1.1 Project Stakeholders

During the public consultation and participation process various stakeholders have been identified that are likely to be adversely or positively affected by the project. Northern Graphite has undertaken to keep these stakeholders informed on its envisaged schedule. The main objective of public participation is to disseminate relevant information about the project and to gather information from stakeholders towards an environmentally and socially sound project.

From initiation of the project the company strived to ensure that stakeholders are not overlooked and that that effective methods for communication are used to identify the role players. Public participation activities were detailed as part of the EIA process.

A stakeholder database was developed as part of the 2020 amendment (Hook, 2020) and includes all interested and affected parties, Government Ministries, local and regional government councillors and key officers, Government Parastatals, affected landowners, etc. It is updated on a regular basis. The latest draft EIA and EMP (ECT, 2023) was made available to these interested and affected parties on 2 February 2023 and was available for comment until 16 March 2023.

20.3.2 Analysis of Social Issues

No social issues of concern have been identified at the time of writing of this report.

20.4 MINE CLOSURE

No Mine Closure plan has been developed at the time of issuing of this PEA. A Mine Closure Plan needs to be developed for the Okanjande Graphite project for the closure of Open Pits, Integrated Waste Dump, Process Plant Buildings and Infrastructure; as well as general surface rehabilitation.

The open pit mining at Okanjande lends itself to a number of risks which need to be considered during the development of closure plans. These risks were identified in 2014 Environmental Impact Assessment (Enviro Dynamics, 2014):

- *“If the mine pit remains unfilled after mining activity ceases it is likely to fill by groundwater seepage to a level in equilibrium with the surrounding groundwater table. The inflow rates are expected to be high when the pit is excavated but later flow are likely to be at low rates.*
- *This could lead to the formation of low pH, high sulphate and TDS water in the pit with possible dissolved metals that is concentrated by loss of water through evaporation. This will particularly apply to seepage from the pit walls above the water level and during seasonal fluctuations of flow and water levels.*
- *The deep water levels, low hydraulic conductivity, limited water in the dewatered tailings or deposited dry tailings and the addition of neutralising agents to the tailings could all work to ensure the security of the groundwater. Inflow rates into the mine pit are estimated to be low and could be neutralised if found to be acidic. If surface water is diverted away from the mine structures groundwater contamination through direct infiltration can be avoided. Namib Hydrosearch (2014) concluded that the general impact of the mine is seen to be limited and can further be reduced by implementing the*

recommended mitigation measures for possible impacts and by continuously monitoring the effectiveness of the mitigation measures during the operation and closure phases of the mine.

Risk Mitigation measures were recommended by Namib Hydrosearch for inclusion in a Mine Closure plan:

- *Berms should be constructed to channel runoff away from the entire north-western boundary of the TSF to avoid erosion and damage to the embankments and possible mixing with effluent in TSF perimeter trenches. The runoff should be discharged to the natural drainage channel in these directions. No runoff should enter the perimeter trench or reach the embankment of the TSF.*
- *Berms on the upslope side would stop runoff entering the stockpiles or mine pit given the relatively small volumes of runoff expected. The runoff is to be diverted away from the stockpiles towards the southwest flowing natural drainage. Contact or mixing of runoff with the stockpile material or effluent is to be avoided.*
- *The proximity to the contact zones along the north-west and south-east side require that the trenches are lined and any effluent collected and directed to the RWD. Sufficient depth of the perimeter trench is required to intercept any lateral flow along the soil zone.*
- *The Rock Waste Dump (RWD) needs to be lined with strip drains to prevent seepage. Steps will also have to be taken to minimise the retention time in the RWD and to pump this water as a priority for use in the plant. The RWD will be designed to accommodate storm water (1:100 rainfall event) and to contain surface runoff from the TSF. Dilution of the effluent to a significant level is expected under these conditions.*
- *Diversion of storm water drainage with berms and peripheral trenches to collect seepage water is recommended together with close monitoring. On closure of the mine the stockpiles are to be graded to encourage runoff and limit infiltration. The surface is to be covered with soil and vegetated. The protective berms diverting surface flow are to remain to avoid any erosion of the soil cover.*
- *The addition of crushed limestone (marble) to the stockpile to raise the pH and precipitate metals in the long term is recommended. On closure of the mine the stockpiles are to be graded to encourage runoff and limit infiltration. The surface is to be covered with soil and vegetated. The protective berms diverting surface flow are to remain to avoid any erosion of the soil cover.*
- *Secure the pit against inflow of surface runoff water and discharge.*
- *The suitability of the water accumulated in the mine pit for use in the plant is to be evaluated.*
- *The mine pit could be dosed with acid neutralisation material such as marble or limestone. Reactions may be hindered by formation of 'armour' of $\text{Fe}(\text{OH})_3$ and has to be ground to sand size particles for effective neutralisation in the long term (The Global Acid Rock Drainage Guide). By implementing the recommended monitoring measures, it will be possible to define if and when the equilibrium can be reached. The monitoring program should be adjusted going forward based on these results.*
- *On closure the mine pit should be cordoned off to avoid access and use by animals and humans.*
- *Rock Waste Dumps and Remaining Stockpiles: On closure of the mine the stockpiles are to be graded to encourage runoff and limit infiltration. The surface is to be covered with soil and vegetated. The protective berms diverting surface flow are to remain to avoid any erosion of the soil cover.*

21 CAPITAL AND OPERATING COST ESTIMATES

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

21.1 INTRODUCTION

The Okanjande overall capital and operating cost estimate was developed by Creo Engineering Solutions Pty Ltd and MSA Group, with input from Northern Graphite Corporation. The cost estimates were developed to support a nominal mining rate of 631,450t/a. The process facility cost estimate is based on a nominal throughput capacity of 631,450t/a, and producing on average 31,103t/a of graphite concentrate per year.

21.2 ESTIMATE ACCURACY

The cost estimates have been prepared in accordance with the recommended practices of the American Association of Cost Engineers (AACE) for which the minimum accuracy for this level of study is considered to have an accuracy of +40 -30% overall.

21.3 PROJECT CURRENCY AND EXCHANGE RATES

The currency for the cost estimate is expressed in 2nd Quarter 2023 US dollars. No provision is included for potential future cost escalation.

The following foreign currency exchange rates apply to CAPEX and OPEX estimates in this study:

- NAD18.50 : USD1.00

21.4 BASE DATE

The base date of the estimate is 30 June 2023 and will be subject to contract price adjustment and escalation as from the estimate base date.

21.5 SCOPE OF ESTIMATES

The scope of facilities addressed in the cost estimate includes the following major elements:

- Mining (OPEX Only)
 - Drill and Blast
 - Load and Haul
 - Re-Handling
 - Crushing (Jaw Crusher)
- Comminution (CAPEX and OPEX)
 - Material Handling and Stockpiling
 - Grinding (AG and Rod Mill)
 - Material Handling (Loading and Conveyors)
 - Classification (Screening & Cyclones)
 - Iron Removal (LIMS)
- Flotation (CAPEX and OPEX)
 - Roughers (Relocated and Repurposed)
 - Cleaners (Relocated and Repurposed)
 - Polishing (Relocated existing and New)
 - Classification (Relocated Screening)

- Dewatering and Bagging (CAPEX and OPEX)
 - Thickening (Relocated Screening)
 - Filtration (Relocated Screening)
 - Drying (Relocated Screening)
 - Classification (Relocated Screening)
 - Bagging (Relocated Screening)
- Tailings Dewatering and Storage (CAPEX and OPEX)
 - Filter Press Feed
 - Plate and Frame Filters
 - Pumps, Piping and Valves
 - Dry integrated waste dump
 - Run-off collection and ponds
 - Pumps and pipelines

21.6 CAPITAL COST

21.6.1 CAPEX Summary

21.6.1.1 Initial Capital

The initial project capital expenditure for the Okanjande Process Plant and ancillaries is given in Table 21-1. The majority portions of the initial capital are required for the construction of the new Grinding circuit, new Tailing Dewatering circuit and Integrated Waste Dump, Sitewide Services and Infrastructure for the new site and the relocation of the existing plant equipment from the Okorusu site to the new Okanjande site.

Table 21-1: Okanjande Initial CAPEX Summary

Year 0 CAPEX		US\$	% of DFC	% of TIC
Direct Field Costs	Site Wide Services and Infrastructure	5,694,823	22.8%	16.4%
	Mining and Stockpiles	0	0.0%	0.0%
	Crushing	0	0.0%	0.0%
	Milling	7,996,303	32.0%	23.1%
	Flotation	3,104,756	12.4%	9.0%
	Concentrate Dewatering	1,100,189	4.4%	3.2%
	Screening & Bagging	287,080	1.1%	0.8%
	Tailings Disposal	5,874,712	23.5%	17.0%
	Reagent Handling	8,108	0.0%	0.0%
	Utilities	335,135	1.3%	1.0%
	General Facilities & Buildings	574,045	2.3%	1.7%
TOTAL : DIRECT FIELD COST		24,975,151	100.0%	72.1%
Indirect Field Costs	Contingency	4,995,030	20.0%	14.4%
	EPCM	2,758,723	11.0%	8.0%
	Indirect Construction Costs	1,467,417	5.9%	4.2%
	Owners Team Cost	424,578	1.7%	1.2%
TOTAL INSTALLED COST		34,620,899	122.3%	100.0%

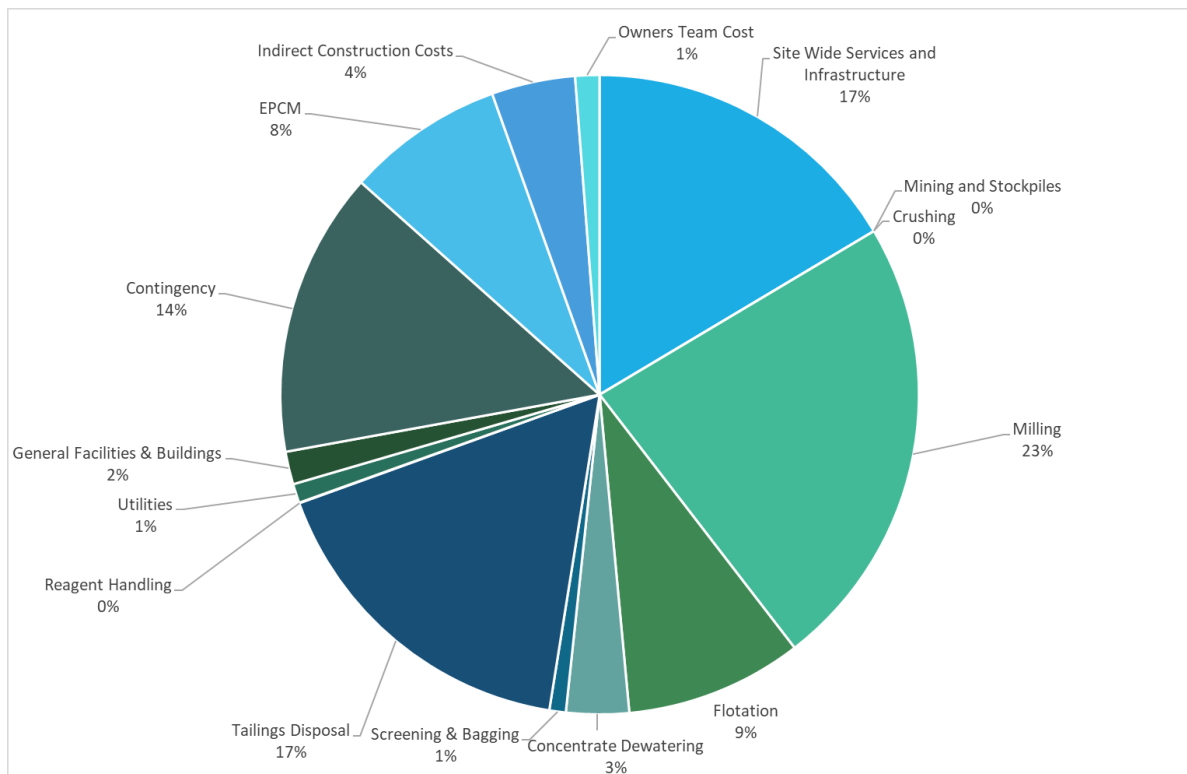


Figure 21-1: Initial CAPEX Distribution

Table 21-2 presents the initial capital related to the construction of the first two years of the integrated waste dump.

Table 21-2: IWD Initial CAPEX

IWD		US\$	% of DFC	% of TIC
	Tailings Disposal	1,386,828	100.0%	72.1%
TOTAL DIRECT FIELD COST		1,386,828	100.0%	72.1%
Indirect Field Costs	Contingency	277,366	20.0%	14.4%
	EPCM	153,187	11.0%	8.0%
	Indirect Construction Costs	81,483	5.9%	4.2%
	Owners Team Cost	23,576	1.7%	1.2%
TOTAL INSTALLED COST		1,922,439	138.6%	100.0%

21.6.1.2 Life of Mine Capital

The Initial Capital cost covers the new site preparation and infrastructure, relocation of the existing flotation plant from Okorusu to Okanjande, construction of the new grinding and tailings dewatering circuits and pad preparation for the integrated waste dump for the first 2 years of deposition for tailings and waste rock.

In year 1 the first down payment is required (30% of total) for the NamPower grid connection to the Okanjande site. The remaining payments are allowed for in 4 equal payments over the next 2 years.

A sustaining capital provision of 1% of Initial Direct Field cost was made annually (Table 21-3). Closedown costs for the IWD reclamation was factorized. A working capital provision was made based on 3 months of anticipated year 1 annual operational cost.

Table 21-3: Okanjande Life of Mine Capital costs

Capital Class	Description	US\$
Initial Capital	Process Plant, 2 Years of IWD	34,620,899
Expansion Capital	NamPower Grid Connection	4,608,447
Sustaining Capital	1% of Initial Direct Field Costs Annually	1,404,501
Close-down costs	IWD Reclamation	198,443
Working Capital	3 Months of year 1 OPEX	3,017,888
Total LOM Capital		43,850,178

21.6.2 Basis of CAPEX Estimate

The estimate is developed based on a blend of material take-offs and factored quantities and costs, semi-detailed unit costs and firm quotes for certain work packages and major equipment supply.

The structure of the estimate is a build-up of the direct and indirect cost of the current quantities; this includes the installation/modification/rectification/construction costs, contractor construction distributable costs, bulk and miscellaneous material and equipment costs, any subcontractor costs and freight. Source data used for developing the estimate included:

- Process design basis and criteria
- Process flow diagrams
- Drawings and Sketches
- Major Equipment list for new equipment
- Minor equipment list from the existing operations requiring refurbishments, modifications or removal.
- Budgetary quotes for major equipment
- In-house historical data from similar recent in country projects
- Multiple site visits and equipment inspections on existing equipment together with historical information reviews and reports.

The pricing and delivery information for quoted equipment, material and services was provided by suppliers based on the market conditions and expectations applicable at the time of developing the estimate. The estimate in this report is based on information provided by suppliers and assumes there are no problems associated with the supply and availability of equipment and services during the execution phase.

21.6.2.1 Direct Field Costs

The estimating approach adopted was to identify and quantify the major mechanical equipment as well as most of the minor equipment, including the activities required for the relocation process. The supply and installation of these identified items were then priced through an enquiry process to selected established equipment/technology suppliers, to obtain a combination of budget and firm pricing. Supply and/or relocation costs for mechanical equipment that were not obtained through an enquiry process are based on developed preliminary material take-offs together with recent local contractor rates.

Specific transport allowance was made for the supply of all equipment from the supplier's works on an indicative cost, apart from the milling equipment supplied from China, which was based on an enquiry process. No provision has been made for any import duties and taxes. Transport provision costs are included in the respective discipline costs.

All other discipline specific costs have been estimated based on the process described under the respective functional area cost estimate section.

The complete estimate, together with developed material take-off's, was based on site layout drawings, process flow diagrams (PFDs), basic engineering to detailed designs, equipment lists and schedules to obtain the required class of estimate.

Included in the estimate is the project execution based on reimbursable engineering, procurement, construction and management (EPCM). The estimate for the EPCM services is based on a continuous engineering, procurement and construction effort, with no interruption of the execution schedule after funding approval has been given. It is assumed that all environmental, licensing and permitting requirements will be fulfilled to allow the continuous execution of the project.

21.6.2.2 Indirect Field Costs

21.6.2.2.1 *Indirect Construction Costs*

The indirect construction related costs with respect to the Civils and Earthworks (Factorised at 15% of the discipline specific direct field costs) has been estimated as a percentage of the direct field costs. The related structural, mechanical, piping, platework, electrical and instrumentation (SMPPEI) indirect construction related costs has been estimated at only 3% due to bulk cost inclusions under the direct field costs. This is supported by

tender returned proposals associated with the envisaged relocation and site construction works in the respective disciplines.

21.6.2.2.2 Engineering, Procurement and Construction Management (EPCM) Costs

These costs cover the project management, engineering, and procurement, and construction management (EPCM) costs directly associated with the implementation of the relocated activities including the planned new additions to the process.

The EPCM estimate was based on a blended factorized provision, split between the different functional areas, taking into consideration the amount of EPCM effort will be required between relocated, modified and new process addition activities.

The estimate provision ranged between 8% for areas partially managed and executed in-house to 14% for areas completely outsourced. The resultant provision for EPCM services amounted to 11.0% of direct field costs.

21.6.2.2.3 Owners Team Costs

This estimate includes a limited amount (1.7% of total project direct field costs) for provision of owner's cost at the request of the client. Generally, Owner's costs do include, but are not necessarily limited to, the following:

- Owner's contingency or management reserve for changes in scope or additional work,
- Pre-development costs (cost of study, etc.);
- Project related insurances;
- Business systems;
- Loss of production and efficiency resulting from implementation;
- Owner's start-up and commissioning crew;
- Project taxes, fees, duties, customs, permitting and approvals;
- Development fees and approval costs of statutory authorities;
- Finance fees or cost of capital;
- Pre-production costs (operator training);
- Cost of any disruption to normal operations;
- Cost of shutdowns;
- Workplace health and safety fees;
- Operational readiness;
- Environmental considerations (HA);
- Additional study fees;
- Mining design, mine planning and related studies costs;
- Withholding tax;
- Import duties;
- Working capital or any additional stay-in-business capital (sustainability capital) not specifically covered in this report

21.6.2.2.4 Contingency

Contingency is a sum of money included in an estimate to allow for uncertainty related to estimating "accuracy" and the risk of undertaking the project. Contingency is a specific allocation of resources (money, people, time, etc.) added to the estimate for uncertainties due to an evaluation of the possibility, probability, risk, and consequences of events, which can cause overrunning the base estimate. Contingency excludes major scope changes, extraordinary events, management reserves, and escalation and currency effects. A blended percentage allowance has been made on the estimate for contingency based on the confidence level of the

individual functional area costs. A contingency amount of 20% was applied across the board. Whilst relatively low for a PEA level study, the lower contingency rate is supported by a significant fraction of equipment costs already sunk.

21.6.2.3 Estimate Specific Exclusions

- Any costs associated with process or engineering design changes necessitated by the outcome of geotechnical investigations;
- Owner's contingency allowance or management reserve;
- Withholding Tax & Value Added Tax (VAT);
- Customs and importation duties.
- Escalation from the estimate base date to project execution;
- Foreign currency exchange rates variations from the estimate base date;
- Schedule acceleration costs;
- Schedule delays and associated costs, such as those caused by:
 - Unexpected site conditions;
 - Weather conditions other than fair;
 - Unidentified ground conditions, and
 - Labour disputes.
 - Any acceleration of project cost;
- EPCM contractor support charges and Owner's costs associated with commissioning activities
- Infrastructure outside the battery limit unless specifically included;
- Any spares costs more than allowances made;
- All mobile equipment required to operate the plant;
- Provision of landscaping and nursery services;
- Medical staff and practitioners including emergency vehicles during construction;
- Mining and mine closure/rehabilitation costs;
- Laboratory information management systems (LIMS);
- Higher level management systems (MIS, MES or ERP);
- Risk and financial modelling;
- CCTV systems;
- Construction camp and Camp management costs (e.g. Day-to-Day Maintenance, Cleaning and Laundry);
- Travelling Costs (Flights and Accommodation) for Quality/Engineering Inspections of Equipment outside of Namibia;
- Two years' operating spares;
- Costs of EIA permitting and regulatory compliance requirements, together with any costs associated with process design changes necessitated by the outcome of such investigations;
- Any costs associated with statutory requirements, local permits, licensing, royalties and approvals, social, community or environmental requirements;
- Financing, Marketing & Business Systems costs;
- Operational costs (included separately under Operating Cost Estimate);
- External auditing costs;
- Force Majeure;

21.6.3 Mining Capital Cost Estimate

The mining associated capital cost estimate has been estimated to be zero due to the company's strategy of employing contractor mining. The mining contractor will be responsible for establishing their own specific

required infrastructure, but where shared infrastructure is cost beneficial (i.e. Power, Water and Internal Roads), the appropriate capital requirement has been included in the plant capital estimate.

21.6.4 Process Plant Capital Cost Estimate

The scope of facilities described herein, together with the exclusions as requested by the client, will be required to develop and successfully execute this project. This section will define the project scope detail, the scope inclusions, project specific exclusions and battery limits for the Okanjande Graphite Project.

21.6.4.1 Scope of Facilities

The scope of facilities covers the equipment and material relating to the relocated process plant, any planned additions and associated infrastructure, within the battery limits of the following declared processes/areas:

- Area SI : Site Wide Services and Infrastructure
- Area MN: Mining and Stockpiles
- Area CR : Crushing
- Area ML : Milling
- Area FL : Flotation
- Area CD : Concentrate Dewatering
- Area SB : Screening & Bagging
- Area TD : Tailings Dewatering and Disposal
- Area RH : Reagent Handling
- Area UT : Utilities
- Area GF : General Facilities & Buildings

21.6.4.2 Area SI: Site Wide Services and Infrastructure

The Okoruso processing plant will be relocated to the Okanjande process site. Site wide services and infrastructure estimates are based on the new site requirements and project site layout.

The site required electrical infrastructure were determined from the existing load requirements of the Okoruso facility as well as the new plant additions to be erected at Okanjande.

The largest single contributor is Nampower grid connection costs which will require a 30% downpayment in year 1, followed by 4 equal payments over the following 2 years. This will allow the lowest power cost option of a combination of grid supply and solar PV supply.

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Table 21-4: Site Wide Services and Infrastructure Direct Field Costs

Site Wide Services and Infrastructure	US\$
Civils	366,663
Concrete small items: 5%	62,162
Fencing - Sewerage & Stormwater Pond (1.8m Security Chainlink)	27,351
Sewer network - 110 Ø UPVC sewer line	33,584
Sewer network - Manholes 0-1m deep	9,946
Sewer network - Manholes 1-2m deep	16,216
Sewer network - Pond 400m ³	162,162
Sewer network - Septic tanks 1.8m ³	6,486
Sewer network - Small items: connections, fencing, blocks	32,432
Water Storage Tank - Ring Beam & Foundation	16,322
Earthworks	2,306,564
Clearing, platforms, roads and storm water	2,009,613
Culverts on Access Road - 2 of 900x450	9,270
24,000m ³ Stormwater Pond	287,681
Electrical & Instrumentation	4,823,460
22 kV indoor substation with 3 x Incomers and 5 x Feeders	194,595
929 IT General	81,081
Admin Building Reticulation	14,446
Canteen Reticulation	0
Change house Reticulation	2,703
Clinic Reticulation	1,351
Earthing	73,212
Kiosks for Site Reticulation (Site Lighting and Buildings)	5,384
Low Voltage Cabling for Site Reticulation and Lighting)	166,036
Medium Voltage Cabling to connect Transformers	22,811
Mining Infrastructure, ITC & Crushing LV Distribution Cabling	34,459
NAMPOWER Grid Supply Connection	4,008,840
Relocation of existing MV Electrical Equipment	29,384
Security and Induction Reticulation	1,351
Site Lighting (10 High Masts)	77,736
Site Lighting (20 Street Lights)	11,103
Terminations of 22 kV Cables for transformers	4,108
Warehouse Reticulation	13,374
Water Supply - Prime Power Generators at BHs	37,838
Water Supply - Prime Power Generators at Booster PS	16,216
Water Supply - Solar at BH and Booster	27,432
Mechanical	1,522,622
926 Fuel & Lubricants	29,730
Sewer network - Trickling filter: Ecosmart 5 stage series 300	98,595
Stormwater Pump Station, Distribution Piping & Filtering	33,784
Water Supply - BH Pipeline	41,135
Water Supply - Booster Pump Station at BHs	45,000
Water Supply - Borehole Pump Stations (Excl. Drilling) 130km ³ /a	21,081
Water Supply - Borehole Reservoir	24,324
Water Supply - Main Pipeline to Site for 130km ³ /a or 17.8m ³ /hr for 20hrs	1,003,568
Water Supply - Storage on Site (Relocate existing tanks)	18,919
Water Supply - Water Reticulation	81,081
Water Supply - Booster Pump Station on Site	37,838
Water Storage Tank	68,649
Stormwater pond return pump and piping	18,919
Total	9,019,308

21.6.4.3 Area MN: Mining and Stockpiles

Mining will be performed by a mining contractor. Existing mining workshop and admin buildings will be used by the mining contractor. No additional capital is allowed for mining and stockpiles.

21.6.4.4 Area CR: Crushing

Primary crushing will be contracted together with mining and as such no capital cost allowance was made for crushing in this study.

21.6.4.5 Area ML: Milling

The milling circuit will consist of the following major equipment:

- Hopper feeder (Mining Contractor scope and cost)
- Stockpile feed conveyor (70m, Mining Contractor scope and cost)
- Stockpile drawdown feeders (x2)
- SAG mill feed conveyor with tramp magnet and belt scale
- SAG mill with feed chute and trommel screen on the discharge
- Trommel screen oversize / scats bunker
- Common mill sump for the SAG and Rod Mill with recirculation pumps
- Hydro-cyclones for classification
- Rod mill
- Low Intensity Magnetic Separator (LIMS)
- Magnetic Rejects bunker
- Flotation Feed buffer tank, agitated

The costs associated with the milling circuit major mechanical equipment were obtained through formal enquiry processes where several suppliers were engaged for the supply of the required equipment. The site installation associated costs were obtained through completing the basic engineering, developing the supported bill of quantities and going out to market for tenders pricing. Returned tenders were evaluated and subsequently included in the estimate provided.

Electrical and Instrumentation installation costs were estimated based on the completed basic engineering, high level material take-offs and applying local contractor rates for the envisaged scope of services. The major electrical equipment supply costs were obtained similarly to the mechanical major equipment costs; through a formal enquiry process.

The estimate for the supply and installation of structural steel in the milling plant followed pricing received from a comprehensive enquiry process for the milling plant. The transport and civil costs associated for the milling plant was also obtained through several enquiries to the market supported by the necessary bill of quantities.

Table 21-5: Milling Total Direct Filed Cost

Milling	US\$
Civils	502,954
Concrete for Mills	297,204
Front End: Concrete Conveyor bases	51,892
Front End: Concrete Stockpile Draw down and culvert	153,858
Electrical & Instrumentation	1,713,553
22 kV Cabling and Transformers - Original Milling Scope	335,534
525V and 690V MCC's	798,847
EC&I Installation	579,172
Mechanical	3,951,793
Agitator Supply	7,304
Cyclones	75,507
Front End: Chute-Ball Loading and Scats (Future, not priced)	0
Front End: Installation	97,786
Front End: Mill Feed Belt Conveyor and Ancillary Equipment (includes weightometer)	326,486
Front End: Pebble Crushing Circuit (FUTURE) - 3 x Conveyors, Chutes & Crusher, Bin & Magnet	0
Front End: Stockpile Belt Conveyor and Ancillary Equipment	0
Front End: Vibrating Feeders and Ancillary Equipment	87,164
LIMS	55,177
Piping, Valves & Fittings Supply - Relocation Replacements	135,135
Pump Supply Package	259,198
ROD Mill and SAG Mill	1,306,306
Sampler Package	27,529
SMPP: Milling Plant (Current SMPP Tender Excl. Structural Steel - Priced Elsewhere)	1,460,688
Relocation of Derrick Screen and Magquip Mag Sepp from Okorusu	113,514
Structural	442,868
SS for Mills	442,868
Transport	1,385,135
Logistics and Transport Package	1,385,135
Total	7,996,303

21.6.4.6 Area FL: Flotation

The largest contributor to flotation circuit mechanical costs are due to a new polishing scrubber #1 and two new polishing vertical stirred media mills, for which firm costs were obtained. The associated installation costs were obtained through an enquiry process going out to market for pricing.

Electrical installation, as well as supply, costs for the new equipment were estimated based on the completed basic engineering, material take-offs and inviting local installation contractors for pricing. The instrumentation equipment supply and installation costs were estimated based on developed material take-offs and applying recent contractor rates by in-country contractors.

The estimate for the supply and installation of structural steel and civil works followed high level developed material take-offs and applying tender rates as received through other enquiries, as for the milling section.

The associated relocation costs (E&I, Civil, Mechanical, Piping and Structural Steel) were obtained through inviting contractors to site and enquiring budget estimates for the relocation of the planned plant infrastructure.

Table 21-6: Flotation Total Direct Field Costs

Flotation	US\$
Civils	561,979
Civils for Electrical Relocated Equipment	12,973
Concrete Flotation Incl NEW Sumps	264,314
Concrete for Polishing Mills, Scrubber & Thickener	257,449
Control Room for SCADA and PLC Servers (Double Story)	27,243
Electrical & Instrumentation	100,627
Equipment replacement - Provisional	54,054
Recloser 2000 kVA transformer	14,496
Relocation of 2000 kVA transformer	10,243
Relocation of MCC and associated equipment	14,186
Relocation of PLC, Field Instruments and Associated C&I	7,648
Mechanical	1,726,491
MPP: Relocation of Mechanical Equipment of Flotation, Drying & Bagging Plant	888,162
Rotary Drum Scrubber	269,612
SMM Stirred Mills	426,826
On-line Analyser XRF	141,892
Structural	715,659
SS Modification & Relocation Flotation	150,811
SS Modification & Relocation Polishing, Thickener	150,811
SS Relocation Flotation	196,305
SS Relocation Polishing, Thickener	210,326
SS Relocation Sheeting Flotation	3,576
SS Relocation Sheeting Polishing, Thickener	3,831
Total	3,104,756

21.6.4.7 Area CD: Concentrate Dewatering

The associated mechanical, electrical and instrumentation costs were included under the flotation section estimate. The relocation costs (Civil and Structural Steel) were obtained through inviting contractors to site and enquiring budget estimates for the relocation of the planned plant infrastructure. The budget enquiries received were also compared and adjusted to include the developed high level material take-offs with applied local received contractor rates from tender enquiries of other plant sections.

Table 21-7: Concentrate Dewatering Total Direct Field Costs

Concentrate Dewatering	US\$
Civils	498,162
Concrete Drying and Bagging	498,162
Electrical & Instrumentation	0
Priced Elsewhere - Flotation Section	0
Mechanical	0
Relocation Costs for MPP - Priced under Flotation Section	0
Structural	602,027
SS Modification & Relocation Drying and Bagging	226,216
SS Relocation Drying and Bagging	370,946
SS Relocation Sheeting Drying and Bagging	4,865
Total	1,100,189

21.6.4.8 Area SB: Screening and Bagging

The associated civil, mechanical and structural steel relocation costs were included under the concentrate dewatering and flotation section estimate respectively. The heat shrinker was obtained from a formal quote from a preferred supplier.

Electrical supply, replacement and relocation costs were estimated based on developed high level material take-offs and applying recent contractor rates by in-country contractors.

Table 21-8: Screening & Bagging Total Direct Field Costs

Screening & Bagging	US\$
Civils	0
Costed Elsewhere (Concentrate Dewatering)	0
Electrical & Instrumentation	126,080
Equipment replacement - Provisional Only	54,054
Recloser 1000 kVA transformer	14,496
Relocation of 1000 kVA transformer	10,243
Relocation of MCC and associated equipment	47,286
Mechanical	161,000
Heat Shrinker	161,000
MPP: Relocation of Mechanical Equipment - Costed Elsewhere (Flotation)	0
Supply and Installation of additional Concentrate Product Screen - By Others	0
Structural	0
SS: Relocation of Structural Steel - Costed Elsewhere (Flotation)	0
Total	287,080

21.6.4.9 Area TD: Tailings Dewatering and Disposal

Tails from the Rougher and cleaner stages will be combined in a filter feed tank. Slurry will be filtered, un-thickened, through 3 plate and frame filter presses. Firm quotes for a turn-key filtration plant were used as the basis for estimate.

Dry (filtered) tailings will be co-deposited with waste rock on the IWD. The IWD will be stacked by the mining fleet over a prepared base with run-off collection. The first 2 years of capacity will be constructed as part of the initial construction phase. The facility run-off will be collected in a lined channel and stored for treatment and re-use in a lined IWD collection pond. Expansion of the facility will be completed by the mining fleet as part of operations. A basic design for the facility was completed and the resultant bill of quantity, coupled with local construction rates, was used as the basis of estimate.

The following key items were considered for the cost estimate:

- 3 Plate and Frame filter presses with hydraulics and local controls
- Filter Feed and Filtrate tanks
- Slurry and solution pumps
- Conveyors and transfer chutes.
- Bush clearance and topsoil stripping
- Compaction of the base
- A layer of bedding sand
- Double 1.5mm HDPE liner with leak detection
- Geotextile (Bidim)
- 600mm Aggregate drainage layer
- Lined IWD run-off collection channel
- Lined IWD run-off collection pond
- Compacted and partially lined berm around perimeter of the IWD.
- IWD return water treatment: pH dosing with static mixer and sand filter unit.

Table 21-9: Tailings Disposal Total Direct Field Costs

Tailings Disposal	US\$
Civils	109,196
Concrete for Tailings Dewatering (Filter Presses, Thickener, Tanks, Bund & Bases)	109,196
Earthworks	1,886,081
Tailing Dewatering - All areas	35,000
32,000m ³ IWD Pond	439,929
IWD Construction 1st 2 years of required area	1,386,828
IWD Pond return water pump and piping	24,324
Electrical & Instrumentation	594,403
Tailing Dewatering - MCC c/w Installation works	575,485
pH, Static Mixer & NaOH Dosing	18,919
Mechanical	2,681,941
Plate and Frame Filter Press	1,135,640
Tailing Dewatering - Compressed Air Installation for Filter Press	249,751
Tailing Dewatering - Conveyors	374,588
Tailing Dewatering - Piping and Valves	253,526
Tailing Dewatering - Process Pumps	132,785
Tailing Dewatering - Process Tanks All	189,677
Tailing Dewatering - Sampler	30,270
Tailing Dewatering - Tailings Dewatering Installation	0
Aggregate Crushing and Screening Plant	291,378
Return Water Sand Filters	24,324
Structural	603,089
SS Dewatering for Tails	603,089
Total	5,874,712

21.6.4.10 Area RH: Reagent Handling

Mechanical relocation estimates were estimated based on developed high level material take-offs and applying recent contractor rates by in-country contractors. The associated structural steel and E&I costs were included under the flotation discipline estimate.

Table 21-10: Reagent Handling Total Direct Field Costs

Reagent Handling	US\$
Electrical & Instrumentation	0
Relocation Costs : Priced Elsewhere (Flotation)	0
Mechanical	8,108
Reagent Relocation	8,108
Structural	0
Relocation Costs : Priced Elsewhere (Flotation)	0
Total	8,108

21.6.4.11 Area UT: Utilities

The provision for utilities included:

- Relocation of the existing compressed air reticulation and provision for the extension of the compressed air reticulation to service the new Milling section.
- Provision has been included to provide a firefighting system supplied from the raw water dam with dedicated fire water pumps and a dedicated fire water ring main. The ring main will further have several hose reels and hydrant connection points throughout the processing facility.
- A RO plant has been included for the provision of potable water to the site, adequately sized to supply approximately 50 employees daily.
- Installation of a new 400m³ Raw Water Storage tank

The relocation and new estimate costs were based on developed high level material take-offs and applying recent contractor rates by in-country contractors.

Table 21-11: Utilities Total Direct Field Costs

Utilities	US\$
Civils	0
Concrete Fuel bay (Estimated Quantities - Not Costed)	0
Concrete Generator Plant - IPP (Estimated Quantities - Not Costed)	0
Mechanical	335,135
Bulk Fuel Storage	0
Compressed Air Installation - New reticulation	14,865
Compressed Air Installation - Relocation of plant (assumed to be sufficient)	81,081
Fire Fighting - Diesel Driven Booster Pump Station	51,351
Fire Fighting - Firefighting Equipment	13,514
Fire Fighting - Firefighting Water Reticulation	40,541
Fire Fighting - Smoke Ventilation (Flotation Plant, Workshops, Warehouse)	40,541
Fire Fighting - Storage Tank	0
Potable Water Plant : Containerised RO Plant incl. Booster Pump	79,730
Raw Water Pump System Relocation	13,514
Structural	0
SS Generator Plant - IPP (Estimated Quantities - Not Costed)	0
SS Relocate existing 3 Sectionalised Tanks (2 x Raw Water & 1 x Process)	0
Total	335,135

21.6.4.12 Area GF: General Facilities and Buildings

Provision has been included in the estimate to construct new facilities and buildings as deemed necessary. The estimate was based on preliminary designs, developed material off takes and using recent local contractor rates.

Table 21-12: General Facilities and Buildings Total Direct Field Costs

General Facilities & Buildings	US\$
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Civils	549,720
Buildings - Admin	52,757
Buildings - Change house	101,293
Buildings - Clinic (Relocate only of existing and Construct base)	19,344
Buildings - General Workshops (Relocate and NEW)	220,541
Buildings - Main Warehouse	119,189
Buildings - Mine store containers, 2 of 12m	8,108
Buildings - Security and Access	28,489
Buildings - SHE, Canteen, training	0
Mechanical	24,324
Compressed Air Installation for Workshop	24,324
HVAC (This include for fire ventilation at plant areas such as workshop and flotation plant)	0
Total	574,045

21.7 OPERATING COSTS

Total operating cost estimates for the Okanjande Project are presented in Table 21-13. The unit operating costs are based on total mined material of 10.1Mt, of which 6.1Mt is mineralized material and 4.05Mt is waste. The estimated mine life is ten years plus one year of pre-mining/construction. Processing will continue into Year 11.

Mining Operating costs include Drill and Blast for Mineralized material and Waste, Load and Haul for Mineralized material and Waste, Mobilization and P&G's.

Process operating costs include comminution, flotation, dewatering, drying, classification, and bagging. Tailings management, labour, power, water, packaging, reagents and consumables are also included. No product transport or shipping costs are included as the pricing for the study is on an ex-gate basis.

General & Administrative (G&A) costs as per Table 21-13, include all standard site general and administration costs, as well as Security Services, Surface Rights Charges, External Services and Royalties. The unit cost for mineralized material processed is based on the total mineralized material tonnes from the Okanjande mine and total graphite concentrate produced from the processing plant. The basis for these rates is 6.1Mt Mineralized material and 311kt of graphite concentrate.

Table 21-13: Summarized Operating Costs (Total Life of Mine)

Operating Costs (C1)	(US\$ 10 ⁶)	\$/t Ore Processed	\$/t Concentrate Produced
Mining	60.80	9.97	195.47
Processing	131.97	21.64	424.28
G&A	19.47	3.19	62.59
Total	212.23	34.81	682.34

21.7.1 Exclusions

The following cost contributors were not accounted for in the OPEX estimate:

- Scope changes
- Escalation beyond the first quarter of 2023
- Financial cost
- Schedule delays
- Operating permits
- Currency fluctuations
- Reagent and Fuel Pricing fluctuations

21.7.2 Basis of Operating Costs

Mining costs are based on engaging a contract miner and budgetary proposals received in June 2022.

Processing labour costs were developed from the anticipated staffing plan. Labour rates were applied with consideration for burden and overtime. The specific rates were based on the estimator's judgment from experience from Namibian processing plants and average salaried rates. Power cost estimate was based on recent costs for IPP provided hybrid solar and diesel generation, the current local utility rate of US\$0.11/kWh, and the power consumption estimates from the electrical load list. Variable costs for reagents and consumables were applied based on recent test work results, historical consumption rates and current local pricing for reagents. Budgetary quotes for all major reagents and consumables were obtained from primary suppliers in the region. Fixed cost elements were derived from factoring and from previous operating budgets for the process at Okorusu.

The supervisory and administrative support staff was sized to accommodate direct front line supervision for operations, as well as provide adequate support personnel for technical services, management, environmental, and administration.

21.7.3 Mining Operating Cost

Contractor mining will be applied for the Okanjande mine and mining costs are based on contract mining tenders received in June 2022. Drilling and Blasting costs are based on third party costs. Ore Load and Haul costs are based on 1,2 km average hauling with contractor fleet. Waste Load and Haul costs are based on 2,1km average hauling distance.

Table 21-14: Mining Operating Cost Estimate (Total Life of Mine)

Operating Costs	(US\$ 10 ⁶)	\$/t Ore Processed	\$/t Concentrate Produced
Ore Mining Costs Variable	21.44	3.52	68.94
Waste Mining Costs Variable	13.39	2.20	43.06
Mining Costs Fixed	7.43	1.22	23.88
Ore Crushing Costs	18.54	3.04	59.60
Total	60.80	9.97	195.47

21.7.4 Processing Plant Operating Cost Estimates

The major processing cost elements include variable costs such as Process General (front-end rehandling costs, grinding media), Tailings Management, Power, Water, Concentrate Packaging plus Reagents and Chemicals. Fixed costs cover Labour, Process Maintenance and Miscellaneous Consumables.

Table 21-15: Total Processing Operating Cost Estimate (Life of Mine)

Operating Costs	(US\$ 10 ⁶)	\$/t Ore Processed	\$/t Concentrate Produced
Process General	5.50	0.90	17.68
Tailings Management	21.98	3.60	70.66
Consumables - Miscellaneous	1.16	0.19	3.72
Process Maintenance	12.22	2.00	39.30
Labour	23.32	3.82	74.97
Power	44.82	7.35	144.09
Water	6.42	1.05	20.66
Concentrate Packaging	9.05	1.48	29.09
Reagents / Chemicals / Consumables	7.50	1.23	24.12
Total	131.97	21.64	424.28

21.7.4.1 IWD Operating Costs

The integrated waste dump will be of the dry stacking type for the disposal and storage of mineralized material tailings and waste rock. Dry (filtered) tailings will be co-deposited with waste rock on the dry IWD facility. Dry filtered tailings will be loaded from the process plant and hauled and dumped on the IWD by the contract miner and the rates are based on tendered waste rock load, haul and dump rates, for the lesser distance from the tailings filtration plant to the IWD. Operational costs include IWD base preparation after year 2. Consulting services include collecting and reporting on TD monitoring data through site visits and inspections with reporting, monitoring data review and quarterly summaries and Ad-Hoc technical support.

Table 21-16: TSF Operating Cost Estimate

IWD Operating Costs	Rates (USD/t)	Average Annual Cost (USD)
Variable Cost Rates	2.68	1,775,047
<i>Load, Haul and Dump Cost</i>	2.46	1,629,476
<i>Capitalize additional mobile equipment</i>	0.22	145,571
Fixed Costs	0.64	421,115
<i>Consulting Services</i>	0.02	16,216
<i>Base Expansions (averaged over LOM)</i>	0.61	404,899
Total IWD Operating Costs	3.32	2,196,162

21.7.4.2 Labour

A labour complement was developed based on the previous operational structure together with some optimization in areas where it was deemed necessary (Table 21-17). Salary scales are based on experience with typical current Namibian salaries for positions in mines and processes of similar complexity and size.

Table 21-17: Labour Complement

Position	Headcount
Management	
General Manager	1
Mine Manager	1
Process Manager	1
Engineering Manager	1
Administration Manager	1
Geologist*	1
Finance and Administration	
Management & Services	3
Procurement	2
Administration/Clerks	10
Health, Safety & Environmental	3
Processing	
Management - Supervisor	4
Process plant operators	20
Dayshift operators	2
Laboratory assistants	8
Engineering	
Mechanical and Boilermaker**	6
Electrical**	6
C&I	2
Assistants	8
TOTAL	80

Table 21-18: Annual Labour Cost Estimate

Labour Costs	Annual Costs (US\$)
Management	560,368
Finance and Administration	165,016
Processing	96,259
Engineering	103,135
Total	924,778

21.7.4.3 Reagents and Consumables

Reagents and consumables were based on historic consumption rates and updated quotes from primary suppliers in the region.

Table 21-19: Reagents and Consumables Operating Cost Estimate

Reagents and Consumables	Unit Costs (US\$/t)
Diesel - Flotation	2.16
Flocculent- Concentrate	0.09
Water Treatment	0.17
MIBC	4.56
Diesel - Dryer	13.31
Filter Cloth Consumption	0.30
Miscellaneous	3.54
Total	24.12

21.7.4.4 Power

The power cost estimate was based on the electrical load list and recent costs for IPP provided hybrid solar and diesel generation, the current local utility rate of US\$0.11/kWh, and the power consumption estimates from the power consumer summary (Table 21-20).

Table 21-20: Okanjande Power Consumer Summary

Rev. No.		Item No.		POWER CONSUMER SUMMARY										OKORUSO GRAPHITE PROJECT				
NORTHERN GRAPHITE CORPORATION		CREO ENGINEERING SOLUTIONS		C0123 - OKANJANDE GRAPHITE PROJECT										Rev. No.		0		
Document No.:				Prep. By		J		Bekker						Area Manager		Client Area Manager		
Checked				D		Louw						E Roux						
Approved				D		Louw												
Rev. No.	Item No.	SUBSTATION / MCC DESCRIPTION		MOTORS							Total Installed Power (kW)	Total Installed Power (kVA)	Total Running Power (kW)	Total Running Power (kVA)	Total Standby Power (kW)	Average Power Factor (Without PFC)	Power Factor Correction (kVAr)	Transformer Sizing (kVA)
		Tag Number	Description	No. of Loads	Total No.	No. of DOL's	No. of VSD's	No. of SS's	No. of FDR's									
0	01	ML-MCC-001	MILLING & GRINDING 525V MCC	23	20	8	9	3	3	1,406	1,706	943	1,145	223.09	0.82	649	3 150	
0	02	ML-MCC-002	MILLING & GRINDING 690V MCC	6	4	3	1	0	2	1,284	1,577	1,145	1,364	0.00	0.84	741	2 500	
0	03	FL-MCC-001	FLOTATION 525V MCC	15	12	1	11	0	3	427	519	342	415	14.98	0.82	236	800	
0	04	MCC-001 & 002	EXISTING FLOTATION & BAGGING MCC's	TBC	TBC	TBC	TBC	TBC	TBC			1,737	2,172					
0	05	TD-MCC-001	TAILINGS 525 V MCC	22	21	13	8	0	1	1,775	2,160	1,017	1,242	384.00	0.82	714	3 150	
SUB TOTALS				66	57	25	29	3	9	4 892	5 962	5 184	6 338		0.82	3 646		
GRAND TOTALS				66	57	25	29	3	9	4 892	5 962	5 184	6 338		0.82	3 646	9 600	

Table 21-21: Okanjande Power Cost

Grid Connection & Solar PV, BESS & Diesel Generation	Total Generation
--	------------------

Project Operating Year	Grid Supply	Solar PV	BESS	Diesel Generation			Life-cycle Cost of Energy
	OPEX (US\$)	OPEX (US\$)	OPEX (US\$)	OPEX (US\$)	CAPEX (US\$, IPP only)	OPEX (US\$, all)	US\$/kWh
Yr 1	0	32,432	12,201	5,968,140	1,134,370	6,012,773	0.221
Yr 2	0	32,432	12,201	5,968,140	1,134,370	6,012,773	0.221
Yr 3	0	32,432	12,201	5,968,140	1,134,370	6,012,773	0.221
Yr 4	0	32,432	12,201	5,968,140	1,134,370	6,012,773	0.221
Yr 5	2,167,071	32,432	0	0	489,507	2,199,503	0.084
Yr 6	2,167,071	32,432	0	0	489,507	2,199,503	0.084
Yr 7	2,167,071	32,432	0	0	489,507	2,199,503	0.084
Yr 8	2,167,071	32,432	0	0	489,507	2,199,503	0.084
Yr 9	2,167,071	32,432	0	0	489,507	2,199,503	0.084
Yr 10	2,169,892	32,432	0	0	489,507	2,202,324	0.084
Total	13,005,245	324,324	48,804	23,872,558	7,474,523	37,250,931	0.139

21.7.5 General and Administration Cost Estimate

General and Administration cost estimate is based on factorization of costs from plants of similar size and complexity as well as reigning Namibian taxation laws and recent concentrate shipment costs.

A fixed allowance was made for external services, mainly in the form of consulting.

Royalties are based on the reigning Namibian Tax legislation and is calculated as 2% of revenue.

Table 21-22: General and Administration Operating Cost Estimate (Total Life of Mine)

Operating Costs	(US\$ 10 ⁶)	\$/t Ore Processed	\$/t Concentrate Produced
Security Services	0.26	0.04	0.83
External Services	2.74	0.45	8.81
G&A	6.83	1.12	21.95
Royalty	9.64	1.58	31.00
Total	19.47	3.19	62.59

22 ECONOMIC ANALYSIS

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

This PEA is preliminary in nature. The current basis of Project information is not sufficient to convert the in-situ Mineral Resources to Mineral Reserves, and Mineral Resources that are not Mineral Reserves do not have demonstrated economic viability. The PEA results are only intended as an initial, first-pass review of the Project economics based on preliminary information.

This PEA is based on technical and economic assumptions which will be evaluated in more advanced studies. The PEA is based on the Mineral Resource estimate in section 15 (effective date November, 2021). The PEA is based on a production plan that includes material in Measured and Indicated classifications from the Okanjande Mineral Resource model. This PEA Report validates an open pit mining operation that recovers graphite mineral from the ground to be sold at a profit. The methods used to generate this PEA are consistent with good industry practice and meet the required standards for project design and estimation. The financial analysis methodology is consistent with current practice and is consistent with the stage of the Okanjande project's development.

The economic analysis of the Project assumed constant 2nd Quarter 2023 US dollars and was performed on an annual basis beginning at the start of Year 0 when operating permits are assumed to have been issued. Construction was assumed to require 1 year starting mid 2023 with processing of mineralized material through the Okanjande processing plant to start mid-2024.

This PEA assumes that the existing Okorusu processing plant will be relocated to the Okanjande mine site and that new Milling and Tailings dewatering circuits and an integrated waste dump will be constructed at the Okanjande site. The relocated processing infrastructure will be reconfigured for suitability to process the Okanjande graphite mineralized material.

The PEA estimated graphite concentrate recovery and concentrate grade is based on historical test work and recent metallurgical tests conducted on grab samples from existing weathered mineralized material stockpiles as well as fresh mineralized material drill core samples. Tests conducted simulated the process flow of the reconfigured Okanjande graphite flotation plant. The relevant test work was completed by SGS Lakefield laboratories in Canada.

22.1 PRINCIPLE ASSUMPTIONS

22.1.1 Resource

No Reserve has been declared at the time of this study. In 2021 MSA completed a mining study and the 10-year mining at 4.0% TGC cut-off grade was selected for the study. See Table 22-1

Table 22-1: 10-Year Mine plan for 4.0% TGC cut-off

Mineral Resource Assumptions	Value	UOM	Comments
Weathered Mineralized material			
Measured and Indicated Mineralized material Tons	2,540,623	Mt	MSA, 2021
Measured and Indicated Graphite Grade	5.03%	% TGC	MSA, 2021
Contained Graphite	127,881	t	MSA, 2021
Fresh Mineralized material			
Measured and Indicated Mineralized material Tons	3,556,327.00	Mt	MSA, 2021
Measured and Indicated Graphite Grade	5.44%	% TGC	MSA, 2021
Contained Graphite	193,293	Kt	MSA, 2021
Total Mineralized material			
Measured and Indicated Mineralized material Tons	6,096,950	Mt	MSA, 2021
Measured and Indicated Graphite Grade	5.27%	% TGC	MSA, 2021
Contained Graphite	321,174	Kt	MSA, 2021

22.1.2 Mining

Mining assumptions are based on the outcomes of the mining study completed by MSA in 2021.

Table 22-2: Mining Assumptions for Economic Evaluation

Mining Production	Value	UOM	Comments
Mineralized material Mining Rate- Mean	609,695	t/a	CES, MSA Mine Plan
Mineralized material Mining Rate - Steady State	631,450	t/a	CES, MSA Mine Plan
Waste Mining - Mean	404,504	t/a	CES, MSA Mine Plan
Strip Ratio	0.7	ratio	CES, MSA Mine Plan
Life-of-Mine	10	a	MSA, Mining Report

22.1.3 Processing

Processing assumption flow out of the mining assumptions and plant performance targeted by the Client. See Table 22-3.

Table 22-3: Processing Assumptions for economic evaluation

Process Production	Value	UOM	Comments
Crushing			
Annual Throughput	631,450	t/a	Northern Graphite
Standard Operating Days	365	d/a	Northern Graphite
Standard Operating Hours	12	h/d	Northern Graphite
Availability	92.0%	%	Northern Graphite
Utilization	90.0%	%	Northern Graphite
Effective Utilization	82.8%	%	CES, Calculated
Mean Crush Rate - Instantaneous	174	t/h	CES, Calculated
Milling			
Annual Throughput	631,450	t/a	Northern Graphite
Standard Operating Days	365	d/a	Northern Graphite
Standard Operating Hours	24	h/d	Northern Graphite
Availability	92.0%	%	Northern Graphite
Utilization	90.0%	%	Northern Graphite
Effective Utilization	82.8%	%	CES, Calculated
Mean Grinding Rate (dry)	87	t/h	CES, Calculated
Flotation			
Throughput (dry)	87	t/h	Northern Graphite
Grade	5.3%	%	METPRO
Recovery	92.0%	%	METPRO
Concentrate Grade	95.0%	%	METPRO
Masspull	5.1%	%	CREO, Calculated
Concentrate Production (dry)	32,213	t/a	CREO, Calculated
Co-Deposited dry Tailings and Waste Rock			
Tailings deposition moisture	12.6%	% H ₂ O m/m	CREO mass balance
IWD	10,837,335	t	CREO, 2023

22.1.4 Financial, Taxes, Royalties and Levies

Financial assumptions were based on information from reputable sources and are listed in Table 22-4.

Table 22-4: Financial assumptions

Financial Assumptions	Value	UOM	Comments
Discount Rate	8%	%	CREO
Exchange Rate	18.5	NAD:USD	XE, 6-months
Royalties	2.0%	%	KPMG, 2021
Graphite Flake Price	1,550.00	USD/t TCG	Ex-Gate Okanjande, NG 2023
Corporate Income Tax, Namibia	37.5%	%	PWC, Dec 2021
Depreciation Period	3.00	a	DM-Northern Graphite
Annual Escalation Applied	6%	%/a	CREO, NCPIX
Escalation Period Applied	1.00	a	CREO

22.2 KEY PERFORMANCE INDICATORS

The Okanjande Project will process 6,096,695t of graphite mineralized material at an average of 5.27%TGC to produce a total of 311,031t of Graphite concentrate at a grade of 95%TGC. See Table 22-5.

With mining costs projected at US\$6.93/t mineralized material mined and processing costs at US\$439.60/t concentrate produced an overall C1 cost of US\$666.07/t is achieved. Life of mine capital expenses come in at US\$115.27/t concentrate produced. See Table 22-6.

Table 22-5: Project Overall KPI's

Project KPI's	UOM	
Measured and Indicated Resource - Weathered	Mt, %TGC	2,54Mt @ 5.03% TGC
Measured and Indicated Resource - Fresh	Mt, %TGC	3,56Mt @ 5.44% TGC
Post Tax NPV @8% CoC	US\$	70,218,979
Post Tax IRR	%	36%
Post Tax and Royalty Discounted Cashflow	US\$	145,674,213
Post Tax Payback Period	a	4
Pre Tax NPV @8% CoC	US\$	120,396,195
Pre Tax IRR	%	47%
Pre Tax and Royalty Discounted Cashflow	US\$	239,077,435
Pre Tax Payback Period	a	3

Table 22-6: Key Project Cost Indices

Cost Indicators	UOM	Cost Index (US\$)
Mining & Haul OPEX/t Concentrate	\$/t con	135.88
Mining OPEX/t Total Mined	\$/t total mined	4.17
Mining OPEX/t Ore Mined	\$/t Ore mined	6.93
Process OPEX/t Concentrate	\$/t con	439.60
Process OPEX/t Graphite Concentrate produced	\$/t concentrate	22.43
Total OPEX/t Concentrate	\$/t con	635.07
Total C1 Cost/t Concentrate	\$/t con	666.07
Total CAPEX/t Concentrate	\$/t con	115.27
Total Expenses/t Concentrate	\$/t con	781.34

The resulting NPV after tax and royalties is US\$70.2M at a discount rate of 8%. The life of mine cashflow after taxes and royalties is US\$145.7M. The project Payback period is 4 years after initial capital outlay.

22.3 CASH FLOW

The projected annual production and cash flow before and after taxes is depicted in Figure 22-1 below. More detail is presented in Table 22-7. Mining production ramp-up is expected to be rapid due to the brownfield’s status of the mine. Operational costs decline after year 5 due to the completion of the grid connection to national power utility, NamPower, which allows for a much lower realized power cost after year 5.

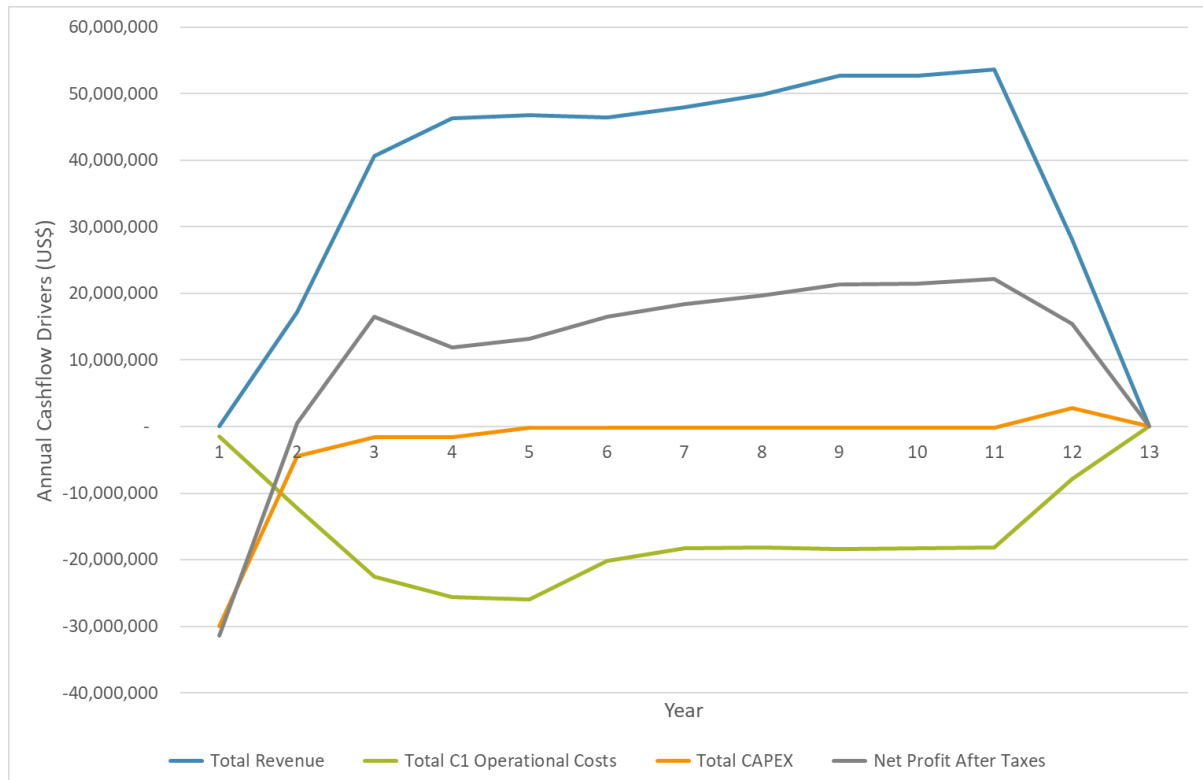


Figure 22-1: Okanjande Projected Cashflow

Table 22-7: Okanjande Project Cash Flow

Year	Mineralized Material Mining	Mineralized Material TCC	Mineralized Material Hauled and Processed	Concentrate Produced and Shipped (dry)	Revenue from Graphite Con Sales	Total OPEX	Total CAPEX	Total Taxable Income ¹	Income Tax	Net Profit After Tax	Cumulative Net Profit After Tax	
	t/a	%TGC	t/a	t/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$/a	US\$	
0	2023	-	-	-	-	(1,431,093)	(30,012,452)	-	-	(31,443,544)	(31,443,544)	
1	2024	205,221	11,451	205,221	11,090	17,189,052	(12,251,842)	(4,388,840)	-	548,370	(30,895,174)	
2	2025	520,946	27,080	520,946	26,225	40,648,084	(22,478,005)	(1,623,721)	(9,818,712)	-	16,546,358	(14,348,816)
3	2026	632,315	30,856	631,450	29,841	46,253,669	(25,548,953)	(1,623,721)	19,151,628	(7,181,860)	11,899,135	(2,449,681)
4	2027	632,315	31,172	631,450	30,146	46,726,295	(25,990,819)	(140,450)	19,606,179	(7,352,317)	13,242,709	10,793,029
5	2028	631,450	30,927	631,450	29,951	46,423,701	(20,200,907)	(140,450)	25,587,920	(9,595,470)	16,486,874	27,279,902
6	2029	631,450	31,920	631,450	30,909	47,909,230	(18,321,463)	(140,450)	29,447,316	(11,042,744)	18,404,573	45,684,475
7	2030	632,315	33,261	631,450	32,164	49,853,647	(18,210,247)	(140,450)	31,502,950	(11,813,606)	19,689,344	65,373,818
8	2031	632,315	35,124	631,450	33,961	52,639,837	(18,421,158)	(140,450)	34,078,229	(12,779,336)	21,298,893	86,672,712
9	2032	631,450	35,109	631,450	34,000	52,699,559	(18,285,094)	(140,450)	34,274,016	(12,852,756)	21,421,260	108,093,972
10	2033	631,450	35,772	631,450	34,639	53,689,696	(18,159,444)	(140,450)	35,389,802	(13,271,176)	22,118,626	130,212,598
11	2034	315,725	18,501	319,183	18,107	28,065,723	(7,868,722)	2,778,572	20,037,220	(7,513,957)	15,461,615	145,674,213
12	2035	-	-	-	-	-	-	(112,964)	-	-	145,674,213	
13	2036	-	-	-	-	-	-	(66,148)	-	-	145,674,213	
Total		6,096,950	321,174	6,096,950	311,031	482,098,493	(207,167,745)	(35,853,312)	239,077,435	(93,403,222)	145,674,213	

¹ After Depreciation and carried forward losses

22.4 SENSITIVITY ANALYSIS

The sensitivity of the PEA with regards project NPV was evaluated in terms of the main revenue drivers – Graphite Price and Graphite recovery, as well as the main cost drivers – Power, Mineralized material Mining Cost, Mineralized material Hauling Cost and NAD:USD exchange rate. All sensitivities were based on the change of the sensitivity parameters over a range of -20% to +20% change from their base case values. The sensitivity parameters were ranked based on their absolute slopes in terms of NPV per percentage change per parameter. See Table 22-8.

Table 22-8: Sensitivities Ranking (absolute slopes)

Parameter	UOM	Absolute Slope	Ranking	Class
Graphite Basket Price	US\$/t	189	1	Market
Graphite Recovery	%	130	2	Operational
Exchange Rate	NS\$:US\$	47	3	Market
Initial CAPEX	US\$	28	4	CAPEX
Power Cost (N\$/kWh)	N\$/kWh	19	5	OPEX
Total Mining Costs Variable	US\$/t	16	6	OPEX
Ore Crushing Costs	US\$/t	8	7	OPEX
NamPower Grid Connection CAPEX	US\$	3	8	CAPEX

As per Figure 22-2, the project NPV is highly sensitive towards Graphite Pricing followed Graphite Recovery (based on slope) and Namibian dollar to US dollar exchange rate. In terms of sensitivity towards operational cost drivers the project NPV is most sensitive to the cost of energy and mining costs.

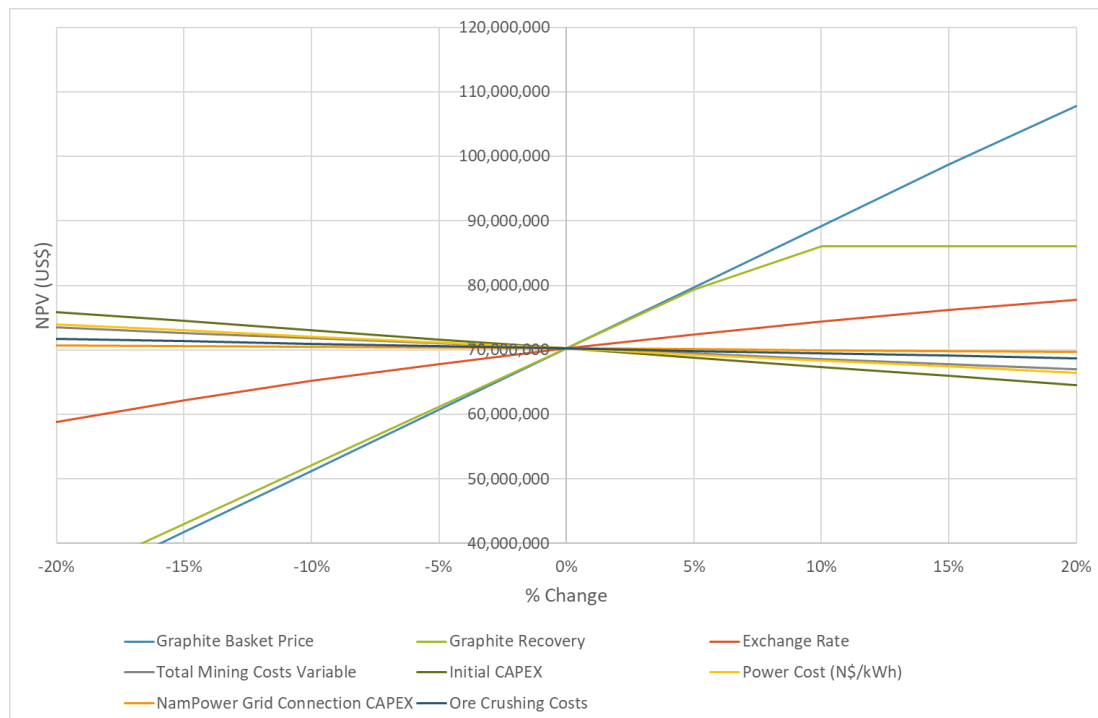


Figure 22-2: Project NPV Sensitivity wrt to main Revenue and Cost Drivers (after Tax)

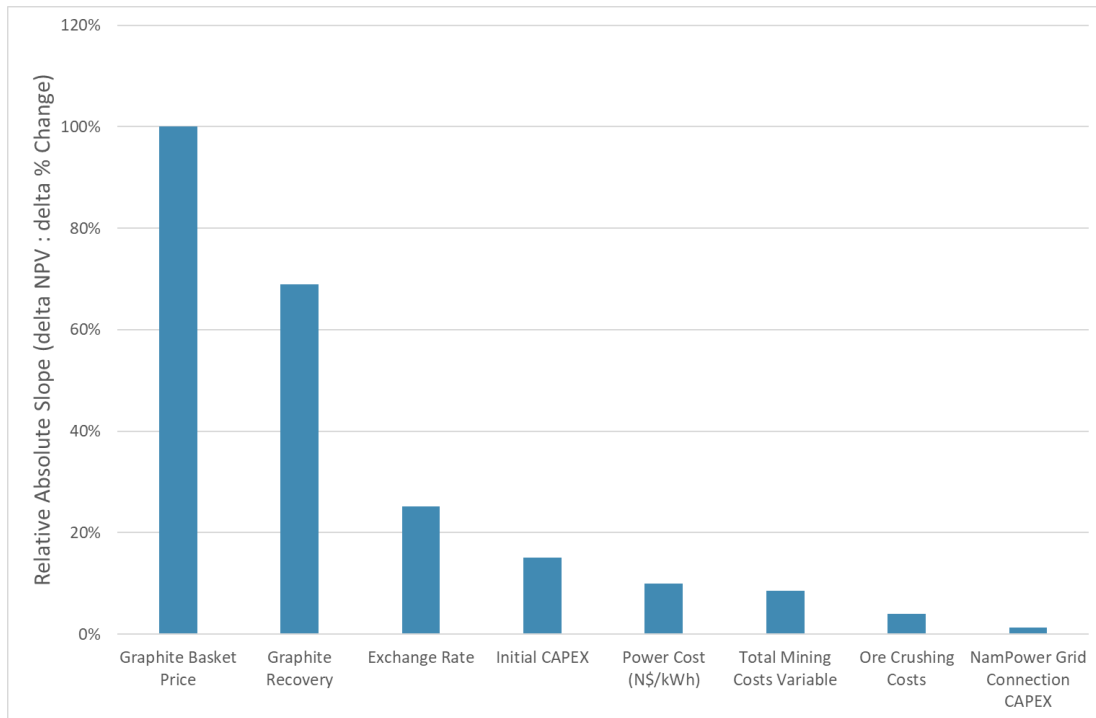


Figure 22-3: Sensitivity Pareto - Relative Absolute Slope

Based on the high sensitivity towards graphite pricing, mining and processing strategies should be aimed at maintaining graphite grade above 95%TGC and maximizing flake size above all else to ensure that optimal graphite price is achieved.

While the project NPV is also highly sensitive towards graphite recovery, the probable range of variation in recovery is relatively small compared to probable changes in graphite pricing.

The Project NPV is relatively sensitive towards negative changes in exchange rate. However, this is unlikely to occur in the short term (<5 years). Based on recent trends the exchange rate is expected to become more favourable. The full effect of major currency exchange rate changes is complex and beyond the scope of the current level of economic analysis.

Initial project capital cost (Process Plant and Infrastructure) impact on NPV is ranked 4th overall in terms of the parameters analysed. Advanced studies will need to pursue capital reduction strategies during the value engineering stages.

Power costs exhibited relatively high sensitivity towards the project NPV. This is driven by the first 4 years of hybrid solar and diesel generated power provided by independent power providers (IPP's), which reduces after year 4 due to connection to the NamPower grid. The main power consumers are the milling and tailings dewatering circuits.

Mining costs are based on the 2022 tender enquiry process. If fuel prices continue to rise, mining costs will be negatively affected and relative sensitivity of mining costs on project NPV may change.

22.5 TAXES, ROYALTIES AND OTHER INTERESTS

22.5.1 Namibian Corporate Income Tax for non-Diamond Mining Companies

The Namibian Corporate Income Tax rate for non-Diamond Mining Companies is 37.5% and was applied as such in this economic analysis.

22.5.2 Value Added Tax

Any legal entity with an annual taxable turnover of greater than N\$500,000 is required to register for Value Added Tax (VAT). Value-added tax is payable on the taxable value of all goods sold or imported. The standard rate is 15%. Direct exports of goods and services are zero-rated, and such VAT was not considered as part of this study. However, VAT rebates will be applicable on services and goods purchased within Namibia. The effect of VAT rebates was not considered in this economic analysis.

22.5.3 Withholding Tax Payable and Double Taxation Agreements

Namibia has Withholding Tax and Double Taxation Agreements with certain countries. However, Canada does not fall in the list of treaty countries and as such this was not considered in this economic evaluation.

22.5.4 Depreciation

Cost recovery for capital invested was estimated using standard depreciation schedules. The estimated cost recovery for calculation of Namibia income tax consisted of a 3 years 100% declining balance calculation, based on initial and expansion capital.

22.5.5 Royalties

Royalties are levied in terms of the Namibian Prospecting and Mining Act as a percentage of the market value of the minerals extracted by licence holders in the course of finding or mining any mineral or group of minerals.

Graphite concentrate will fall into the group of minerals “Semi-precious stones/Industrial metals/Non-Nuclear fuel minerals” for which the Royalty rate is 2%. Royalties were considered on this basis in the economic evaluation.

23 ADJACENT PROPERTIES

(Taken from the MSA Group Report Technical Report titled “Northern Graphite Corporation, Okanjande Mineral Resource Estimate, Namibia” with effective date 29 November 2021)

The Okanjande Mining Licence (ML196) is surrounded by EPL 4717, which is also held by Imerys-Gecko (35,050 ha). Adjacent to this licence, to the northwest, is EPL (7123) held by Cobe Investment CC (valid until 08/10/2023) for a range of minerals including industrial minerals (20,966 ha). To the north is EPL (5847) held by Kunene Resources (Pty) Ltd for precious stones and industrial minerals (6,869 ha). This EPL is valid to 15/07/2022.

To the southeast is an EPL held by Josua Neshuku, EPL 7113 which expired in August 2021. To the south-east is EPL 6734 held by Osino Gold Exploration and Mining (Pty) Ltd which expires on 09/06/2023. All adjacent properties are EPLs granted for precious minerals, dimension stone and industrial minerals.

To MSA’s knowledge, there are no other mining licences for graphite in the region and/or there is no reported economic graphite mineralization on the adjacent licences.

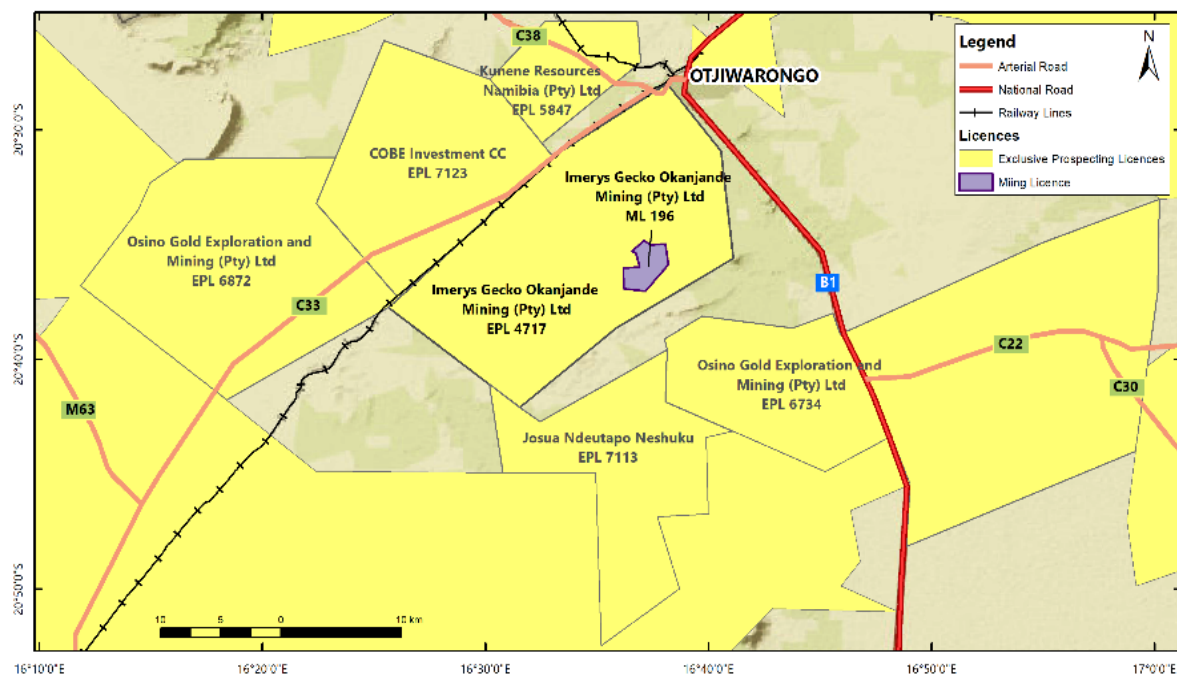


Figure 23-1: Adjacent Properties (Source:Namibia Felxicadastre)

24 OTHER RELEVANT DATA AND INFORMATION

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

None of the contributors are aware of any additional relevant data that might materially impact the interpretation and conclusion of this PEA Report

25 INTERPRETATIONS AND CONCLUSIONS

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

This report was prepared by a group of independent consultants, all qualified persons as defined by NI 43-101, to demonstrate the economic viability of open pit mining and processing, based upon the estimated Mineral Resources at the Okanjande Project (MLP). This report provides a summary of the results and findings to the level that would be expected for a Preliminary Economic Assessment. Standard industry practices and assumptions have been applied in this PEA.

This report is based on all available technical and scientific data available as of 30 June 2023, the effective date of this PEA Report.

25.1 PROJECT STUDY

This PEA study is preliminary in nature and is based on technical and economic assumptions which will be evaluated in more advanced studies. The PEA is based on the Mineral Resource estimate in Chapter 14 (effective date December 2021). The PEA is based on a production plan that includes material in Measured and Indicated classifications from the Okanjande Resource model. This report validates that the Project would support an open pit mining operation to recover graphite from the ground to be sold at a profit.

Estimated Mineral Resources were assumed to be processed with commonly utilized graphite recovery methods. Upgrading of graphite mineralized material through flotation process are commonly applied.

Under the base case assumptions for the Project, the PEA indicates an undiscounted pre-tax cash flow of US\$239M, and a post-tax NPV at 8% discount rate of US\$70.2M. The resulting post-tax IRR is 36% for an initial capital investment of US\$34.6M.

The updated project plan produces upside in terms of cashflow when compared to the previous un-discounted cashflow of US\$194.3M. The post-tax NPV of US70.2M compares well to the previous post-tax NPV of US\$65.1M. The lower IRR of 36% compared to the previous study IRR of 62% is the result of increase CAPEX related to the relocation of the Okorusu plant and new site infrastructure requirements. The lower IRR of the current study is offset by the benefit of reduced future relocation and site infrastructure costs related to planned expansion of operations.

The results of sensitivity analyses of post-tax cash flow and post-tax IRR show that the project is most sensitive to graphite price and recovery. Breakeven graphite price is US\$976.90/t and post-Tax NPV doubles at US\$2,152.80/t.

The base case assumptions demonstrate that the Project would produce an average of 31,103 tonnes per year of graphite concentrate at an average grade of 95% contained carbon as graphite over life-of-mine. The Project will produce 311,031 tonnes of graphite concentrate over the 10-year life-of-mine.

25.2 GEOLOGY AND EXPLORATION

An Independent Technical Report relating to the estimation of mineral resources was previously prepared and compiled by MSA under the supervision of the QPs at the request of Northern Graphite Corporation. This Independent PEA Report has been prepared in accordance with the provisions of National Instrument 43-101 Standards of Disclosure for Mineral Projects.

The PEA Report provides a summary of the Mineral Resource estimate based on the drilling campaign undertaken up to September 2018. The Okanjande graphite deposit has demonstrated its potential for a large tonnage variable grade graphite deposit. The continuity of the deposit is well-defined. The current Mineral Resource is reported within pit shells.

The Mineral Resource Estimate demonstrates the potential of the Okanjande Project.

25.3 MINERAL RESOURCE

The mineral resource estimate was completed in November 2021. The high-level economic assessment used to determine RPEEE was based on 2021 cost estimates for mining and operations. The current PEA update for the processing of graphite mineralized material through the Okanjande processing plant is based on Q1 2023 mining and processing costs. Thus, the RPEEE does not account for the reduction in operational costs due to elimination of the need for mineralized material transport. In addition, graphite prices increased from the 2021 study to the current 2023 study, which is offset to some degree by increases in other operational costs.

25.4 MINERAL PROCESSING AND METALLURGY DEVELOPMENT

The mineral process and metallurgy for this project has been developed since the late 1980's. While the process flow implemented for the project by Imerys-Gecko encountered difficulties achieving nameplate throughput and concentrate grade, the flowsheet was not based on successful test work conducted previously.

The mineral processing test work program completed at SGS in 2021 indicated that the modifications made to the flow sheet could improve on the process flow sheet implemented by Imerys-Gecko and corresponded well with early test work and pilot campaigns conducted in the 1990's.

Based on the latest test work conducted improved recovery and flake distribution was achieved when compared to the test work conducted by MINTEK in the 1990's and 2015.

25.5 MINING

This 10-year mining plan (4%TGC cut-off) considers the mining of weathered as well as fresh graphite bearing material. The IWD design is suitable for all waste and ore rock types. All testing to date indicates that the same processing flowsheet can be used for both fresh and hard rock.

If the cut-off grade is increased, the above adjustments to process the fresh graphite bearing material will be required sooner.

25.6 PROCESSING

A new comminution circuit will provide sufficient capacity to liberate graphite flake from the Okanjande mineralized material for down-stream processing through the Okanjande flotation circuit. Modifications to the original Okorusu graphite flotation flow sheet will correct the processing difficulties experienced by Imerys-Gecko and the flow sheet was tested and verified through a test work program completed at SGS in 2021. Tests indicated that the new modified process flow is expected to exceed performance in terms of recovery and concentrate grade, compared to the Imerys-Gecko flow sheet as well as that of the MINTEK flow sheet developed in the 1990's.

The new tailings dewatering circuit will produce tailings with a residual moisture content below 15%w/w which will be suitable for dry co-disposal on the IWD facility.

The existing concentrate dewatering, drying, classification and bagging plant constructed by Imerys-Gecko is fit-for-purpose and will require no modifications or upgrades, other than relocation to the Okanjande site, to achieve nameplate throughput and performance.

25.7 ENVIRONMENTAL STUDIES AND PERMITTING

The Okanjande site has an ECCs in place in terms of the Environmental Management Act. It is necessary to amend both the ECC to incorporate the proposed changes of the project. This will first and foremost entail engagement with the MEFT as the regulating agency. The process to follow will entail two separate Scoping Studies with impact assessments of the new activities, and an updated EMP.

The impact assessment and EMPs should be informed by appropriate specialist studies where necessary. In this regard, a Geohydrological study should be included, as well as an Air Quality Study and Social Impact Assessment. The amendment process should also undergo a public participation process as required by the EMA.

25.8 CAPITAL AND OPERATING COSTS

The cost estimates were developed to support a nominal mining rate of 631,450t/a. The process facility cost estimate is based on a nominal throughput capacity of 631,450t/a and producing on average 31,103t/a of upgraded and bagged graphite flakes per year. The cost estimates have been prepared in accordance with the recommended practices of the American Association of Cost Engineers (AACE) and are considered to have an accuracy of +40 -30%.

The Capital Cost estimate was developed based on a combination actual costs and firm tenders, material take-offs and factored quantities and costs, semi-detailed unit costs and work packages for major equipment supply. Operating costs estimates derived from quotes for contractor mining, crushing and mineralized material transport as well quotes for reagents, consumables, fuel, power and water provision. Consumption values were based on mass balance, load lists and historic process consumptions from the Imerys-Gecko processing period.

25.9 ECONOMIC ASSESSMENT

This PEA confirms that the project presents very robust economics and will generate a product for which there is an increasing demand that is likely to continue well into the future. Should graphite prices continue to increase as a result of this demand the positive economic benefit of this project can be expected to improve even further.

25.10 RISKS AND UNCERTAINTIES

The current design of these facilities is considered to meet PEA requirements. The qualified persons have reviewed areas critical to a successful project to identify key risks. General risks associated with mining projects include but are not limited to:

- General business, social, political, regulatory and business competition;
- Change in project parameters as development plans are advanced;
- Labour costs and other costs of production;
- Lower graphite price;
- Compliance with laws and regulations or other regulatory requirements;
- Availability of management, technical and skilled operations personnel.

Identified Project risks are listed in specific items that follow with a relative risk rating. The identified risk areas are not ordered in rank of importance due to the relative early stage of development:

- **Geology** – resource quantity and quality (grade) – generally low risk. The risks identified include lack of local accuracy measurements in some of the data due to lack of CRM, variability of density in the weathered zone, and poor accuracy for sulphur assays based on the CRM results for sulphur.
- **Mining** – pit slope geotechnical conditions – considered low risk due to this being a brownfields project with approximately 2 years of mining completed at the currently considered mining parameters.
- **Processing** – recent test work data indicates good graphite recovery and concentrate grade characteristics - considered moderate risk due to lack of continuous closed-circuit testing for the modified process flow sheet. Piloting recommended for future studies.
- **Processing** – The Integrate Waste Dump (IWD) for co-deposition is based on 100% of waste and tailings being co-deposited on the IWD with mobile equipment. No allowance is made for additional layer works or blending of material types. Geotechnical analysis of the proposed IWD is required during advanced studies. Considered high risk based on the level of detail at the current study stage.
- **Permitting** – New co-deposition IWD will require new environmental permitting. Moderate Risk based on aggressive project schedule.
- **Environmental** – The interim report for geo-chemical testing on Okanjande mineralized material and waste rock indicated (Knight Piesold Consulting, 2022) that transitional and fresh material is acid generating and that weathered material is potentially acid generating. The conceptual lined co-deposited IWD is planned to be constructed on a natural rock formation of marble which should neutralize any weak acid that might form. The tailings and waste storage co-deposition facility represents best industry practice.
- **Construction Schedule** – preliminary integration of permitting and construction planning – moderate risk due to low level of detail in construction and permitting schedule.
- **Capital Cost** – based on actual costs from Chinese suppliers and detailed designs at this stage – low risk due to detailed designs and fixed quotes.
- **Capital Cost** – Tailings geochemistry test work was concluded. The fresh mineralized material tailings and waste rock will require a lining system and solution recovery. Moderate risk due to the high-level definition and design of the Phase 2 IWD lining and run-off collection and storage system.
- **Operating Cost** – Power cost. The current power supply strategy employed is a hybrid Solar PV, BESS and Diesel generation for the first four years, after which the diesel and BESS generation is replaced with grid power. Moderate risk due to current level of certainty regarding the project schedule for grid power connection.
- **Operating Cost** – Diesel Pricing. Moderate Risk short-term risk due to current global fuel price increase trends.
- **Land** – Existing mining claims. – low risk.

25.11 **OPPORTUNITIES**

- **Mining** – Potential to reduce costs through a competitive bidding process.
- **Resource** – The Okanjande deposit hosts a hard rock resource of 24,200,00t of mineralized material with 1,287,000t of contained graphite in the measured and indicated category and 7,200,00t of mineralized material with 359,000t of graphite in the inferred category (@3.1%TGC cut-off grade and \$1,200/t Graphite price). The Company may wish to undertake a Preliminary Economic Assessment with respect to expansion of the new processing plant at the mine site. This may reduce operational costs due to economies of scale. The currently site layout makes provision for such future expansions.

- **Exploration** – There is evidence of significantly more mineralisation in the region, but this need to be quantified by exploration and drilling.
- **Processing** – Consider flash flotation in milling circuit for increased large flake recovery.
- **Processing** – Investigate Rutile recovery from graphite flotation tails.
- **Processing** – Investigate concentrate upgrading technologies: Chemical and Physical treatments of concentrate to remove impurities.

26 RECOMMENDATIONS

(Prepared by CREO Engineering Solutions (Pty) Ltd.)

This PEA study is preliminary in nature and is based on technical and economic assumptions which need to be evaluated in more advanced studies. The detail developed is sufficient for a PEA level study. However, additional work would be required to meet a pre-feasibility level.

26.1 MINING AND GEOLOGY

It is recommended additional work continue to further define the deposit as outlined below:

- Assaying of the unsampled drillholes which intersected mineralization;
- Further drilling to determine the margins of the resource and evaluate regional exploration targets;
- Mining inputs and Modifying Factors are to be sourced for the deposit;
- Optimisation of the Mineral Resource to be undertaken;
- Mine design and schedule to be produced; and
- Mineral Reserve to be estimated in the medium-term.

The cost for the additional exploration and Mining studies is estimated at \$385K and \$77K respectively.

26.2 METALLURGY AND PROCESSING

The current PEA is based on the outcomes of historic mineral processing and a recent bench scale flow sheet confirmation test program. It is recommended to complete additional test programs that include closed circuit or locked-cycle tests as well as continuous process testing. Samples should be collected to represent the expected mill feed for the first 1-2 years of operation to determine the metallurgical response. Locked-cycle tests are estimated at \$35k.

It is recommended to leverage the existing onsite metallurgical laboratory facilities to set up flotation testing as well analytical assaying of both process and metallurgical test work samples. The onsite analytical lab will also be required to provide the assay certificates for the final concentrates of the commercial plant. A reconciliation of available site laboratory equipment was ongoing at the time of writing of the PEA Report.

26.3 ENVIRONMENTAL

It is recommended the following work be undertaken in connection with environmental activities:

- Continue on-going consulting and environmental studies required to support permitting requirements and to optimize the site layout.
- Identify environmental requirements for site closure and estimate the cost. It is recommended that a full and dedicated review of the permit register be undertaken as the project moves from the PEA to the next phase, as some permits are valid for certain time periods or addendums to existing permits require approval.
- Define mine closure plans.

The cost for the additional environmental studies is estimated at \$100K.

26.4 CAPITAL AND OPERATING COSTS

The following studies are recommended in to be undertaken to improve estimate confidence levels and follow up on potential cost saving opportunities. These studies are estimated at \$40k to complete.

- Owner vs. Contract Crushing on Site
- Alternative packaging options


27 REFERENCES

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28 APPENDICES

28.1 WATER ABSTRACTION PERMITS



REPUBLIC OF NAMIBIA

MINISTRY OF AGRICULTURE, WATER AND LAND REFORM

Telephone: (061) 2087222	Department of Water Affairs
Fax: (061) 2087697	Private Bag 13193
Enquiries: C. Pieters	Windhoek
Reference: PD 299	9000

Imerys Gecko Okanjande Mining (Pty)Ltd
P O Box 81307
Olympia
Windhoek

APPLICATION FOR THE RENEWAL OF PERMIT NO. 10946 FOR THE ABSTRACTION OF GROUNDWATER FOR MINING PURPOSES FROM ON THE FARM DOORNLAAAGTE NO. 299, OTJIWARONGO DISTRICT

1. The above-mentioned application has been approved. Attached please find permit number 10946 which authorizes the abstraction of water for the mining purposes.
2. You are kindly requested to comply with all the permit conditions, especially conditions number 4 and 5.
3. Please be informed that a high amount of over abstraction of the given quota can lead to the withdrawal of the permit.



Percy W Miska
EXECUTIVE DIRECTOR

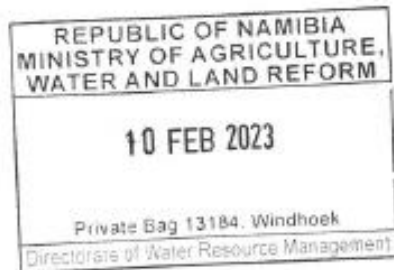
All official correspondence must be addressed to the Executive Director

Figure 28-1: Water Abstraction Permit - Farm Doornlaagte



10946 Dornberg
#10490
COPY
COPY

FOR OFFICIAL USE ONLY
FILE REFERENCE NO.
DATE RECEIVED
CASE NO.



Name: Northern Graphite Okanjande
Mining (Pty) Ltd
Title
Postal address of applicant:
P O Box 424, Otjiwarongo,
Namibia
Telephone: +264 67 304 699
Area code: 067
Date: 07/02/2023

The Under Secretary
Department of Water Affairs
Division: Law Administration
Private Bag 13193
WINDHOEK

APPLICATION FOR A PERMIT TO UTILIZE A CONTROLLED WATER SOURCE

1. This application is for authorization (Delete which is not applicable)
 - 1.1 To utilize water for miner water supply purposes from an existing water source or sources, which had already received a permit for use before, but where the permit will lapse later this year, on 30 June 2023.
2. A complete application form is attached for your consideration.



SIGNATURE OF APPLICANT



REPUBLIC OF NAMIBIA

MINISTRY OF AGRICULTURE, WATER AND LAND REFORM

Telephone: (061) 2087228

Fax: (061) 2087697

Enquiries: J N Mouton

Reference: PD 145

Department of Water Affairs

Private Bag 13193

Windhoek

9000

The Managing Director
Imerys Gecko Okanjande Mining (Pty) Ltd
P O Box 81307
Olympia
WINDHOEK



Dear Sir

APPLICATION FOR THE RENEWAL OF PERMIT NO. 11163 FOR THE ABSTRACTION OF GROUNDWATER FOR MINING PURPOSES ON THE REMAINING PORTION (OSDAM) OF THE FARM OKANJANDE NO. 145, OTJIWARONGO DISTRICT

1. The above-mentioned application has been approved. Attached please find permit number 111 63 which authorizes the abstraction of groundwater for mining purposes.
2. You are kindly requested to comply with all the permit conditions, especially conditions number 4 and 5.



Percy W. Misika
EXECUTIVE DIRECTOR



REPUBLIC OF NAMIBIA

MINISTRY OF AGRICULTURE, WATER AND FORESTRY

Telephone: 2087222
Fax: 2087697

Enquiries: G. Moshosho

Ref. No.: 10/ 7/1/11/2

Department of Water Affairs and Forestry
Private Bag 13193

Windhoek

NAMIBIA


Imerys Gecko Okanjande Graphite Mining (Pty) Ltd
P. O. Box 31307
Olympia
Windhoek
Namibia

Dear Sir

Att: Mr. Krappmann

**APPLICATION FOR A WASTEWATER AND EFFLUENT DISPOSAL EXEMPTION
PERMIT: IMERYS GECKO OKANJANDE GRAPHITE MINING (Pty) Ltd**

1. Your application in the above regard has been approved.
2. Attached is permit number 754
3. You are kindly requested to comply with all the permit conditions.



PERCY. W. MISIKA

EXECUTIVE DIRECTOR

All official correspondence must be addressed to the Executive Director

28.2 LICENSES AND PERMITS

Table 28-1: Mineral Licenses

Permit	Permit #	Site	Company	Authorizing Entity	Validity	Comments
Mining licence	ML196	Okanjande	NG	Ministry of Mines and Energy	08-Feb-42	Annual Licence Fee \$25,000 due 1 st Jan 24
Exclusive prospecting licence	EPL4717	Okanjande	NG		05-Mar-23	Renewal submitted 6/12/22
Export Permit		Okanjande	NG			Applied for as and when required

Table 28-2: Water Licenses

Permit	Permit #	Site	Company	Authorizing Entity	Validity	Comments
Groundwater Abstraction Permit Doornlaagte	10946	Okanjande	NG	Ministry of Agriculture, Water and Forestry	30-Jun-23	Renewal in progress – 6 months prior to expiry
Effluence Disposal Permit (mine offices and workshops)	754	Okanjande	NG		07-Aug-24	Valid – Renewable 3 months prior to expiry
Water abstraction and conveyance permit	11163	Okanjande	NG		29-Jun-23	Renewal in progress – 6 months prior to expiry
Water abstraction and conveyance permit	31575	Okanjande	NG		06-Apr-18	BH not in use

Table 28-3: Environmental and Forestry Permits

Permit	Permit #	Site	Company	Authorizing Entity	Validity	Comments
Species and Forestry removal permit				Ministry of Agriculture, Water and Forestry		Applied for as and when required
ECC (Env. Clearance Certificate) ML 196	11/11/2021	ML196 Okanjande	NG	Ministry of Environment and Tourism	11-Nov-24	Valid – Renewable 1 month prior to expiry
ECC (Env. Clearance Certificate) EPL4717	18/9/2017	EPL 4717 Okanjande	NG	Ministry of Environment and Tourism	20 Feb 25	Valid – Renewable 1 month prior to expiry
Environmental Impact Assessment (Sept 2014)		Okanjande	NG	Author: Enviro Dynamics		Valid – Updates/New EIA required for Power Plants set-up.

Table 28-4: Accessory and General Permits

Permit	Permit #	Site	Company	Authorizing Entity	Validity	Comments
Accessory works permit		ML196 Okanjande	NG	Ministry of Mines and Energy		Application in progress. Layout sent to Oliver.
Diesel installation Certificate		ML196 Okanjande	NG	Ministry of Mines and Energy		Application required once we have a fuel supply contract, detailed designs of tanks and layout.
Petrol installation Certificate		ML196 Okanjande	NG	Ministry of Mines and Energy		Application required once we have a fuel supply contract, detailed designs of tanks and layout.
Used oil permit		ML196 Okanjande	NG	Ministry of Mines and Energy		Application required once we have a contractor, detailed designs and layouts.
Explosives Magazine Permit		ML196 Okanjande	NG	Ministry of Safety and Security		Application required in future.

28.3 ENVIRONMENTAL CLEARANCE CERTIFICATES

ECC – 01730

Serial: yc8pPd1730



REPUBLIC OF NAMIBIA
MINISTRY OF ENVIRONMENT, FORESTRY AND TOURISM

OFFICE OF THE ENVIRONMENTAL COMMISSIONER

ENVIRONMENTAL CLEARANCE CERTIFICATE

ISSUED

In accordance with Section 37(2) of the Environmental
Management Act (Act No. 7 of 2007)

TO

Imerys Gecko Okanjande Mining (Pty) Ltd.
P. O. Box 81307, Olympia, Windhoek.

TO UNDERTAKE THE FOLLOWING LISTED ACTIVITY

The Proposed Continuation, Development and Expansion of the Okanjande Graphite Mine on ML196 held by Imerys Gecko Okanjande Mining, including the Processing of Graphite Ore and By-Products, Accessory Works Operations and the Storage and Dispensing of Fuel within the Mining Licence, Otjozondjupa Region.

Issued on the date: 2021-11-11

Expires on this date: 2024-11-11

(See conditions printed over leaf)



Reduce
Reuse
Recycle



This certificate is printed without erasures or alterations

ECC –

CONDITIONS OF APPROVAL

1. This environmental clearance is valid for a period of 3 (three) years, from the date of issue unless withdrawn by this office
2. This certificate does not in any way hold the Ministry of Environment and Tourism accountable for misleading information, nor any adverse effects that may arise from these activities. Instead, full accountability rests with the proponent and its consultants
3. This Ministry reserves the right to attach further legislative and regulatory conditions during the operational phase of the project
4. All applicable and required permits are obtained and mitigation measures stipulated in the EMP are applied particularly with respect to management of ecological impacts.
5. Strict compliance with national heritage guidelines and regulations is expected throughout the life-span of the proposed activity, therefore any new archaeological finds must be reported to the National Heritage Council for appropriate handling of such.

Figure 28-2: Okanjande ECC

ECC – 01996

Serial: OZE7Ln1996



REPUBLIC OF NAMIBIA
MINISTRY OF ENVIRONMENT, FORESTRY AND TOURISM

OFFICE OF THE ENVIRONMENTAL COMMISSIONER

ENVIRONMENTAL CLEARANCE CERTIFICATE

ISSUED

In accordance with Section 37(2) of the Environmental
Management Act (Act No. 7 of 2007)

TO

Imerys Gecko Okanjande Mining (Pty) Ltd
P. O. Box 81307, Windhoek

TO UNDERTAKE THE FOLLOWING LISTED ACTIVITY

**The Exploration of Graphite in the Okanjande farm area on Exclusive
Prospecting License (EPL) 4717, Otjiwarongo District,
Otjozondjupa Region.**

Issued on the date: 2022-02-20

Expires on this date: 2025-02-20

(See conditions printed over leaf)



Reuse
Recycle



This certificate is printed without erasures or alterations

ECC –

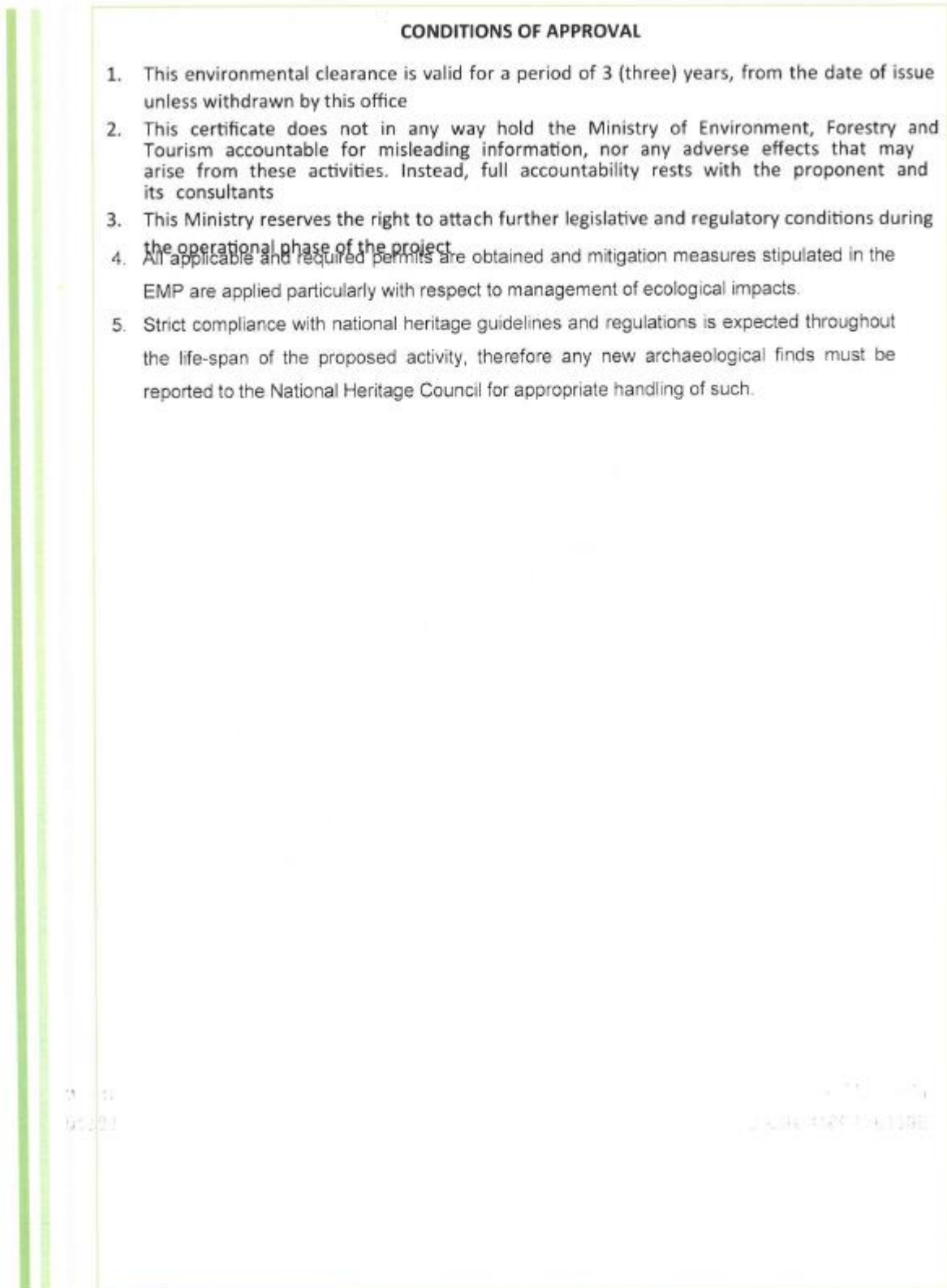


Figure 28-3: Okanjande ECC

28.4 ACCESSORY WORK PERMITS



NORTHERN GRAPHITE

COPY
COPY

Date: 21 February 2023

The Executive Director
Ministry of Mines and Energy
1 Aviation Road
Private Bag 13297
Windhoek
Namibia



Attention: The Mining Commissioner, Mrs. I. Chirchir
The Mine Surveyor, Mr. M. Endjala



Dear Madame and Sir,

Application for Accessory Works Area: Mining Licence 196

The meetings and discussion held between Mr Endjala and our Oliver Krappmann refers. Northern Graphite Okanjande Mining (Pty) Ltd, formerly Imerys Gecko Okanjande Mining, as the holder of ML-196, hereby would like to apply to the Mining Commissioner's office for an Accessory Works Permit in terms of Section 90(3) of the Minerals Act 33 of 1992. The map, **Attachment-1**, to this letter gives details with description and positions of the individual Accessory Works components and their relation to the linear infrastructure, all concerning the company's graphite mining and processing operations at Okanjande.

Thank you for your assistance. We trust that you find this in order. Any queries regarding the matter can be directed to the undersigned.

Yours faithfully,



Gerson Shipena
General Manager
Mobile: +264811409590
Email: gshipena@northerngraphite.com

Acknowledge receipt:



Name:
Position:
Date:

28.5 MINING LICENCE

**REPUBLIC OF NAMIBIA
MINISTRY OF MINES AND ENERGY**

MINING LICENCE

(Issued in terms of Section 93 of the Minerals (Prospecting and Mining) Act, 1992)

Mining Licence No Office Reference No

Subject to the provisions of the Minerals (Prospecting and Mining) Act, 1992, this Mining licence is
Hereby issued to

Full Name of Licence Holder

Identity/Passport No or Company Registration No

Address (natural person) or Registered Address (company)

Full Name of Accredited Agent (if applicable)
Address of Accredited Agent (if applicable)

for the period from (date of to (date of expiry)
of issue)

unless abandoned or cancelled on any prior date, or extended to such later date as may be endorsed on this licence in the event that this licence is renewed.

This Mining licence is issued in respect of

Name of Mineral(s)/Group(s) of Minerals

over a certain portion of land situate in Region(s)

Registration Division(s) Magisterial District(s)

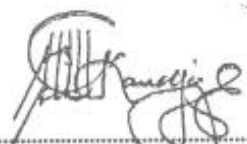
as more fully depicted in the attached diagram No signed by the Commissioner

and is further subject to the terms and conditions contained in the notice of the Minister's intention to grant the

licence dated and agreed to in writing by the applicant on

as appended hereto.


Signed at WINDHOEK this day of


.....
MINISTER OF MINES AND ENERGY



OFFICIAL STAMP

28.6 EXPLORATION LICENSE



MINISTRY OF MINES AND ENERGY
ENDORSEMENT (RENEWAL)

With this approval of the Minister of Mines and Energy, this mineral licence has been renewed for a period of

Two (2)

.....years from

09 March 2021 to 09 March 2023

09.03.2021

Shine

Commissioner

REPUBLIC OF NAMIBIA
MINISTRY OF MINES AND ENERGY

Exclusive Prospecting Licence
(Issued in terms of Section 70 of the Minerals (Prospecting and Mining) Act, 1992)

Exclusive Prospecting Licence No Office Reference No

Subject to the provisions of the Minerals (Prospecting and Mining) Act, 1992, this exclusive prospecting licence is hereby issued to

Full Name of Licence Holder

Identity/Passport or Company Registration No

Address (natural person) or Registered Address (company)

Full Name of Accredited Agent (if applicable)

Address of Accredited Agent (if applicable)

for the period of from To
(date of issue) (date of expiry)

unless abandoned or cancelled on any prior date, or extended to such later date as may be endorsed on this licence in the event that this licence is renewed.

This exclusive prospecting licence is issued in respect of


Name of Mineral(s)/Group(s) of Minerals

over a certain portion of land situate in Region(s)


Registration Division(s) Magisterial District(s)

as more fully depicted in the attached diagram No signed by the Commissioner and is further subject to the terms and conditions contained in the notice of the Minister's intention to grant the licence dated and agreed to in writing by the applicant on as appended hereto.

Signed at WINDHOEK this day of



MINISTER



OFFICIAL STAMP



Northern Graphite Okanjande Mining (Pty) Ltd
P.O. Box 424
Otjiwarongo, Namibia

The Mining Commissioner
Ministry of Mines and Energy
1 Aviation Road
Private Bag 13297
Windhoek, Namibia

Att: Mrs I. Chirchir
Cc: Mrs F. Flavianu

5 December 2022

Dear Madam,

Re: Application for renewal of EPL4717 without size reduction

Attached hereto kindly receive the renewal application and technical reporting for EPL-4717, located at Otjiwarongo in the Otjozondjupa region.

Northern Graphite Okanjande Mining, in May 2022, concluded an agreement to take over the Okanjande graphite project from the Imerys-Gecko JV. Exploration work conducted thus far include an EM survey over the entire area of the EPL, geological mapping, diamond and RC drilling, geochemical and mineralogical analysis of exploration samples, various processing tests and water search. Application is hereby made for re-issuing of the license for a period of two years in its present extent of 35 050.49 hectares.

We request the renewal without reduction in size for the following reasons:

- The Covid-19 pandemic substantially prolonged Northern Graphite's due-diligence process and take-over of the project and this EPL from the previous holder, and was only finalised in May 2022 and the new company only registered in July 2022.
- The mine is presently still in the state of care and maintenance since November 2018. With no income from production, coupled with the effect of the global pandemic, the project experienced budget constraints that limited the exploration work.
- Exploration conducted thus far by the previous holder successfully generated a number of potential targets on this EPL which couldn't be followed up, due to the limited time, budget and Northern Graphite's focus on expediting the refurbishment and commissioning of its graphite processing plant. Since take-over, Northern Graphite spent about N\$35 million on the project.

Northern Graphite Okanjande Mining (Pty) Ltd | Reg. No: 2010/0771
Okanjande Mine, Farm Highlands 311, Otjiwarongo, Namibia |
P.O. Box 424, Otjiwarongo, Namibia
Directors: GB Bowles (Canadian), N Pilon (Canadian)

northerngraphite.com

- The ground water search for processing requirements, initiated by the previous holder of the EPL, still requires finalization.

I trust that the above reasons are received favorably and support our request for renewal of the license without a reduction in size. Your kind consideration of the matter will be highly appreciated.

Should additional information be required, please do not hesitate to contact me.

Yours faithfully,

Daniel M Machoko
Surface Operations Manager

Attachments:

- Renewal Application for EPL 4717, incl. attachments.
- Report for EPL 4717 – in duplicate hard-copy and in digital copy on memory stick.



REPUBLIC OF NAMIBIA


MINISTRY OF MINES AND ENERGY

**RENEWAL APPLICATION FOR AN
EXCLUSIVE PROSPECTING LICENCE
MINERALS (PROSPECTING & MINING) ACT, 1992, SECTIONS 68
& 72
(MINIMUM REQUIREMENT)**

Handwritten:
06/12/2022
checked: [Signature]
06/12/2022

Date	5 December 2022	Informal name of area	Okatjemunde
------	-----------------	-----------------------	-------------

For Office Use

Reference	14/2/1/4/2/ 4717	
Deposit	N\$ 20,000-00	
Receipt no.	Eft Payment	
GIS sign		
date		
TAM sign		
date		

SECTION A – PARTICULARS OF APPLICANT

(i) Natural person

Full names	n/a		
Nationality *		Date of birth	
Telephone		Facsimile	
Residential address	Postal address		

28.7 NAMIBIAN ENERGY COSTS

SCHEDULE OF APPROVED TARIFFS (2021/2022)

NAMPOWER (PTY) LTD

TIME OF USE TARIFF SCHEDULE

TYPE	CUSTOMER SERVICE CHARGE	POINT OF SUPPLY CHARGE		MAXIMUM DEMAND CHARGE		NETWORK ACCESS CHARGE	
		N\$/PoS/Month		N\$/kVA	N\$/kW	N\$/kVA	N\$/kW
		No Diversity/ < 10 MW	With Diversity/ > 10 MW	Peak and Standard		All Periods	
Tariff > 33kV	10,250.00	4,950.00	6,720.00	96.37	105.60	89.71	98.31
Tariff =< 33 kV	10,250.00	4,950.00	6,720.00	100.22	109.83	93.30	102.26

TYPE	CHARGES			LEVIES	
	Peak	Standard	Off-peak	NEF LEVY	ECB LEVY
	c/kWh	c/kWh	c/kWh	c/kWh	c/kWh
Energy Tariff > 33kV	141.31	105.98	70.65	1.600	2.120
Energy Tariff =< 33 kV	144.13	108.09	72.07	1.600	2.120
Losses >33kV	15.44	11.58	7.72	-	-
Losses =< 33kV	15.75	11.81	7.88	-	-
Reliability	10.48	10.48	10.48	-	-
Long Run Marginal Cost	-	-	-	-	-

- *Notified Maximum Demand (NMD) Penalty Charge*

The NMD Penalty Charge shall be 100% of the ECB approved NamPower Maximum Demand Charge PLUS Network Access Charge on capacity utilised over and above the customer's contractual NMD, exceeding for three (3) consecutive months, payable as from month three (3).

TIME PERIODS FOR TIME-OF-USE TARIFFS (2021/2022)

Time Periods	Namibian Time		Time Periods
	Peak	Standard	
Day			Off-peak
Week Day	07h00 – 10h00	06h00 – 07h00	00h00 – 06h00
	17h00 – 20h00	10h00 – 17h00	22h00 – 24h00
		20h00 – 22h00	
Saturday		07h00 – 12h00	00h00 – 07h00
		18h00 – 21h00	12h00 – 18h00
			21h00 – 24h00
Sunday			00h00 – 24h00

28.8 MASS BALANCE

Northern Graphite Namibia 80 tph Mass Balance - 5.3% C(g)Head Grade								
	Mass %	Solids		Water t/h	Slurry			
		t/h	S.G		t/h	m3/h	% solids	SG
Rod Mill								
Hydrocyclone Feed	100.0	90.0	2.70	206.7	296.7	240.0	30.33	1.24
Cyclone Overflow	100.0	90.0	2.70	206.7	296.7	240.0	30.33	1.24
Cyclone Underflow	250.0	225.0	2.75	107.7	332.7	189.5	67.63	1.76
Graphite Rougher								
Rougher feed	100.0	90.0	2.70	206.7	296.7	240.0	30.33	1.24
Rougher concentrate	9.7	8.7	2.50	14.4	23.1	17.9	37.72	1.29
Rougher concentrate wash water				2.1				
Rougher concentrate & wash water	9.7	8.7	2.50	16.5	25.2	20.0	34.62	1.26
Rougher tailings	90.3	81.3	2.75	192.3	273.6	221.8	29.71	1.23
Polishing Mill #1								
Polishing mill feed	9.70	8.7	2.50	16.5	25.2	20.0	34.62	1.26
Polishing mill discharge	9.70	8.7	2.50	16.5	25.2	20.0	34.62	1.26
Polishing mill wash water				14.6				
Polishing mill discharge & wash water	9.70	8.7	2.50	31.0	39.8	34.5	21.95	1.15
1st Cleaner Flotation								
1st Cleaner Feed	9.70	8.7	2.50	31.0	39.8	34.5	21.95	1.15
1st Clnr concentrate	6.20	5.58	2.45	9.2	14.8	11.5	37.72	1.29
1st Clnr concentrate wash water				1.3				
1st Clnr concentrate & wash water	6.20	5.58	2.45	10.5	16.1	12.8	34.62	1.26
1st Clnr tailings	3.50	3.15	2.70	21.8	25.0	23.0	12.61	1.09
Polishing Mill #2								
Polishing mill feed	6.20	5.58	2.45	10.54	16.1	12.8	34.62	1.26
Polishing mill wash water				9.3				
Polishing mill discharge & wash water	6.20	5.58	2.45	19.8	25.4	22.1	21.95	1.15
2nd Cleaner Flotation								
2nd Cleaner Feed	6.60	5.94	2.45	30.5	36.4	32.9	16.32	1.11
2nd Clnr concentrate	5.99	5.39	2.35	14.4	19.8	16.7	27.27	1.19
2nd Clnr concentrate wash water				5.1				
2nd Clnr concentrate & wash water	6.09	5.48	2.35	19.5	25.0	21.8	21.95	1.14
2nd Clnr tailings	0.61	0.55	2.55	16.1	16.6	16.3	3.30	1.02
3rd Cleaner Flotation								
3rd Cleaner Feed	6.09	5.48	2.35	24.3	29.8	26.6	18.42	1.12
3rd Clnr concentrate	5.69	5.12	2.35	13.7	18.8	15.8	27.27	1.19
3rd Clnr concentrate wash water				4.6				
3rd Clnr concentrate & wash water	5.69	5.12	2.35	18.2	23.3	20.4	21.95	1.14
3rd Clnr tailings	0.40	0.36	2.35	10.6	11.0	10.8	3.28	1.02
4th Cleaner Flotation								
4th Cleaner Feed	5.69	5.12	2.35	18.2	23.3	20.4	21.95	1.14
4th Clnr concentrate	5.59	5.03	2.35	13.4	18.4	15.6	27.27	1.19
4th Clnr concentrate wash water				4.5				
4th Clnr concentrate & wash water	5.59	5.03	2.35	17.9	22.9	20.0	21.95	1.14
4th Clnr tailings	0.10	0.09	2.35	4.8	4.9	4.8	1.84	1.01
Classification Screen								
Screen feed	5.59	5.03	2.35	17.9	22.9	20.0	21.95	1.14
Screen wash water				10.7				
Screen oversize	3.34	3.01	2.35	1.8	4.8	3.1	62.79	1.56
Screen undersize	2.25	2.03	2.35	26.8	28.9	27.7	7.02	1.04
Coarse Stirred Media Mill								
Coarse Stirred media feed	3.34	3.01	2.35	1.8	4.8	3.1	62.79	1.56
Coarse Stirred media dilution water				10.4				
Coarse Stirred media mill discharge	3.34	3.01	2.35	12.2	15.2	13.5	19.80	1.13
Coarse Stirred media mill wash water				0.0				
Coarse Stirred media discharge & wash water	3.34	3.01	2.35	12.2	15.2	13.5	19.80	1.13

Northern Graphite Namibia 80 tph Mass Balance - 5.3% C(g)Head Grade								
Coarse Cleaner 1 Flotation								
Coarse cleaner 1 feed	3.49	3.14	2.35	16.9	20.0	18.2	15.67	1.10
Coarse cleaner 1 concentrate	3.34	3.01	2.35	6.2	9.2	7.5	32.53	1.23
Coarse cleaner 1 concentrate wash water				4.5				
Coarse cleaner 1 concentrate & wash water	3.34	3.01	2.35	10.7	13.7	12.0	21.95	1.14
Coarse cleaner 1 tailings	0.15	0.14	2.35	10.7	10.8	10.7	1.25	1.01
Coarse Cleaner 2 Flotation								
Coarse cleaner 2 feed	3.34	3.01	2.35	10.7	13.7	12.0	21.95	1.14
Coarse cleaner 2 concentrate	3.19	2.87	2.35	6.0	8.8	7.2	32.53	1.23
Coarse cleaner 2 concentrate wash water				1.7				
Coarse cleaner 2 concentrate & wash water	3.19	2.87	2.35	7.7	10.5	8.9	27.27	1.19
Coarse cleaner 2 tailings	0.15	0.14	2.35	4.7	4.9	4.8	2.77	1.02
Dewatering Flotation								
Dewatering flotation feed	2.25	2.03	2.35	26.8	28.9	27.7	7.02	1.04
Dewatering flotation concentrate	2.25	2.03	2.35	7.2	9.2	8.1	21.95	1.14
Dewatering flotation wash water				4.8				
Dewatering flotation concentrate & wash water	2.25	2.03	2.35	12.0	14.1	12.9	14.39	1.09
Dewatering flotation tailings	0.00	0.00	2.35	19.6	19.6	19.6	0.00	1.00
Fine Stirred Media Mill 1								
Fine stirred mill 1 feed	2.25	2.03	2.35	12.0	14.1	12.9	14.39	1.09
Fine stirred media mill 1 dilution water				0.0				
Fine stirred media mill 1 discharge	2.25	2.03	2.35	12.0	14.1	12.9	14.39	1.09
Fine stirred media mill 1 wash water				0.0				
Fine stirred media mill 1 discharge & wash water	2.25	2.03	2.35	12.0	14.1	12.9	14.39	1.09
Fine Cleaner 1 Flotation								
Fine cleaner 1 feed	2.35	2.12	2.35	16.4	18.5	17.3	11.42	1.07
Fine cleaner 1 concentrate	2.10	1.89	2.35	6.7	8.6	7.5	21.95	1.14
Fine cleaner 1 wash water				2.0				
Fine cleaner 1 concentrate & wash water	2.10	1.89	2.35	8.7	10.6	9.5	17.79	1.11
Fine cleaner 1 tailings	0.25	0.23	2.35	9.7	9.9	9.8	2.27	1.01
Fine Cleaner 2 Flotation								
Fine cleaner 2 feed	2.20	1.98	2.35	11.1	13.1	11.9	15.17	1.10
Fine cleaner 2 concentrate	2.10	1.89	2.35	6.7	8.6	7.5	21.95	1.14
Fine cleaner 2 wash water				2.0				
Fine cleaner 2 concentrate & wash water	2.10	1.89	2.35	8.7	10.6	9.5	17.79	1.11
Fine cleaner 2 tailings	0.10	0.09	2.35	4.4	4.4	4.4	2.03	1.01
Fine Cleaner 3 Flotation								
Fine cleaner 3 feed	2.10	1.89	2.35	8.7	10.6	9.5	17.79	1.11
Fine cleaner 3 concentrate	2.00	1.80	2.35	6.4	8.2	7.2	21.95	1.14
Fine cleaner 3 wash water				1.9				
Fine cleaner 3 concentrate & wash water	2.00	1.80	2.35	8.3	10.1	9.1	17.79	1.11
Fine cleaner 3 tailings	0.10	0.09	2.35	2.3	2.4	2.4	3.71	1.02
Concentrate Pressure filter Feed Tank								
Concentrate Pressure filter Feed Tank	5.19	4.67	2.35	19.0	23.7	21.0	19.73	1.13
Concentrate Pressure Filter								
Pressure filter feed	5.19	4.67	2.35	19.0	23.7	21.0	19.73	1.13
Pressure filter filtrate	0.00	0.00	2.35	18.3	18.3	18.3	0.00	1.00
Pressure filter cake	5.19	4.67	2.35	0.7	5.4	2.7	86.44	1.99
Concentrate Dryer								
Concentrate dryer feed	5.19	4.67	2.35	0.7	5.4	2.7	86.44	1.99
Concentrate dryer offgas	0.00			0.7	0.72	0.7	0.00	1.00
Dried Product	5.19	4.67	2.35	0.0	4.7	2.0	99.69	2.34
Tailings Thickener								
Thickener feed	94.8	85.3	2.66	254.3	339.6	286.4	25.12	1.19
Thickener O/F	0.00		1.00	218.3	218.3	218.3	0.00	1.00
Thickener U/F	94.8	85.3	2.66	40.8	126.2	72.9	67.63	1.73

28.9 GRAPHITE PRICING

NORTHERN GRAPHITE CORPORATION

MANAGEMENT'S DISCUSSION AND ANALYSIS

For the Three and Six Month Periods Ended June 30, 2022

(Information as at August 26, 2022 unless otherwise noted)

The following provides management's discussion and analysis ("MD&A") of results of operations and financial condition of Northern Graphite Corporation ("Northern" or the "Company") for the three and six month periods ended June 30, 2022 and 2021. This MD&A was prepared by the Company's management and approved by its Board of Directors on August 26, 2022.

Average Realized Sales Price Per Tonne of Graphite Concentrate Sold

Average realized sales price per tonne of graphite concentrate sold is used by management and investors to better understand the graphite price realized throughout a period.

<i>(Shown in Thousands of Dollars, Except Tonnes and Per Tonnes Figures)</i>	Three and Six Months Ended June 30, 2022
Gross revenue	\$3,693
Graphite concentrate tonnes sold	1,773
Average realized sales price per tonne of graphite concentrate sold	\$2,083

Figure 28-4: Realized Graphite Pricing Q2 2022, NG Management Discussion and Analysis

NORTHERN GRAPHITE CORPORATION

MANAGEMENT'S DISCUSSION AND ANALYSIS

For the Three and Nine Month Periods Ended September 30, 2022

(Information as at November 25, 2022 unless otherwise noted)

The following provides management's discussion and analysis ("MD&A") of results of operations and financial condition of Northern Graphite Corporation ("Northern" or the "Company") for the three and nine month periods ended September 30, 2022 and 2021. This MD&A was prepared by the Company's management and approved by its Board of Directors on November 25, 2022.

Average Realized Sales Price Per Tonne of Graphite Concentrate Sold

Average realized sales price per tonne of graphite concentrate sold is used by management and investors to better understand the graphite price realized throughout a period.

<i>(Shown in Thousands of Dollars, Except Tonnes and Per Tonnes Figures)</i>	Three months Ended September 30, 2022	Nine months Ended September 30, 2022
Gross revenue	\$ 4,483	\$ 8,176
Graphite concentrate tonnes sold	2,184	3,957
Average realized sales price per tonne of graphite concentrate sold	\$ 2,053	\$ 2,066

Figure 28-5: Realized Graphite Pricing Q3 2022, NG Management Discussion and Analysis