GLENCORE Resources & Reserves as at 31 December 2016

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About this report

We report our resources and reserves in accordance with the 2012 edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code), the 2007 edition (as amended July 2009) of the South African Code for Reporting of Mineral Resources and Mineral Reserves (SAMREC), the Canadian Institute of Mining, Metallurgy and Petroleum (CIM) Standards on Mineral Resources and Reserves (2014 edition) and the Petroleum Resources Management System (PRMS) for reporting oil and natural gas Reserves and Resources.

Overview

The resource and reserve data in the following tables are as at 31 December 2016, unless otherwise noted. For comparison purposes, data for 2015 has been included.

Metric units are used throughout.

All data is presented on a 100% asset basis, with the Glencore attributable percentage shown against each asset, with the exception of Oil assets which are shown on a working interest basis.

All tonnage information has been rounded to reflect the relative uncertainty in the estimates; there may therefore be small differences in the totals.

The Measured and Indicated resources are reported inclusive of those resources modified to produce reserves, unless otherwise noted.

Commodity prices and exchange rates used to establish the economic viability of reserves are based on long-term forecasts applied at the time the reserve was estimated.

Where resources and reserves have not been updated, on the basis that the information has not materially changed since it was reported under JORC 2004, this information has not been updated to comply with the JORC code 2012. Reference is given in the report where this is the case.

Copper

Copper has adopted the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code) as its standard for all public reports of exploration results, Mineral Resources and Ore Reserves.

The Mineral Resources and Ore Reserves Statement has been compiled in accordance with the JORC Code.

The Mineral Resources and Ore Reserves statements for Australia have been reviewed and the relevant data extracted and compiled by Jason Hosken, Glencore Copper.

The Mineral Resources and Ore Reserves statements for each African and South American operation and project have been reviewed, with the data being extracted and compiled by the respective competent person for each operation and project.

Zinc

The Zinc Mineral Resource and Ore Reserve Statement at 31 December 2016 has been compiled in accordance with the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code).

The term 'Ore Reserves', as defined in Clause 28 of the JORC Code, has the same meaning as 'Mineral Reserves' as defined in The Canadian Institute of Mining, Metallurgy and Petroleum (CIM) Definition Standards for Mineral Resources and Mineral Reserves.

The Mineral Resource and Ore Reserve statements have been reviewed and the relevant data extracted and compiled by Ignacio Seebold, Glencore Zinc (ICOG-EurGeol).

Nickel

The Mineral Resources and Mineral Reserves estimates are prepared in accordance with the Canadian Institute of Mining, Metallurgy and Petroleum (CIM) Definition Standards on Mineral Resources and Mineral Reserves, adopted by CIM Council on 10 May 2014, and the CIM Estimation of Mineral Resources and Mineral Reserves Best Practice Guidelines, adopted by CIM Council on 23 November 2003, and have been compiled using geostatistical and/or classical methods, plus economic and mining parameters appropriate to each project.

The Mineral Resource and Ore Reserve estimates at Murrin in Australia have been prepared in accordance with the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code).

The Mineral Resource and Ore Reserve statements have been reviewed and the relevant data extracted and compiled by Steve Kormos, Glencore Nickel.

Ferroalloys

South African chromite, vanadium and PGM's (platinum group metals) Mineral Resources and Ore Reserves in this report were prepared in accordance with the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code).

The Chromite, Vanadium and PGM's Mineral Resource and Ore Reserve Statement at 31 December 2016 is based on the Glencore Ferroalloys "Procedure for the Estimation of Mineral Resources and Ore Reserves". Definitions of all the terms used in this report can be found in the relevant code.

About this report

The Mineral Resource and Ore Reserve statements have been reviewed and the relevant data extracted and compiled by Pieter-Jan Gräbe, Glencore Ferroalloys (SACNASP).

Iron Ore

Iron ore Mineral Resources and Ore Reserves have been compiled in accordance with the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code), unless otherwise stated in the notes for a particular Mineral Resource and Ore Reserve.

Iron Ore Mineral Resources and Ore Reserves have not been re-estimated since 2015.

Coal

Australian, Canadian and Colombian (Prodeco) Coal Resources and Reserves have been prepared in accordance with the 2012 Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC Code).

South African and Colombian (Cerrejón) Coal Resources and Reserves have been prepared in accordance with the 2007 edition (as amended July 2009) of the South African Code for Reporting of Mineral Resources and Mineral Reserves (SAMREC).

The Coal Resource and Reserve Statements as at 31 December 2016 conform to the requirements of these Codes and are consistent with Glencore Coal's internal Coal Resource and Reserve Estimation and Reporting Standard.

Coal resources have been estimated for all coal seams that have reasonable prospects for eventual economic extraction by open cut or underground mining methods within mining leases or exploration licences. In general, Coal Resources are reported within a geoshell limited by the areal and depth extent of the drill holes; i.e. there is very little inclusion of Coal Resources extrapolated beyond the extent of the geological data.

Coal Resources are excluded from those areas where the seam has been extracted or sterilised by mining.

All tonnage information has been rounded to reflect the relative uncertainty in the estimates; there may therefore be small differences in the totals.

Coal Resource and Reserve totals are rounded to appropriate levels of accuracy in accordance with the Glencore Coal rounding procedures. The following table summarises the data rounding assumptions for the 2016 report.

Classification	Tonnage Range	Rounding				
Measured +	< 10Mt	1 significant figure				
Indicated Resources	10Mt - 30Mt	2 significant figures Nearest 5Mt				
/	30Mt - 100Mt					
Proved + Probable Reserves	> 100Mt	2 significant figures				
	>1000Mt	Nearest 50Mt				
	< 100Mt	Nearest 10Mt				
Inferred	100Mt - 400Mt	Nearest 50Mt				
	> 400Mt	Nearest 100Mt				

Coal Reserves are rounded to the same assumptions as Measured and Indicated Coal Resources above. Individual tonnage assessments are added to show Group or Complex tonnages and geographical accumulations. These are not subjected to further rounding.

The Coal Resource and Reserve Competent Person statements have been reviewed and the relevant data extracted and compiled by Jeff Gerard, Glencore Coal.

Oil

Oil and natural gas Resources and Reserves have been prepared in accordance with the PRMS jointly published by the Society of Petroleum Engineers, the World Petroleum Council, the American Association of Petroleum Geologists and the Society of Petroleum Evaluation Engineers, as amended.

The Equatorial Guinea Oil Reserves statement has been reviewed and the relevant data extracted and compiled by Gaffney, Cline & Associates (GCA).

The Chad Oil Reserves statement has been reviewed and the relevant data extracted and compiled by McDaniel & Associates (McDaniel).

The Oil Resources statements for Equatorial Guinea, Chad and Cameroon have been reviewed and the relevant data extracted and compiled by Glencore.

Competent/Qualified Persons

Resource and reserve estimates are based on information compiled by Competent Persons (as defined by the JORC, SAMREC Codes), Qualified Persons (as defined by CIM Definition Standards for Mineral Resources and Mineral Reserves) and Adequately Qualified Persons (as defined by PRMS).

Each of the Competent/Qualified Persons has the appropriate professional membership and the relevant experience in relation to the resources and/or reserves being reported by them to qualify as a Competent or Qualified Person as defined in the relevant code or standard. Each has consented to the inclusion of their resource and reserve estimates in the form and context in which it appears in this report.

Definitions

Throughout this report, the following abbreviations and definitions have been used:

3PGE	Three Platinum Group Elements (Platinum, Palladium and Rhodium)
AIG	Australian Institute of Geoscientists
APEGBC	Association of Professional Engineers and Geoscientists of BC
APEGGA	Association of Professional Engineers Geologists and Geophysicists of Alberta
APEGNB	Association of Professional Engineers and Geoscientists of New Brunswick
APGO	Association of Professional Geoscientists of Ontario
AusIMM	Australasian Institute of Mining and Metallurgy
CIM	Canadian Institute of Mining, Metallurgy and Petroleum
CV (kcal/kg)	Calorific Value, kilo calories per kilogramme
DTC	Davis Tube Concentrate
ECSA	Engineering Council of South Africa
EL	Exploration Licence
FPSO	Floating production, storage and offloading
Geoshell	A broad envelope limited by the depth and areal extent of geological data points (primarily drill holes)
GSL	Geological Society of London
ICOG-EurGeol	Ilustre Colegio Oficial de Geólogos – European Geologist
JORC	Joint Ore Reserves Committee
kt	Thousand tonnes
LOM	Life of Mine
LOX	Limit of Oxidation
LOZ	Lower Oxidised Zone
Mt	Million tonnes
NSR	Net Smelter Return
OC	Open cast or Open cut
OGQ	Ordre des Géologues du Québec
OIQ	Ordre des Ingénieurs du Québec
OR	Ore Reserves
PEO	Professional Engineers of Ontario
PLATO	South African Council for Professional and Technical Surveyors
PRMS	Petroleum Resources Management System
QQ	Quantile quantile plot, a geostatistical method to assess modelled data against actual data
ROM	Run of Mine
SACNASP	The South African Council for Natural Scientific Professions
SAMREC	South African Code for Reporting of Mineral Resources and Mineral Reserves
UG	Underground
UG2	Upper Group No2 Chromitite layer
VMS	Volcanogenic Massive Sulphide

Marketable Coal Reserves (CIM/JORC) and Saleable Coal Reserves (SAMREC) are the tonnage and coal quality that will be available for sale, either in the raw ROM state at specific moisture content or after beneficiation of the ROM Coal Reserve has produced materials at specified qualities, moisture contents and size ranges.

Definitions of many of the terms used in this report can be found in the relevant codes.

African Copper Mineral Resources (Katanga, Mutanda, Mopani)

Name of	Attributab	le Mining		Measured Resou		Indicated Mineral Resources		Measur Indicated F		Infe Mineral R	ompetent	
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16		person
Katanga												
Kamoto	56.4%	UG	(Mt)	12.2	9.5	65.9	72.9	78.1	82.4	48.5	49.2	CS
			Copper (%)	3.90	3.91	3.92	3.84	3.92	3.85	3.83	4.44	
			Cobalt (%)	0.59	0.48	0.46	0.43	0.48	0.43	0.38	0.34	
T17 (OC&UG)	56.4%		(Mt)	4.2	4.2	9.4	9.4	13.6	13.6	5.2	5.2	CS
		UG/OC	Copper (%)	2.66	2.66	4.44	4.44	3.89	3.89	4.21	4.21	
			Cobalt (%)	0.51	0.51	0.65	0.65	0.61	0.61	0.98	0.98	
Mashamba	56.4%	OC	(Mt)	-	-	60.0	26.2	60.0	26.2	18.3	37.2	CS
East			Copper (%)	-	-	1.68	1.51	1.68	1.51	2.85	2.33	
			Cobalt (%)	_	_	0.62	0.71	0.62	0.71	0.47	0.53	
KOV OC,	56.4%	OC	(Mt)	-	-	110.5	71.4	110.5	71.4	78.2	91.0	CS
KOV UG,			Copper (%)	-	-	4.75	4.16	4.75	4.16	4.39	4.44	
KTE			Cobalt (%)	_	_	0.53	0.51	0.53	0.51	0.38	0.36	
Kananga	56.4%		(Mt)	-	-	4.1	4.1	4.1	4.1	4.0	4.0	CS
			Copper (%)	-	-	1.61	1.61	1.61	1.61	2.00	2.00	
			Cobalt (%)	-	-	0.79	0.79	0.79	0.79	0.98	0.98	
Tilwezembe	56.4%		(Mt)	-	-	9.5	9.5	9.5	9.5	13.8	13.8	CS
			Copper (%)	-	-	1.89	1.89	1.89	1.89	1.75	1.75	
			Cobalt (%)	_	_	0.60	0.60	0.60	0.60	0.60	0.60	
Total Katanga	I		(Mt)	16	14	259	194	276	207	168	200	
			Copper (%)	3.58	3.53	3.66	3.53	3.66	3.53	3.78	3.81	
			Cobalt (%)	0.57	0.49	0.54	0.52	0.55	0.52	0.44	0.43	
Mutanda												
Mutanda	69.0%	OC	(Mt)	197.3	193.7	164.4	168.1	361.7	361.8	175.8	162.4	CS
South			Copper (%)	1.71	1.78	1.20	1.16	1.48	1.49	0.92	0.90	
			Cobalt (%)	0.66	0.70	0.45	0.43	0.56	0.57	0.29	0.29	
Mutanda	69.0%	OC	Mt	48.1	48.1	29.1	29.1	77.2	77.2	58.9	58.9	CS
North			Copper (%)	0.73	0.73	0.51	0.51	0.65	0.65	0.48	0.48	
			Cobalt (%)	0.34	0.34	0.14	0.14	0.26	0.26	0.08	0.08	
Total Mutanda	a		(Mt)	245	242	194	197	439	439	235	221	
			Copper (%)	1.52	1.57	1.10	1.06	1.33	1.34	0.81	0.79	
			Cobalt (%)	0.60	0.63	0.40	0.39	0.51	0.52	0.24	0.23	
Mopani												
Nkana	73.1%	UG	(Mt)	149.8	151.7	46.6	47.0	196.4	198.7	50.2	51.1	CS
Sulphides			Copper (%)	2.09	2.10	2.02	2.00	2.07	2.08	1.97	1.97	
			Cobalt (%)	0.11	0.11	0.12	0.12	0.11	0.11	0.15	0.14	
Nkana	73.1%	UG/OC	(Mt)	7.3	7.5	1.9	1.9	9.2	9.4	1.5	1.6	CS
Oxides			Copper (%)	2.28	2.32	1.93	1.94	2.21	2.24	1.91	1.92	
			Cobalt (%)	0.03	0.03	0.05	0.05	0.03	0.03	0.04	0.04	
Nkana	73.1%		(Mt)	-	-	5.7	5.7	5.7	5.7	0.8	0.8	CS
Tailings			Copper (%)	-	-	0.71	0.71	0.71	0.71	0.94	0.94	
Dump			Cobalt (%)	-	-	0.07	0.07	0.07	0.07	0.07	0.07	
Mufulira	73.1%	UG	(Mt)	33.5	28.0	15.9	12.5	49.4	40.5	21.9	24.0	CS
Sulphides			Copper (%)	2.45	2.46	2.57	2.75	2.49	2.55	2.42	2.46	
Mufulira	73.1%	UG	(Mt)	8.8	8.8	2.0	2.0	10.8	10.7	1.3	1.3	CS
Oxides			Copper (%)	1.19	1.19	0.90	0.90	1.14	1.14	0.81	0.82	
Mufulira	73.1%		(Mt)	3.0	3.0	1.8	1.8	4.8	4.9	1.3	1.3	CS
Surface			Copper (%)	1.81	1.81	1.80	1.80	1.81	1.81	1.76	1.76	
Total Mopani			(Mt)	202	199	74	71	277	270	77	80	
			Copper (%)	2.11	2.11	2.00	1.99	2.08	2.08	2.06	2.08	
			•••••••••	2.11	2.11	2.00	1.55	2.00	2.00		2.00	

African Copper Ore Reserves (Katanga, Mutanda, Mopani)

Name of	Attributable	Mining		Proved Ore	Reserves	Probable Or	e Reserves	Total Ore F	Reserves C	ompetent
operation	interest	method	Commodity		31.12.15		31.12.15		31.12.15	person
Katanga										
Kamoto	56.4%	UG	Ore (Mt)	8.2	8.2	17.2	17.2	25.5	25.5	JL
			Copper (%)	3.68	3.68	3.57	3.57	3.60	3.60	
			Cobalt (%)	0.37	0.37	0.52	0.52	0.47	0.47	
T17 (OC & UG)	56.4%	UG/OC	Ore (Mt)	2.2	2.2	9.1	9.1	11.3	11.3	JL
			Copper (%)	3.42	3.42	3.71	3.71	3.65	3.65	
			Cobalt (%)	0.54	0.54	0.64	0.64	0.62	0.62	
Mashamba East	56.4%	OC	Ore (Mt)	-	-	32.1	5.8	32.1	5.8	JL
			Copper (%)	-	-	2.13	2.68	2.13	2.68	
			Cobalt (%)	-	-	0.60	0.37	0.60	0.37	
KOV OC,	56.4%	OC	Ore (Mt)	-	-	55.9	55.9	55.9	55.9	JL
KOV UG, KTE			Copper (%)	-	-	4.23	4.23	4.23	4.23	
			Cobalt (%)	-	-	0.47	0.47	0.47	0.47	
Total Katanga			(Mt)	10	10	114	88	125	99	
			Copper (%)	3.63	3.62	3.50	3.94	3.51	3.91	
			Cobalt (%)	0.41	0.41	0.53	0.49	0.52	0.48	
Mutanda										
Mutanda South	69.0%	OC	Ore (Mt)	91.1	93.1	42.7	42.9	133.8	136.0	JL
			Copper (%)	2.04	2.24	1.53	1.77	1.88	2.09	
			Cobalt (%)	0.78	0.84	0.65	0.69	0.74	0.79	
Stockpiles	69.0%		Ore (Mt)	25.1	20.8	-	-	25.1	20.8	JL
			Copper (%)	1.31	1.68	-	-	1.31	1.68	
			Cobalt (%)	0.61	0.92	-	-	0.61	0.92	
Total Mutanda			(Mt)	116	114	43	43	159	157	
			Copper (%)	1.88	2.14	1.53	1.77	1.79	2.04	
			Cobalt (%)	0.74	0.85	0.65	0.69	0.72	0.81	
Mopani										
Nkana Sulphides	73.1%	UG	Ore (Mt)	96.6	100.2	23.7	22.9	120.3	123.1	HT
			Copper (%)	1.85	1.86	1.85	1.78	1.85	1.84	
			Cobalt (%)	0.09	0.09	0.10	0.09	0.09	0.09	
Mufulira Sulphides	73.1%	UG	Ore (Mt)	15.3	14.1	6.3	7.0	21.6	21.1	HT
·			Copper (%)	2.21	2.19	2.31	2.36	2.24	2.25	
Total Mopani			(Mt)	112	114	30	30	142	144	
			Copper (%)	1.90	1.90	1.95	1.92	1.91	1.90	
			Cobalt (%)	0.08	0.08	0.08	0.07	0.08	0.08	

Notes

Katanga: Remaining life of mine: expected to be in excess of 20 years. Expiry date of relevant permits: 7 May 2022 for the Kananga Extension and 3 April 2024 for all remaining permits (KTO and Mashamba East Open Pit, T-17 Open Pit, KOV Open Pit, Tilwezembe Open Pit, Kananga Mine), renewable in accordance with the DRC mining code for a period of 15 years.

Glencore owns 75.2% of Katanga Mining Limited ("KML"), which in turn owns 75% of Kamoto Copper Company SARL ("KCC"). KCC owns the material assets, including the mining and exploration rights related to the mining assets. La Generale des Carrieres et des Mines and La Société Immobilière du Congo, which are state-owned mining companies in the DRC, own the other 25% of KCC.

With the exception of Tilwezembe, primary mineralisation, in the form of sulphides, within the Lower Roan is associated with the Stratified Dolomite and Silicified Rocks for the Orebody Inferior and the Basal Schists and Upper Dolomitic Shales for the Orebody Superior and is thought to be sys-sedimentary in origin. Typical primary copper sulphide minerals are bornite, chalcopyrite, chalcocite and occasional native copper while cobalt is in the form of carrolite. The mineralisation occurs as disseminations or in association with hydrothermal carbonate alteration and silicification.

The mineralisation at Tilwezembe Mine is atypical being hosted by the Mwashya or R4 Formation. The mineralisation generally occurs as infilling of fissures and open fractures associated with the brecciation. The typical copper minerals are mainly chalcopyrite, malachite and pseudomalachite while cobalt is in the form of heterogenite, carrolite and spherocobaltite. Manganese minerals are psilomelane and manganite.

The increase in Mineral Resources from 2015 to 2016 is due to new drilling data and updated geological models developed for Mashamba East, KOV and KTO. Ore Reserves for Mashamba East increased as a result of revised pit optimisation and redesign based on the updated Mineral Resource model.

Mutanda: Remaining mine life: estimated in excess of 15 years. Expiry date of relevant mining permits: 26 May 2022 for Mutanda South ("Mutanda") and 1 July 2022 for Mutanda North ("Kansuki"). Both mining permits are renewable in accordance with the DRC mining code for periods of 15 years.

Kansuki consists of Kabwimia, Area 2 East, and Area 2 West. Mutanda consists of East (includes Area 1), Central (includes Central North West), and West.

Mopani: Remaining life of mine: 28 years for Nkana and 12 years for Mufulira. Expiry date of relevant mining/ concession licences: 31 March 2025 for both of these mines.

Within the Nkana mining area there are four underground mines and a series of open pits. The open pits are under care and maintenance and have therefore been excluded from Ore Reserves and included under Mineral Resources. All are situated on the north-eastern limb of the Nkana Syncline area. Other cupriferous zones are present in the nose and southwest limb of the syncline. The orebodies are stratiform and are mainly confined to a recognisable ore formation, which occurs near the base of the Katangan sequence within the Lower Roan Group of the Mine Series. In the underground workings, the principal copper ore minerals are chalcopyrite and bornite with subordinate chalcocite. There is a zoning in the geographical distribution of these minerals. Cobalt occurs as carrollite and cobaltiferious pyrite. The principal ore minerals are malachite, pseudomalchite, chrysocolla, native copper, cuprite and libethenite. In the open pit, malachite and chrysocolla are the principal ore minerals in the zone of oxidation closer to the surface. In some places however, vermiculite, malachite pseudomalachite and accessory wad are more important. At deeper levels chalcopyrite, bornite and chalcocite are predominantly present.

In the Mufulira mining area, the Basement Complex topography appears to have exerted a significant structural control during deformation. The distribution of ore minerals in all three orebodies is stratigraphically controlled, occurring dominantly as disseminations, blebs and irregular masses. The principal copper minerals are chalcopyrite (60%), bornite (40%), and minor/trace chalcocite. Oxide minerals are confined to near surface occurrences, and supergene enrichment zones. Generally the deposit is structurally simple being characterised by three main folds that are in part overturned with a plunge and dip approximately 10° to the northeast. The basin is open and untested at depth.

- CS = Christiano Santos Goncalves of Golder Associates Africa (Pty) Ltd, (MAusIMM CP (Geo)).
- HT = Hugo Tukker of Golder Associates Africa (Pty) Ltd, (ECSA, PrEng).
- JL = Jacobus Lotheringen of Ukwazi Mining Solutions (Pty) Ltd, for Golder Associates Africa (Pty) Ltd, (SAIMM, PrEng.).

Collahuasi Mineral Resources

Name of	Attributab	le Minina		Measured Resou		Indicated Reso	d Mineral urces	Measur Indicated F		Infe Mineral R		Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	person
Collahuasi	44%	OC	Sulphide (Mt)	824	757	4,218	4,261	5,042	5,019	4,801	4,851	NP
			Copper (%)	0.81	0.82	0.80	0.82	0.80	0.82	0.76	0.80	
			Molybdenum (%)	0.019	0.020	0.022	0.023	0.022	0.023	0.012	0.017	
			Oxide & Mixed (Mt)	35	34	45	49	80	83	52	25	NP
			Copper (%)	0.67	0.66	0.67	0.68	0.67	0.67	0.53	0.54	
Total Collah	nuasi		(Mt)	859	791	4,263	4,310	5,122	5,102	4,853	4,876	
			Copper (%)	0.80	0.81	0.79	0.82	0.80	0.82	0.76	0.80	
			Molybdenum (%)	0.02	0.02	0.02	0.02	0.02	0.02	0.01	0.02	

Collahuasi Ore Reserves

Name of	Attributab	le Minina		Proved Ore	Reserves	Probable Ore	e Reserves	Total Ore F	Reserves	Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	
Collahuasi	44%	OC	Sulphide (Mt)	419	501	2,669	2,592	3,088	3,093	AP
			Copper (%)	1.08	1.00	0.87	0.82	0.90	0.85	
			Molybdenum (%)	0.025	0.023	0.023	0.020	0.023	0.020	
			Oxide & Mixed (Mt)	-	15	-	15	-	30	AP
			Copper (%)	-	0.63	-	0.73	-	0.68	
Total Collahu	asi		(Mt)	419	516	2,669	2,607	3,088	3,123	
			Copper (%)	1.08	0.99	0.87	0.82	0.90	0.85	
			Molybdenum (%)	0.03	0.02	0.02	0.02	0.02	0.02	

Notes

Collahuasi: The Collahuasi district is located in northern Chile, Tarapacá Region, and hosts large Copper-Molybdenum porphyry-type deposits: Ujina, Rosario, and peripheral vein deposits such as Rosario Oeste and Rosario Sur. Sulphides as chalcopyrite, bornite and less chalcosite are the main copper minerals at Ujina and Rosario. At Rosario Oeste, supergene chalcosite is the main copper mineral. Rosario Sur is a small oxide-bearing deposit, mainly chrysocolla. Collahuasi is mined by open cut methods. The Rosario orebody is the main source of mineral for current Collahuasi operation. Mineral Resources for Rosario, Uiina. Rosario Oeste and Rosario Sur have been classified into Measured, Indicated and Inferred categories based on minimum search radius constraints and drill hole spacing. Mineral Resources are reported within the constraints of optimised pit shells. Mineral Resources and Ore Reserves have been updated on the basis of new drilling information, geological-geostatistical models, updated minina parameters and adjustments to metal prices forecast. The 49.6Mt ore production for 2016 is already depleted from the reported Ore Reserves. Ore Reserves estimates are based only on Measured and Indicated Mineral Resources, and optimised mine plans. Following the suspension of leaching operations, Oxide Ore Reserves have been reallocated to Mineral Resources. The Ore Reserves include all Proved and Probable Ore Reserves above the mill (operational) cut-off grade. The process of generating the resource model and optimised pit shells incorporated diluting materials as such no additional dilution factors were required in converting Mineral Resources to Ore Reserves. Both Mineral Resources and Ore Reserves include estimates of stockpile material at time of reporting. Material included in stockpile is periodically verified and reevaluated through drilling, trenching, and sampling. Stockpile material from Rosario and Ujina pits are classified according to the level of perceived uncertainty. These materials have been downgraded to Probable Ore Reserves. All sulphide stockpiles are considered Indicated

Mineral Resources and are classified as Probable Ore Reserves.

Collahuasi has a life of mine of 69 years from 2017 to 2085, according to the most recent Life of Mine plan that supports the present Ore Reserves report.

- NP = Nicolas Pizarro, Employee of Compañía Minera Doña Inés de Collahuasi (APEGBC).
- AP = Andrés Pérez, Employee of Compañía Minera Doña Inés de Collahuasi (APEGBC).

Antamina Mineral Resources

Name of	Attributable	e Minina		Measured Resou			d Mineral ources	Measu Indicated I	ed and Resources		I Mineral ources	Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	
Antamina	33.75%	OC	Sulphide Cu (Mt)	161	173	520	506	681	679	816	780	LC
			Copper (%)	0.89	0.89	0.86	0.88	0.87	0.88	0.82	0.83	
			Zinc (%)	0.14	0.14	0.15	0.14	0.15	0.14	0.14	0.14	
			Silver (g/t)	7	8	9	8	9	8	8	8	
			Molybdenum (%)	0.033	0.034	0.026	0.027	0.028	0.029	0.024	0.022	
			Sulphide Cu-Zn (Mt)	78	83	326	335	404	418	431	493	LC
			Copper (%)	0.95	1.00	0.92	0.94	0.93	0.95	0.98	1.02	
			Zinc (%)	1.93	1.88	1.82	1.84	1.84	1.85	1.52	1.53	
			Silver (g/t)	17	16	15	15	15	15	15	16	
			Molybdenum (%)	0.009	0.011	0.008	0.009	0.008	0.009	0.008	0.008	
Total Antai	mina		(Mt)	239	256	846	841	1,085	1,097	1,247	1,273	
			Copper (%)	0.91	0.93	0.88	0.90	0.89	0.91	0.88	0.90	
			Zinc (%)	0.72	0.70	0.79	0.82	0.78	0.79	0.62	0.68	
			Silver (g/t)	10	11	11	11	11	11	10	11	
			Molybdenum (%)	0.025	0.027	0.019	0.020	0.021	0.021	0.018	0.017	

Antamina Ore Reserves

Name of	Attributable	Mining		Proved Ore	Reserves	Probable Ore	Reserves	Total Ore F	Reserves	Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	person
Antamina	33.75%	OC	Sulphide Cu (Mt)	115	128	190	207	305	335	LM
			Copper (%)	1.03	1.02	1.02	0.99	1.02	1.00	
			Zinc (%)	0.15	0.14	0.19	0.17	0.17	0.16	
			Silver (g/t)	8	8	8	8	8	8	
			Molybdenum (%)	0.038	0.037	0.032	0.032	0.034	0.034	
			Sulphide Cu-Zn (Mt)	59	63	188	200	247	263	LM
			Copper (%)	0.98	1.08	0.82	0.83	0.86	0.89	
			Zinc (%)	2.13	2.10	2.02	1.98	2.05	2.01	
			Silver (g/t)	17	17	13	13	14	14	
			Molybdenum (%)	0.008	0.009	0.008	0.008	0.008	0.008	
Total Antam	ina		(Mt)	174	191	378	407	552	598	
			Copper (%)	1.01	1.04	0.92	0.91	0.95	0.95	
			Zinc (%)	0.82	0.79	1.10	1.06	1.01	0.97	
			Silver (g/t)	11	11	11	11	11	11	
			Molybdenum (%)	0.028	0.028	0.020	0.020	0.022	0.023	

Notes

Antamina: Antamina is a polymetallic (copper, zinc and molybdenum predominate) skarn deposit resulting from complex multiple intrusive events. Copper mineralization occurs mainly as chalcopyrite except for some areas of bornite, representing approximately 5% of the deposit. Zinc mineralization generally occurs as sphalerite. Other significant sulphide minerals include molybdenite and pyrite, while trace amounts of numerous silver and bismuth bearing minerals and local areas of galena (lead sulphide) are also found within the deposit.

The Mineral Resource classification is based on geologic risk factors (proportion of breccia indicator), geologic continuity (intrusive, endoskarn and exoskarn-waste) and drill hole spacing. The Antamina deposit is sufficiently well drilled to support the classification criteria.

Ore Reserves results were developed during the mine planning process in 2016, which is based on the end of year 2016 topography projection. The life-of-mine plan, and subsequent Ore Reserve estimation used for this statement considers only Measured and Indicated Mineral Resources; all Inferred Mineral Resources within this pit has been treated as waste. The cut-off grade for the Ore Reserves estimate varies by year in an effort to maximize the net present value of the life-of-mine. Ore Reserves are limited to the current operation tailings dam capacity.

The total Sulphide Ore Reserves (Cu and Cu-Zn) decreased 46Mt, primarily due to depletion during 2016.

Antamina has an approved life of mine plan of 13 years (2017-2029), based on Ore Reserves. Operating permits are valid until the end of the life of mine.

- LC = Lucio Canchis, Employee of Compania Minera Antamina S.A. (AusIMM);
- LM = Luis Mamani, Employee of Compañia Minera Antamina S.A. (AusIMM).

Other Oouth America Mineral Resources (Alumbrera, Eomas Dayas, Antapaccay, Fundaqui)	Other South America Mineral Resources	(Alumbrera, Lomas Ba	ayas, Antapaccay, Punitaqui)
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Name of	Attributable	Minin	g	Measured Resou	rces	Reso	d Mineral urces	Measur Indicated F	Resources	Reso		Competent
operation	interest	metho	od Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	person
Alumbrera	50%											
Bajo de la Alum	brera	OC	Ore (Mt)	53	82	3	4	56	86	1	1	FM
			Copper (%)	0.31	0.31	0.24	0.25	0.31	0.31	0.22	0.23	
			Gold (g/t)	0.30	0.30	0.21	0.22	0.29	0.29	0.22	0.22	
	· · · · · · · · · · · · · · · · · · ·		Molybdenum (%)	0.011	0.012	0.013	0.014	0.011	0.012	0.014	0.014	
Bajo El Durazno)	OC	(Mt)	50	57	43	44	93	101	56	59	FM
			Copper (%)	0.15	0.15	0.15	0.15	0.15	0.15	0.14	0.14	
			Gold (g/t)	0.40	0.39	0.41	0.41	0.41	0.40	0.33	0.33	
Lomas Bayas	100%											
Lomas Bayas I		OC	Oxide & Mixed (Mt)	(Mt) 50 57 43 44 93 101 56 59 FM Copper (%) 0.15 0.15 0.15 0.15 0.15 0.15 0.14 0.12 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.20 0.21 0.21 0.21 0.21 0.23 0.23 0.22 0.26 0.10 0.10 0.10								
			Copper (%)	0.32	0.28	0.22	0.21	0.24	0.22	0.20	0.20	
			Soluble Copper (%)	0.16	0.17	0.11	0.19	0.12	0.13	0.10	0.10	
Lomas Bayas II		OC	Oxide & Mixed (Mt)	182	198	342	359	524	557	57	49	MR
			Copper (%)	0.30	0.30	0.23	0.23	0.25	0.26	0.10	0.10	
			Soluble Copper (%)	0.21	0.21	0.15	0.14	0.17	0.17	0.10	0.10	
Lomas Bayas III		00 8	Sulphide&Mixed (Mt)	18	17	422	431	440	448	449	443	MR
			Copper (%)	0.55	0.54	0.37	0.36	0.38	0.37	0.30	0.30	
			Oxide & Mixed (Mt)	1	1	50	58	51	59	32	26	
			Copper (%)	0.26	0.23	0.24	0.22	0.24	0.22	0.20	0.20	
Antapaccay	100%	OC	(Mt)	206	198	452	488	658	686	156	165	HB
			Copper (%)	0.57	0.60	0.43	0.46	0.47	0.50	0.40	0.40	
			Gold (q/t)	0.13	0.13	0.08	0.09	0.10	0.10	0.10	0.10	
			Silver (g/t)	1.45	1.6	1.17	1.3	1.26	1.4	0.80	0.9	
			Molybdenum (%)	0.005	0.005	0.005	0.005	0.005	0.005	0.005	0.005	
Coroccohuayc	o 100%	UG/C	DC (Mt)	33	9	256	247	289	256	56	80	HB
			Copper (%)	0.85	0.71	0.91	1.02	0.90	1.01	1.08	1.20	
			Gold (g/t)	0.09	0.08	0.10	0.10	0.10	0.10	0.10	0.10	
			Silver (g/t)	3.13	2.08	2.68	3.18	2.73	3.14	3.51	4.70	
Punitagui	100%	UG	(Mt)	0.92	1.25	4.22	4.19	5.14	5.44	1.87	1.86	CM
•			Copper (%)	1.69	1.70	1.04	1.00	1.16	1.19	1.12	1.12	
			Silver (g/t)	5.31	5.92	1.85	1.84	2.47	2.77	1.12	1.11	
Total Other So	uth Americ	a	(Mt)	701	673	2,195	2,104	2,896	2,777	886	870	
			Copper (%)	0.41	0.39	0.37	0.40	0.38	0.40	0.33	0.37	
			Gold (g/t)	0.09	0.11	0.04	0.04	0.05	0.06	0.05	0.05	
			Silver (g/t)	0.6	0.5	0.6	0.7	0.6	0.6	0.4	0.6	

Name of	Attributable	Mining		Proved Ore		Probable Ore		Total Ore	Reserves C	
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	persor
Alumbrera	50%									
Bajo de la Alumb	orera	OC	Ore (Mt)	16	25	-	1	16	26	FM
			Copper (%)	0.34	0.39	-	0.27	0.34	0.39	
			Gold (g/t)	0.33	0.38	-	0.29	0.33	0.38	
			Molybdenum (%)	0.009	0.012		0.012	0.009	0.012	
Bajo el Durazno		OC	Ore (Mt)	24	3	4	-	28	3	FM
			Copper (%)	0.17	0.18	0.16	-	0.17	0.18	
			Gold (g/t)	0.43	0.36	0.37	-	0.42	0.36	
Lomas Bayas	100%									
Lomas Bayas I		OC	Oxide & Mixed (Mt)	88	82	145	151	233	233	MR
			Copper (%)	0.28	0.28	0.22	0.21	0.24	0.23	
			Soluble Copper (%)	0.16	0.17	0.13	0.13	0.14	0.14	
Lomas Bayas II		OC	Oxide & Mixed (Mt)	165	185	208	235	373	420	MR
			Copper (%)	0.30	0.30	0.24	0.23	0.27	0.26	
			Soluble Copper (%)	0.21	0.21	0.15	0.15	0.18	0.18	
Antapaccay	100%	OC	(Mt)	207	194	324	353	531	547	HB
			Copper (%)	0.56	0.60	0.43	0.48	0.48	0.52	
			Gold (g/t)	0.12	0.13	0.08	0.10	0.10	0.11	
			Silver (g/t)	1.44	1.6	1.19	1.4	1.29	1.4	
Coroccohuayco) 100%									
		OC	(Mt)	22	_	41	_	63	_	HB
			Copper (%)	0.75	_	0.56	_	0.63	_	
			Gold (g/t)	0.08	_	0.06	_	0.07	_	
			Silver (g/t)	2.75	_	1.57	_	1.98	_	
		UG	(Mt)	9	_	112	_	121	_	AZ
			Copper (%)	0.82	_	1.25	_	1.22	_	
			Gold (g/t)	0.09	_	0.12	_	0.12	_	
			Silver (g/t)	3.02	_	3.93	_	3.86	_	
Punitaqui	100%	UG	Ore (Mt)	0.32	1.08	0.08	0.22	0.40	1.30	CM
			Copper (%)	1.63	1.47	1.69	1.30	1.64	1.44	511
			Silver (g/t)	4.40	4.94	4.71	5.53	4.46	5.04	
Total Other Sou	th America		(Mt)	531	490	834	740	1,365	1,230	
			Copper (%)	0.42	0.42	0.46	0.35	0.45	0.37	
			Gold (g/t)	0.08	0.42	0.40	0.05	0.43	0.06	
			Silver (g/t)	0.00	0.6	1.1	0.03	0.9	0.00	

Other South America Ore Reserves (Alumbrera, Lomas Bayas, Antapaccay, Punitaqui)

Notes

Bajo de La Alumbrera and Bajo El Durazno:

The **Bajo de la Alumbrera** copper-gold porphyry deposit is located in the Hualfin district, Belen department of the Catamarca province, in the northwest region of Argentina. The orebody consists of primary sulphide mineralised ore which comprises disseminated, vein and fracture controlled chalcopyrite in altered dacite and andesite host rocks, with minor chalcocite and covellite in the enriched zone that surrounds the major faults.

The variations from the 31 December 2015 published statement are a result of the latest review of the geotechnical model, the final design of the pit slopes, updated new economic parameters, and depletion for milled tonnage.

The Ore Reserves and Mineral Resources figures are obtained using Ordinary Kriging interpolation within geological constraints from an assay database comprising some 116,000m of diamond drilling and 15,000m of reverse circulation drilling. Ore Reserves are based on a pit optimisation, and are reported using an economic 0.22% Cu equivalent cut-off grade. The strip ratio for the updated pit is

0.2 and defined as the in-pit tonnes of waste divided by the in-pit tonnes of ore. The life-of-mine is up to 2018.

The concession and permits of the operation/exploration are valid to the end of the life of the mine.

The Mineral Resources (Measured and Indicated) are 56Mt (down 30 Mt) due to changes in phase design, metal prices and operational cost

The **Bajo el Durazno** deposit belongs to the mining district of Agua de Dionisio and is located 5km northeast of the Bajo de la Alumbrera deposit and 2km from Minera Alumbrera's concentrator plant. The deposit is located within the Farallón Negro volcanic complex, which is eroded, deformed by faults and mineralized, resulting in a "depressed geomorphological" with hydrothermal alteration known as "Bajo".

39,444m of diamond drilling was completed between 2009 and 2015. Ore Reserves and Mineral Resources figures are obtained using Ordinary Kriging interpolation for Cu and Au. The size of the block in the geological model (10x10x15) was based on geological criteria stemming from the geological interpretation and the proposed drilling grid.

The Mineral Resources were estimated using a cut-off 0.20g/t for Au. Ore Reserves are based on a pit optimization and are reported on an economic 0.30g/t Au equivalent cut-off grade.

The increase in reserves compared to the 2015 statement is the result of optimisation, which added two additional phases to the mine. The life of mine is up to 2018.

Lomas Bayas (I) – Lomas Bayas (III): Lomas Bayas is a low grade Cu-Mo deposit resulting from the intrusion of several porphyry and breccia systems that were later exposed to leaching and subsequent supergene enrichment and *in situ* oxidation. Green copper oxides, copper sulphates in various forms and less partially mixed ores are the main source of ore for the existing SX/EW operation. The copper oxidessulphate mineralisation is the basis of the Lomas Bayas Ore Reserves (Lomas Bayas I).

Mineral Resources and Ore Reserves estimation has been completed using Ordinary Kriging on a block model that includes historic drill holes totalling 318,788m and sampled at 2m intervals. Ore Reserves are as of 31 December 2016 using projected topography; operational pit life-of-mine developed in 2016; cut-off grade Heap: 0.27% CuT and ROM: 0.07% CuT. Lomas Bayas (I) Ore Reserves are maintained. The Lomas Bayas (II) Mineral Resources estimates sulphides in a pit shell calculated using Measured, Indicated and Inferred Mineral Resources; Oxides-Mixed within this pit are also considered Mineral Resources that will eventually feed the SX/EW operation.

Drilling continues in the Lomas Bayas (I) and Lomas Bayas (III) ore bodies, whereby the Lomas Bayas (III) sulphide layer lies beneath the Lomas Bayas (I) oxide layer. Additional oxide-mixed material at depth has been added to the Lomas Bayas (I) resource base during the year.

The SX/EW operation plant has a current life of mine that extends to 2028; permits for the operation are valid to the end of the life of the mine.

Lomas Bayas (II): This low grade Cu deposit is located 2km south of Lomas I pit, in the same district and geological environment as Lomas Bayas deposit. The main difference is a larger presence of water soluble copper oxides and lower geotechnical rock quality. The Lomas II block model includes historic drill holes totalling 127,369m and sampled at 2m intervals. Mineral Resources and Ore Reserves estimation has been completed using Ordinary Kriging. Ore Reserves are as of 31 December 2016, considering: operational pit life-of-mine developed in 2016; cut-off grade Heap: 0.27% CuT and ROM: 0.07% CuT. Mineral Resources as of 31 December 2016 considers economic pit shell at similar cut-off grade.

The SX/EW operation plant has a current life of mine that extends to 2028; permits for the operation are valid to the end of the life of the mine.

Antapaccay: The Antapaccay orebody is located 10km southwest of Tintaya, and together with Coroccohuayco is part of the Tintaya mineralised district. It is a sulphide mineralised system comprising disseminated, vein and fracture controlled chalcopyrite and bornite in altered quartz-monzonite and diorite in a limestone host rock, with some mineralised exoskarn areas and minor copper oxides and copper carbonates in the upper part of the deposit.

Antapaccay Ore Reserves and Mineral Resources as at 31 December 2016. Mineral Resource categorisation is based on assessment of orebody and grade continuity, structural complexity, data quality, adequacy of data coverage, and reasonable prospects of economical extraction. The Mineral Resources estimation is based on a block model with grade interpolation by using Ordinary Kriging. Mineral Resources are stated at defined internal copper cut–off, which is maintained at 0.15% TCu. Ore Reserves are then derived from Measured and Indicated Mineral Resources after applying economical and technical modifying factors. Ore Reserves decreased by 16Mt during 2016.

The deposit geology model and Mineral Resources estimate has been updated in 2015 using a drill hole database that now includes over than 248,060m of total drilling data. The major variations from the previous published Mineral Resources statement are due to the inclusion of new infill drilling to the updated geological model and block grade interpolate.

The Ore Reserves are for 13 years ending in 2029, with ore processed through the Tintaya and Antapaccay plants. Operation permits are valid until the end of the life of mine.

- Coroccohuayco: The Coroccohuayco copper-gold skarn deposit is located 9km southeast of Tintaya. Together with Tintaya and Antapaccay this orebody is part of the Tintaya mineralised district. The main copper bearing minerals are bornite, chalcopyrite and chalcocite. The Coroccohuayco orebody is defined as a copper skarn deposit, which consists of Cretaceous sedimentary rocks of the Ferrobamba and Mara formation intruded by monzonitic plutons of the Eocene-Oligocene Andahuaylas-Yauri batholiths. The deposit geology model and Mineral Resources and Reserves estimate has been updated at 31 December 2016. The major variations from the previous published Mineral Resources statement are due to the inclusion of new infill drilling to the updated geological model and block grade interpolate, and the classification of Ore Reserves.
- Punitaqui: Remaining life of mine is approximately 1 year based on Ore Reserves and 10 years based on Mineral Resources. Several epigenetic stratifiorm copper mineralisation (manto type) bodies with variable thicknesses between 20 to 40m are distributed along 900m strike length mineralised corridor named Cinabrio zone. Mineralisation is composed of crisocole, brochantita and malachite in upper oxide levels turning into a mixed zone composed of malachite, crisocole and chalcopyrite. Main sulphide zones are composed of pyrite, bornite and chalcopyrite. All mineralisation is distributed in calcareous shales also within minor pre-existing faults. Dalmacia deposit is located 7km south of the Punitaqui concentrator plant. Also described as a stratiform copper deposit the mineralisation occurs as irregular lenses of chalcopyrite, chalcocite and covellite sulphides within the porphyritic andestites of the reloj formation.

Competent Persons

AZ = Americo Zuzunaga, Employee of BISA (AusIMM).

CM = Carlos Manchego, independent consultant (AusIMM).

- FM = Flavio Montini, Employee of Glencore (AusIMM).
- HB = Heller Bernabé, Employee of Glencore (AusIMM).
- MR = Mauricio Rubio, Employee of Glencore (AusIMM).

Australia Mineral Resources (Ernest Henry, Mount Isa, Cobar)

Name of	Attributab	le Minina		Measured Resou		Indicated Reso		Measur Indicated F		Inferred Reso		ompetent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	person
Ernest Henry												
Underground	70%	UG	(Mt)	12.1	16.1	68.7	71.0	80.7	87.1	9.0	9.0	CS
			Copper (%)	1.33	1.29	1.15	1.15	1.18	1.17	1.1	1.1	
			Gold (g/t)	0.70	0.67	0.59	0.59	0.61	0.60	0.5	0.5	
E1	100%	OC	(Mt)	4.6	4.6	5.5	5.5	10.1	10.1	0.4	0.4	CS
			Copper (%)	0.70	0.70	0.75	0.75	0.73	0.73	0.9	0.9	
			Gold (g/t)	0.20	0.20	0.23	0.23	0.22	0.22	0.3	0.3	
Monakoff	100%	OC	(Mt)	-	_	2.4	2.4	2.4	2.4	0.1	0.1	CS
			Copper (%)	-	-	0.95	0.95	0.95	0.95	0.8	0.8	
Mountlea			Gold (g/t)	-	-	0.30	0.30	0.30	0.30	0.2	0.2	
Mount Isa	100%											
X41 Mine 500, 6	50,	UG	(Mt)	25.2	24.9	15.0	15.4	40.2	40.3	6.2	6.5	PI
1100 & 1900 Ore	bodies		Copper (%)	1.90	1.93	1.80	1.83	1.86	1.90	1.7	1.7	
Enterprise Mine 3	3000	UG	(Mt)	28.5	21.9	6.2	3.4	34.7	25.2	1.7	0.5	HC
& 3500 Orebodie	s		Copper (%)	2.59	2.90	2.37	2.75	2.55	2.88	2.2	2.7	
Open Pit		OC	(Mt)	48	48	82	82	130	130	138	138	CD
			Copper (%)	1.46	1.46	1.32	1.32	1.37	1.37	0.89	0.89	
Cobar	100%	UG	(Mt)	3.1	3.0	2.6	2.5	5.7	5.6	6.3	6.4	JH
			Copper (%)	5.95	5.96	5.34	5.42	5.67	5.71	5.5	5.6	
			Silver (g/t)	23.4	23.0	25.0	25.0	24.1	24.2	21	21	
Total Australia			(Mt)	122	119	182	182	304	301	162	161	
			Copper (%)	1.89	1.89	1.37	1.36	1.58	1.57	1.1	1.1	
			Gold (g/t)	0.08	0.10	0.23	0.24	0.17	0.18	0.03	0.03	
			Silver (g/t)	0.6	0.6	0.4	0.3	0.5	0.5	0.8	0.8	

Australia Ore Reserves (Ernest Henry, Mount Isa, Cobar)

Name of	Attributab	le Mining		Proved Ore	e Reserves	Probable Or	e Reserves	Total Ore	Reserves	Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	person
Ernest Henry	70%									
Underground		UG	Ore (Mt)	7.1	10.9	52.3	47.0	59.4	57.9	AC
			Copper (%)	1.41	1.17	0.96	1.03	1.02	1.06	
			Gold (g/t)	0.72	0.60	0.48	0.53	0.50	0.54	
Mount Isa	100%									
X41 Mine 500, 650,		UG	Ore (Mt)	6.7	8.0	12.1	7.5	18.8	15.5	GC
1100 & 1900 Orebo	dies		Copper (%)	1.84	1.91	1.79	1.78	1.80	1.85	
Enterprise Mine 300	00	UG	Ore (Mt)	8.9	12.3	12.9	1.8	21.8	14.1	GC
& 3500 Orebodies			Copper (%)	2.47	2.64	2.22	2.51	2.32	2.62	
Cobar	100%	UG	Ore (Mt)	3.2	3.2	2.5	2.6	5.7	5.8	LW
			Copper (%)	4.38	4.43	3.78	4.04	4.12	4.25	
			Silver (g/t)	17.6	17.7	17.4	17.7	17.5	17.7	
Total Australia			(Mt)	26	34	80	59	106	93	
			Copper (%)	2.25	2.17	1.38	1.30	1.59	1.63	
			Gold (g/t)	0.20	0.19	0.31	0.42	0.28	0.34	
			Silver (g/t)	2.2	1.6	0.5	0.8	0.9	1.1	

Notes

Ernest Henry Underground: In October 2016, Glencore entered into an agreement with Evolution Mining Limited ("Evolution"), whereby Glencore received cash in return for a 30% economic interest in the Ernest Henry Mine underground mining operation ("EHM"). Evolution's 30% economic interest is subject to an agreed life of mine and block model. The E1 and Monakoff resources are not a part of the agreement.

The current expected mine life is 10 years (completion in 2026) with the Mining Lease tenements due to expire in 10 years (2026).

Copper and gold mineralisation occurs in a breccia comprised of strongly altered and replaced intermediate volcanic fragments in a matrix assemblage of predominantly magnetite, chalcopyrite and carbonate. Copper occurs as chalcopyrite and gold is strongly associated with chalcopyrite.

Changes to Mineral Resources are due exclusively to depletion through mining.

Net of depletion for mining over the year, and changes in draw strategy and mine design, the Ore Reserve has decreased by 1.5Mt since December 2015.

E1: Mining of the reserve was completed in early 2014. The expiry date of relevant tenements is 31 December 2032.

Economic mineralisation at E1 occurs as breccia-hosted mineralisation within the footwall volcanics at E1 North, and as strata-bound, replacement style mineralisation within the mineralised sedimentary units at E1 North, Central, South and East.

Monakoff: As per 2015, Monakoff and Monakoff East remain excluded from the Ore Reserves. The expiry date of relevant tenements is 31 October 2032.

Economic mineralisation at Monakoff and Monakoff East occurs in very-fine to medium grained steeply dipping metasediment units.

Mount Isa X41 Copper Mine: Mineralisation occurs generally as breccia hosted massive to disseminated chalcopyrite in "silica dolomite" altered pyritic dolomitic siltstone. Mining depletion, sterilisation and changes to mine design amounted to a net 3.3Mt increase in Ore Reserves. The increase results from the inclusion of additional material based on cut-off grade studies.

The X41 Mine Mineral Resources increased due to a revision in cut-off grade.

Mount Isa Enterprise Copper Mine: Mineralisation occurs generally as breccia hosted massive to disseminated chalcopyrite in "silica dolomite" altered pyritic dolomitic siltstone. Mining depletion, sterilisation and mine design changes amounted to a net 7.7Mt increase in Ore Reserves. The increase results from the inclusion of additional material based on cut-off grade studies.

The Enterprise Mine Mineral Resources increased due to a revision in cut-off grade.

The underground life of mine estimate for the X41 and Enterprise Mines is 7 years (2023) with the tenements due to expire in 20 years (2036).

Mount Isa Open pit: The expiry date of relevant concession licences occurs in 2036.

Copper mineralisation occurs generally as breccia hosted massive to disseminated copper minerals in "silica dolomite" altered pyritic dolomitic siltstone. Approximately half of the Mineral Resources are defined by primary chalcopyrite, the remainder being oxidised or partially oxidised, with a minor amount of supergene chalcocite mineralisation. The 2016 Mineral Resources remain unchanged from 2015 and are reported inside an optimised pit shell using a cut-off grade of 0.5% Cu.

Cobar: The expected remaining life of mine is approximately 5 years based on Ore Reserves and approximately 10 years based on Mineral Resources, although Cobar has previously been able to extend its expected life of mine through exploratory drilling over the past 50 years. The expiry date of relevant mining/concession licences is 24 June 2028.

Economic mineralisation at Cobar occurs mostly as narrow lenses with short strike lengths that are depth extensive. Lenses consist of vein or semi massive to massive chalcopyrite hosted by sub-vertical quartz-chlorite shear zones within a siltstone unit. The Cobar Mineral Resource is reported within five 'systems': Western, Eastern, QTS North, QTS South and QTS Central.

Mining depletion since December 2015 correlates with exploration upside over the same period resulting in minimal variation to the reported Mineral Resources and Ore Reserves.

Competent Persons

AC = Alexander Campbell, Glencore Copper (AusIMM).

- CD = Chris De-Vitry, ARANZ GEO (AusIMM).
- CS = Colin Stelzer, Glencore Copper (AusIMM).
- GC = Gibson Chitumbura, Glencore Copper (AusIMM).
- HC = Helen Coackley, Glencore Copper (AusIMM).
- JH = Jason Hosken, Glencore Copper (AusIMM).
- LW = Leanne Waddell, Glencore Copper (AusIMM).
- PI = Patricia Ila'ava, Glencore Copper (AusIMM).

Other projects Mineral Resources

Name of	Attributab	ole Minina		Measured Resou			d Mineral urces	Measured ar Mineral R		Inferred Reso		Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	
El Pachón	100%	OC	(Mt)	534	534	1,056	1,054	1,590	1,588	1,567	1,509	FM
			Copper (%)	0.67	0.67	0.49	0.49	0.55	0.55	0.41	0.41	
			Silver (g/t)	2.4	2.4	2.0	2.1	2.2	2.2	1.8	1.8	
		Mc	lybdenum (%)	0.013	0.013	0.011	0.011	0.012	0.012	0.009	0.009	
West Wall	50%		(Mt)	-	-	495	495	495	495	970	970	MMV
Copper Project	Copper Project		Copper (%)	-	-	0.55	0.55	0.55	0.55	0.48	0.48	
			Gold (g/t)	-	-	0.05	0.05	0.05	0.05	0.05	0.05	
		Mo	lybdenum (%)	-	-	0.009	0.009	0.009	0.009	0.008	0.008	
Total Other p	rojects		(Mt)	534	534	1,551	1,549	2,085	2,083	2,537	2,479	
			Copper (%)	0.67	0.67	0.51	0.51	0.55	0.55	0.44	0.44	
			Gold (g/t)	_	_	0.02	0.02	0.01	0.01	0.02	0.02	
			Silver (g/t)	2.4	2.4	1.4	1.4	1.7	1.7	1.1	1.1	
		Мо	ybdenum (%)	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	

Notes

- El Pachón: Located in the southwest of San Juan Province of Argentina, the El Pachón orebody is a porphyry coppermolybdenum deposit in which two major stages of sulphide mineralisation contributed to the formation of the orebody. The bulk of the ore takes the form of disseminated Chalcopyrite-Molibdenite primary sulphides on top which an immature, relatively small copper enrichment blanket has been developed. In this upper part of the deposit Chalcocite and minor Covellite are partially replacing the primary Chalcopyirite ore. Mineral Resources stated herein are based on assay and geology information from 135,000m of mainly diamond drill holes. Mineral Resources have been classified using a combination of criteria including geological continuity and Kriging parameters. Mineral Resources are constrained by the use of an economic pit shell determined using Measured, Indicated and Inferred Mineral Resources and current assumption for the economic and technical modifying factors conditioning the resource pit. The increase in resources compared to the 2015 report is a result of changes in economic assumptions.
- West Wall: The West Wall Copper Project is located in central Chile's Valparaiso Region, 100km northeast of Santiago and 70km north of the Rio Blanco - Los Bronces mineral district. Glencore and Anglo American each have a 50% interest in the mining comwpany West Wall SCM which holds the project. Porphyry copper style hydrothermal alteration covers a large area of approximately 7km by 3km. Exploration activities have focused in the south of the prospect at Lagunillas and West Wall North, where drilling has outlined copper sulphide Mineral Resources associated with porphyry intrusive bodies. Mineral Resources have been classified using a combination of criteria including geological continuity and Kriging parameters. Mineral Resources are constrained by the use of an economic pit shell, determined using Indicated and Inferred Mineral Resources and current assumption for the economic and technical modifying factors conditioning the limits of the resource pit.

Competent Persons

FM = Flavio Montini, Employee of Glencore (AusIMM).

MMV = Manuel Machuca Valderrama, Employee of Anglo American for the West Wall Project (AusIMM).

Kazzinc Mineral Resources

Name of	Attributable	Minina		Measured Resou		Indicated Reso		Measur Indicated F			Mineral urces (Competent
operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16		person
Maleevsky	69.7%	UG	(Mt)	4.38	6.58	5.8	5.0	10	12	5	6.2	AC
			Zinc (%)	9.13	6.51	7.8	6.8	8.4	6.6	6	7	
			Lead (%)	1.36	1.04	1.3	1.2	1.3	1.1	1	2	
			Copper (%)	2.91	2.33	2.5	1.9	2.7	2.2	2	2	
			Silver (g/t)	77	79	68	71	72	75	46	75	
			Gold (g/t)	0.7	0.6	0.6	0.6	0.7	0.6	0.6	0.6	
Ridder-Sokolny	69.7%	UG	(Mt)	16.9	17.3	14	14	31	31	6	6	AC
			Zinc (%)	1.5	1.5	1.3	1.3	1.4	1.4	1	1	
			Lead (%)	0.66	0.7	0.5	0.5	0.6	0.6	0.5	0.5	
			Copper (%)	0.72	0.8	0.6	0.7	0.7	0.7	1	1	
			Silver (g/t)	24.0	24.3	27	27	25	26	22	22	
			Gold (g/t)	1.81	1.9	1.5	1.6	1.7	1.8	2	2	
Tishinsky	69.7%	UG	(Mt)	4.02	2.94	3.3	3.6	7.4	6.6	2.0	1.7	AC
			Zinc (%)	6.07	7.84	5.9	6.3	6.0	7.0	5	6	
			Lead (%)	1.11	1.76	1.0	1.2	1.0	1.5	0.5	0.5	
			Copper (%)	0.62	0.78	0.7	0.6	0.7	0.7	0.3	0.5	
			Silver (g/t)	13.6	11.6	12	10	13	11	5	7	
			Gold (g/t)	0.83	0.66	0.9	0.7	0.9	0.7	0.4	0.5	
Shubinsky	69.7%	UG	(Mt)	0.71	0.85	0.64	0.36	1.4	1.2	0.3	0.2	AC
			Zinc (%)	3.4	3.2	2.1	2.0	2.8	2.9	2	2	
			Lead (%)	0.6	0.5	0.4	0.3	0.5	0.4	0.3	0.4	
			Copper (%)	2.4	2.2	2.0	1.9	2.2	2.1	1	1	
			Silver (g/t)	25.7	24.8	22	21	24	24	13	17	
			Gold (g/t)	0.56	0.64	0.7	0.9	0.6	0.7	0.4	0.3	
Staroye Tailings	69.7%		(Mt)	-	_	2.4	2.4	2.4	2.4	1.4	1.4	AC
Dam			Silver (g/t)	-	_	11	11	11	11	10	10	
			Gold (g/t)	-	-	1.0	1.0	1.0	1.0	0.8	0.8	
Chashinskoye	69.7%	OC	(Mt)	_	_	58	58	58	58	30	30	AC
Tailings Dam			Silver (g/t)	-	_	5.2	5.2	5.2	5.2	5	5	
			Gold (g/t)	-	_	0.7	0.7	0.7	0.7	0.5	0.5	
Shaimerden	69.7%	OC	(Mt)	_	_	1.8	1.9	1.8	1.9	_	_	AC
Stockpiles			Zinc (%)	_	_	22	22	22	22	_	-	
Dolinnoe	69.7%	UG	(Mt)	_	_	2.4	2.4	2.4	2.4	4.4	4.4	AC
			Zinc (%)	_	_	2.4	2.4	2.4	2.4	1	1	
			Lead (%)	_	_	1.2	1.2	1.2	1.2	1	1	
			Copper (%)	_	_	0.3	0.3	0.3	0.3	0.3	0.3	
			Silver (g/t)	_	_	67	67	67	67	76	76	
			Gold (g/t)	_	_	6.3	6.3	6.3	6.3	6	6	
Obruchevskoe	69.7%	UG	(Mt)	_	_	4.1	4.1	4.1	4.1	2.9	2.9	AC
			Zinc (%)	_	_	10.5	10.5	10.5	10.5	3	3	
			Lead (%)	_	_	4.8	4.8	4.8	4.8	1	1	
			Copper (%)	_	_	1.0	1.0	1.0	1.0	0.6	0.6	
			Silver (g/t)	-	_	62	62	62	62	48	48	
			Gold (g/t)	_	_	1.8	1.8	1.8	1.8	0.7	0.7	
Zhairemsky	69.7%		(Mt)	23.3	_	0.6	23	24	23	1	1	AC
Zapadny			Zinc (%)	4.11	_	4.7	4.3	4.1	4.3	4	4	
			Lead (%)	2.43	_	0.9	2.6	2.4	2.6	3	3	
			Silver (g/t)	25.9	_	27	28	26	28	30	33	
Zhairemsky	69.7%	OC	(Mt)	36.5	_	3.5	34	40	34			AC
Dalnezapadny		-	Zinc (%)	4.39	_	3.9	4.9	4.3	4.9	_	_	
			Lead (%)	1.24	_	0.9	1.4	1.2	1.4	_	_	
			Silver (g/t)	5.9	_	2.7	6.7	5.5	6.7	_	_	

Kazzinc Mineral Resources (continued)

Nome of	ame of Attributable Mining peration interest method			Measured Resou		Indicated Reso	d Mineral	Measur		Inferred Reso		Competent
operation		method	Commodity		31.12.15		31.12.15		31.12.15		31.12.15	person
Zhairemsky	69.7%	OC	(Mt)	1.4	-	2.1	3.5	3.5	3.5	0.1	0.2	AC
Ushkatyn			Zinc (%)	0.1	-	0.1	0.1	0.1	0.1	0.2	0.2	
			Lead (%)	7.3	_	5.3	7.1	6.1	7.1	4	8	
			Silver (g/t)	49.3	-	37	43	41	43	17	20	
Uzhynzal	100%	OC	(Mt)	-	-	_	_	_	-	14	-	AC
			Zinc (%)	_	-	_	_	_	-	1	_	
			Lead (%)	_	_	_	_	-	_	4	_	
			Silver (g/t)	_	_	_	-	_	_	57	-	
Novo-	35.5%	OC	(Mt)	-	-	_	_	_	-	30	40	AC
Leninogorsko	уе		Zinc (%)	-	-	-	-	-	-	5	4	
			Lead (%)	-	-	_	_	-	-	2	1	
			Copper (%)	-	-	_	_	_	-	0.2	0.2	
			Silver (g/t)	-	-	-	-	-	-	40	33	
			Gold (g/t)	-	-	_	_	-	-	2	2	
Total Polymeta	allic Kazzino	;	(Mt)	87.2	27.7	100	152	190	180	97	94	
			Zinc (%)	3.98	3.42	1.9	2.8	2.9	2.9	2	3	
			Lead (%)	1.54	0.89	0.6	1.1	1.0	1.1	1	0.7	
			Copper (%)	0.33	1.20	0.3	0.2	0.3	0.3	0.3	0.3	
			Silver (g/t)	19.5	36.0	17	17	18	20	31	28	
			Gold (g/t)	0.43	1.42	0.9	0.6	0.7	0.7	1	2	
Vasilkovskoye	e 69.7%	ос	(Mt)	87.0	61.5	40	22	127	83	26	7.6	AC
(Gold)			Gold (g/t)	2.2	2.2	2.1	2.2	2.2	2.2	1.7	1.9	

Kazzinc Ore Reserves

(Gold)			Gold (g/t)	2.3	2.0	2.2	2.0	2.3	2.0	
Vasilkovskoye	69.7%	OC	Ore (Mt)	70.6	64.2	30	23	101	87	AC
			Gold (g/t)	0.20	1.15	0.6	0.8	0.3	1.0	
			Silver (g/t)	16.1	32.3	20	27	17	29	
			Copper (%)	0.14	0.95	0.6	0.8	0.2	0.9	
			Lead (%)	1.57	0.72	0.8	0.6	1.4	4.2	
rotari orymetanic r	WELLING		(iiii) Zinc (%)	3.88	2.80	4.8	5.5	4.1	4.2	
Total Polymetallic	(azzinc		(Mt)	54.6 70.7	14.4	39 20	15	46 91	30	
			Lead (%) Silver (g/t)	7.43 54.6	-	39	_	6.4 46	-	
Ushkatyn			Zinc (%)	0.09	-	0.1 5.5	_	0.1 6.4	-	
Zhairemsky Ushkatvn	69.7%	OC	Ore (Mt)	0.88	-	1.0	-	1.9	-	AC
7hairamaku	60.7%	00	Silver (g/t)	5.84	_	2.0	_	5.6	_	
			Lead (%)	1.22	-	0.9	-	1.2	-	
Dalnezapadny			Zinc (%)	4.29	-	3.9	-	4.3	-	
Zhairemsky	69.7%	OC	Ore (Mt)	35.4	-	2.6	-	38	-	AC
71	CO 70/	00	Silver (g/t)	25.4		27	_	25	_	
			Lead (%)	2.40	-	0.9	_	2.4	-	
Zapadny			Zinc (%)	4.06	-	4.6	-	4.1	-	
Zhairemsky	69.7%	OC	Ore (Mt)	22.5	-	0.5	-	23	-	AC
Stockpiles			Zinc (%)	-	-	22	22	22	22	
Shaimerden	69.7%	OC	Ore (Mt)	-	-	1.8	1.8	1.8	1.8	AC
			Gold (g/t)	0.4	0.4	0.3	0.3	0.4	0.4	
			Silver (g/t)	24	23	16	20	22	22	
			Copper (%)	2.03	2.0	1.3	1.9	2.0	2.0	
			Lead (%)	0.57	0.5	0.3	0.4	0.5	0.5	
			Zinc (%)	3.81	3.5	1.7	3.5	3.4	3.5	
Shubinsky	69.7%	UG	Ore (Mt)	0.12	0.22	0.03	0.07	0.2	0.3	AC
			Gold (g/t)	0.51	0.48	0.8	0.8	0.7	0.7	
			Silver (g/t)	7.64	10.2	10	11	10	11	
			Copper (%)	0.35	0.66	0.6	0.6	0.5	0.6	
			Lead (%)	0.70	1.86	0.8	1.3	0.8	1.4	
- -			Zinc (%)	4.32	6.99	5.0	6.2	4.9	6.4	
Tishinsky	69.7%	UG	Ore (Mt)	0.81	0.61	3.2	1.8	4.0	2.4	AC
			Gold (g/t)	1.5	1.6	1.1	1.3	1.3	1.5	
			Silver (g/t)	19.5	19.7	20	20	19	20	
			Copper (%)	0.5	0.5	0.5	0.6	0.5	0.6	
			Lead (%)	0.5	0.6	0.4	0.4	0.5	0.5	
Ridder-Sokolity	09.7 /0	00	Zinc (%)	1.2	1.2	1.0	1.0	1.1	1.1	AC
Ridder-Sokolny	69.7%	UG	Gold (g/t) Ore (Mt)	0.47	0.5	0.4	0.4	0.4	0.4	AC
			Silver (g/t)	50.7	57.2	40	51	44	54	
			Copper (%)	1.94	1.7	1.5	1.5	1.6	1.6	
			Lead (%)	0.85	0.8	0.7	0.9	0.8	0.8	
			Zinc (%)	6.06	5.0	4.5	5.3	5.0	5.2	
Maleevsky	69.7%	UG	Ore (Mt)	2.77	5.0	4.9	5.0	7.6	10	AC

Notes

Remaining mine life: different for each mine, ranging from 5 to 20 years. Expiry date of relevant mining/concession licences: different for each mine, ranging from February 2019 to March 2041. There is a routine licence extension procedure in Kazakhstan which Kazzinc undertakes as required.

Maleevsky: It is a typical syngenetic VMS deposit hosting ores of sulphide-polymetallic formation (with associated gold and silver).

The mined material from Maleevsky during 2016 was 2.3Mt at 5.8% Zn, 0.9% Pb, 1.5% Cu, 55g/t Ag and 0.5g/t Au.

Ridder-Sokolny: The gold-polymetallic deposit is also VMS type of syngenetic deposits hosting ores of gold bearing sulphide-polymetallic formation.

Mineral Resource changes are due to mining exploitation during 2016.

For Ridder-Sokolny, Mineral Resources and Ore Reserves are reported as totals for Polymetallic and Cu-type mineralisation combined.

The mined material from Ridder-Sokolny during 2016 was 1.5Mt at 0.5% Zn, 0.2% Pb, 0.7% Cu, 6g/t Ag and 2.1g/t Au.

Tishinsky: It is a syngenetic VMS deposit of Au- and Agbearing sulphide polymetallic ores.

Tishinsky Mineral Resources are different from previous estimates, due to several factors, including reinterpretation of geological wireframes, sterilisation of resources, and ongoing exploration.

The Ore Reserve estimate has also increased, as a result of a change in cut-off.

The mined material from Tishinsky during 2016 was 1.0Mt at 5.1% Zn, 1.0% Pb, 0.5% Cu, 11g/t Ag and 0.7g/t Au.

Shubinsky: The polymetallic deposit is a syngenetic VMS deposit hosting ores of sulphide-polymetallic formation (with associated gold and silver).

The mined material from Shubinsky during 2016 was 0.2Mt at 2.6% Zn, 0.5% Pb, 1.3% Cu, 22 g/t Ag and 0.5 g/t Au was extracted.

- Shaimerden: The Shaimerden stockpile is composed of highgrade, crushed zinc oxide ore which is not amenable to the concentration process; it is directly sent to the Ridder Zinc refinery. Total material processed during 2016 was 84kt at 20.6% Zn.
- Dolinnoe and Obruchevskoe: The Dolinnoe and Obruchevskoe deposits are situated in the centre and deepest south-eastern portions respectively of the Ridder mining district in the Rudnyi Altay geotectonic block. Gold is the main mineral of economic interest at Dolinnoe; veinletdisseminated polymetallic mineralisation predominates at Obruchevskoe.

A feasibility study, regarding the mining of both deposits is underway.

Zhairemsky: The various iron, manganese, barite and polymetallic deposits of the Zhairemsky area, central Kazakhstan were discovered by geological and geophysical prospecting between the 1930s and 1960s. Between 1978 and 1995, some 22 million tonnes of low-grade zinc-lead ore including barite-dominated mineralisation were mined. As of 1996, focus was set on manganese and iron ore production.

Resource and reserve data was generated from a feasibility study completed this year.

- **Uzhynzal:** The Uzhynzal deposit is located in central Kazakhstan, in the same belt as the Zhairemsky deposits. It has been identified as a sedimentary exhalative deposit. Its Pb-Zn ores show close spatial correlations with barite and manganese ores. The deposit is made up of an oxide cap, containing mainly Pb-oxide ores, while the sulphide portion of the deposit contains both Zn and Pb sulphides.
- **Novo-Leninogorskoye:** The Novo-Leninogorskoye deposit is part of the Ridder-Sokolny group of VMS polymetallic deposits. Novo-Leninogorskoye was discovered in 1981 and was explored between 1981 and 1985. Two styles of mineralisation can be found at Novo-Leninogorskoye, barite-polymetallic and polymetallic with the mineralisation hosted by siltstones and quartzites.

Resource tonnages and grade changes are linked to a new geological interpretation and statistical interpolation.

Vasilkovskoye: It is a gold deposit of epigenetic (stockwork) type and beresite subtype of deposits hosting ores of goldquartz formation.

The Resource data reported is based on a new ore model updated using 17 new drill holes drilled in 2016; the Resource was constrained in a newly designed pit shell.

Ore Reserves were estimated based upon an updated Resource interpretation and interpolation parameters; the Reserve was constrained by a new pit design, which is fully encompassed in the Resource pit shell.

The mined material from Vasilkovskoye during 2016 was 8.4Mt at 2.1g/t Au.

The Komarovsky gold mine was sold in 2016.

Competent Person

AC = Aline Côté, Project Manager, Glencore Zinc, (OGQ).

Australia Mineral Resources (Mount Isa, McArthur River)

	Attributable	Mining			d Mineral urces	Indicated Reso		Measur Indicated F		Inferred Reso		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Mount Isa	100%											
Black Star		OC	(Mt)	7.9	10.7	1.9	2.0	10	13	0.8	0.8	BY
Open Cut			Zinc (%)	5.22	5.41	3.5	3.6	4.9	5.1	5	5	
			Lead (%)	4.66	4.26	2.7	2.7	4.3	4.0	4	4	
			Silver (g/t)	81.4	72.8	46	46	75	68	70	70	
Mount Isa Open P	it	OC	(Mt)	45.3	45.3	191	191	236	236	120	120	BY
Excl. Black Star			Zinc (%)	4.13	4.13	3.6	3.6	3.7	3.7	3	3	
			Lead (%)	3.95	3.95	2.7	2.7	2.9	2.9	2	2	
			Silver (g/t)	80.9	80.9	57	57	62	62	50	50	
George Fisher	100%											
South (P49) Oreb	odies	UG	(Mt)	26.6	34.2	24	26	50	60	30	17	NS
			Zinc (%)	8.29	7.87	8.1	7.6	8.2	7.8	8	7	
			Lead (%)	5.64	5.30	4.7	4.7	5.2	5.0	5	5	
			Silver (g/t)	122	106	94	92	109	100	90	90	
North (L72) Orebo	odies	UG	(Mt)	31.3	43.2	121	103	153	147	48	68	NS
			Zinc (%)	9.76	8.17	8.9	7.2	9.0	7.5	9	7	
			Lead (%)	3.96	3.53	3.5	2.9	3.6	3.1	3	3	
			Silver (g/t)	69.0	59.7	54	47	57	51	50	50	
Handlebar Hill		OC	(Mt)	1.60	1.60	3.6	3.6	5.2	5.2	0.8	0.8	BY
Open Cut (primar	y)		Zinc (%)	7.77	7.77	6.1	6.1	6.6	6.6	5	5	
			Lead (%)	2.56	2.56	2.0	2.0	2.2	2.2	2	2	
			Silver (g/t)	40.6	40.6	35	35	37	37	30	30	
Handlebar Hill		OC	(Mt)	0.49	0.49	0.1	0.1	0.6	0.6	-	-	BY
Open Cut (oxide)			Zinc (%)	0.37	0.37	0.4	0.4	0.4	0.4	-	-	
			Lead (%)	8.52	8.52	4.1	4.1	7.8	7.8	-	-	
			Silver (g/t)	88.8	88.8	65	65	85	85	-	-	
Lady Loretta	100%	UG	(Mt)	5.28	5.28	3.9	3.9	9.2	9.2	0.3	0.3	NS
			Zinc (%)	16.3	16.3	14	14	15	15	11	11	
			Lead (%)	6.40	6.40	4.6	4.6	5.6	5.6	5	5	
			Silver (g/t)	104	104	86	86	96	96	90	90	
Total Mount Isa			(Mt)	118	141	340	330	460	470	200	210	
			Zinc (%)	7.20	6.86	5.9	5.2	6.2	5.7	5	5	
			Lead (%)	4.49	4.26	3.1	2.9	3.5	3.3	3	3	
			Silver (g/t)	87.5	80.3	59	57	66	64	60	50	
McArthur River	100%											
Open Cut		OC	(Mt)	123	118	56	52	180	170	-	_	KM
			Zinc (%)	9.94	10.2	8.2	8.7	9.4	9.7	_	_	
			Lead (%)	4.64	4.81	3.9	4.2	4.4	4.6	_	_	
			Silver (g/t)	46.9	48.6	41	44	45	47	_	_	
Woyzbun South Z	one	UG	(Mt)	_	_	8.3	8.3	8.3	8.3	_	_	KM
,	-		Zinc (%)	_	_	14	14	14	14	_	_	
			Lead (%)	_	_	5.6	5.6	5.6	5.6	_	_	
			Silver (g/t)	_	_	58	58	58	58	_	_	
Total McArthur R	iver		(Mt)	123	118	64	60	190	180	_	_	
			Zinc (%)	9.94	10.2	8.9	9.4	9.6	9.9	_	_	
			Lead (%)	4.64	4.81	4.1	4.4	4.5	4.7	_	-	

Australia Ore Reserves (Mount Isa, McArthur River)

	Attributable	Mining		Proved Ore		Probable Or		Total Ore		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Mount Isa	100%									
Black Star Open Cut		OC	Ore (Mt)	-	2.91	-	0.10	-	3.0	BY
			Zinc (%)	-	5.59	-	6.8	-	5.6	
			Lead (%)	-	2.87	-	2.5	-	2.9	
			Silver (g/t)	-	50.8	-	44	-	51	
George Fisher	100%									
South (P49) Orebodies		UG	Ore (Mt)	5.99	7.31	10	9.0	16	16	GB
			Zinc (%)	6.35	6.23	6.9	6.3	6.7	6.3	
			Lead (%)	5.00	4.99	4.8	4.8	4.9	4.9	
			Silver (g/t)	122	98.2	104	93	111	95	
North (L72) Orebodies		UG	Ore (Mt)	10.7	14.6	42	35	53	49	GB
			Zinc (%)	8.39	7.87	7.8	6.9	7.9	7.2	
			Lead (%)	3.95	3.92	3.6	3.4	3.7	3.6	
			Silver (g/t)	73.2	66.1	60	56	63	59	
Handlebar Hill Open Cut		OC	Ore (Mt)	0.49	0.49	-	_	0.49	0.49	BY
(oxide)			Zinc (%)	0.37	0.37	-	-	0.37	0.37	
			Lead (%)	8.52	8.52	-	_	8.52	8.52	
			Silver (g/t)	88.8	88.8	_	_	88.8	88.8	
Lady Loretta	100%	UG	Ore (Mt)	5.18	5.18	3.3	3.3	8.5	8.5	SVDM
			Zinc (%)	14.5	14.5	11	11	13	13	
			Lead (%)	5.90	5.90	3.8	3.8	5.1	5.1	
			Silver (g/t)	97.0	97.0	70	70	87	87	
Total Mount Isa			(Mt)	22.4	30.5	55	47	78	78	
			Zinc (%)	9.08	8.27	7.9	7.1	8.2	7.5	
			Lead (%)	4.78	4.49	3.8	3.7	4.1	4.0	
			Silver (g/t)	92.1	77.9	69	64	75	69	
McArthur River	100%	00	(Mt)	71.2	49.4	45	53	117	102	DH
			Zinc (%)	10.6	12.1	7.4	8.3	9.4	10	
			Lead (%)	5.00	5.66	3.6	4.0	4.5	4.8	
			Silver (g/t)	50.1	57.5	37	41	45	49	

Notes

Mount Isa

Black Star Open Cut: Lead-zinc-silver mineralisation occurs in galena and sphalerite-rich bedding parallel horizons in dolomitic and variably carbonaceous pyritic shales and siltstones.

The Black Star Open Cut resource is additional to the Mount Isa Open Pit resource.

The resource mineralisation occurs inside a designed Stage 7 pit based on a pit optimisation utilising Measured and Indicated Resources.

A total of 3.6Mt at 5.1% Zn, 2.6% Pb and 64g/t Ag was depleted from the Ore Reserves in the 12 months to 31 December 2016 due to mining operations, completing the Stage 5 Reserve.

The open cut was moved to a care and maintenance phase during the last quarter of 2016 after mining out the existing reserves.

The Black Star Open Cut is located on Mining Lease ML8058 which expires on 30 November 2036.

Isa Open Pit: Lead-zinc-silver mineralisation occurs in galena and sphalerite-rich bedding parallel horizons in dolomitic and variably carbonaceous pyritic shales and siltstones. Approximately 85% of the lead-zinc-silver resource is primary sulphide; the remainder being considered as transitional (mixed primary sulphide and secondary oxide/carbonate) mineralisation.

The Black Star Open Cut resource has been removed from the Isa Open Pit and is reported separately.

Pit optimisation was conducted using Measured, Indicated and Inferred Mineral Resources, and a pit shell was generated to constrain the Mineral Resource.

The copper resource inside the Isa Open Pit has not been included here; it is reported separately in the Copper section of this report. The Isa Open Pit is located on Mining Lease ML8058 which expires on 30 November 2036.

George Fisher Mine

North (L72) & South (P49) Orebodies: Lead-zinc-silver mineralisation occurs in galena and sphalerite-rich bedding parallel horizons in dolomitic and variably carbonaceous pyritic shales and siltstones.

Orebody and structural interpretation, modelling and classification of the Mineral Resource was completed on the basis of additional geological information and improved systems.

Changes to the Mineral Resources are due to geological modelling and the addition of new drilling information.

Mine production for the period January 2016 to December 2016 totalled 3.1Mt at 7.6% Zn, 4.5% Pb and 84g/t Ag.

The mine is located on Mining Lease ML8058 which expires on 30 November 2036.

Handlebar Hill Open Cut: Lead-zinc-silver mineralisation occurs in galena and sphalerite-rich bedding parallel horizons in dolomitic and variably carbonaceous pyritic shales and siltstones.

The Handlebar Hill Open Cut resource is up dip of and additional to the George Fisher South resource.

Material from the oxidised portion of the mineralisation has been reported as a Mineral Resource and Ore Reserve following confirmation of an achievable treatment path.

No depletion has occurred through mining during 2016.

The Handlebar Hill Open Cut is located on Mining Lease ML8058 which expires on 30 November 2036. The mine was placed in care and maintenance in July 2014.

Lady Loretta

Lead-zinc-silver mineralisation occurs in a galena and sphalerite rich massive sulphide lens located in carbonaceous pyritic shales and siltstones. The deposit consists of a tight syncline dislocated by a number of major faults. The deeper and high grade portion of the deposit reaches 500m below the surface.

There was no mine production for 2016 as Lady Loretta was placed on care and maintenance during the last quarter of 2015 with a view to restarting in the appropriate economic conditions. There is no change to the Resource or Reserve for 2016.

The Mining Lease (ML5568) is current until 31 January 2026.

McArthur River Mine

Zinc-lead-silver mineralisation occurs predominantly as ultra-fine bedded parallel sphalerite and galena rich bands hosted by dolomitic and carbonaceous pyritic siltstones, graded beds and chaotic debris flow breccias.

Mineral Resources and Ore Reserves are based on the approved mine plan.

All relevant Modifying Factors for the conversion of Mineral Resources to Ore Reserves have been considered, with confidence levels in these factors reflected in the classification categories. There are no known issues that could materially affect the estimates.

The Ore Reserves and Mineral Resources have been depleted during 2016 by a total of 3.4Mt at 7.8% Zn and 3.3% Pb. Mining activities are temporarily progressing at a reduced rate, as announced by Glencore in October 2015. Changes beyond depletion were primarily due to revised cut-off calculations.

Mineral Resources and Ore Reserves are located within leases that are valid to 2043, with the current mine plan ending in 2040.

- BY = Ben Young, Mine Operations Manager, Glencore Zinc, (AusIMM).
- DH = Drew Herbert, Mining Consultant, The Minserve Group Pty Ltd, (AusIMM).
- GB = Guilherme Barboza, Mine Planning Superintendent, Glencore Zinc, (AusIMM).
- NS = Nicholas Spanswick, Senior Mine Geologist, Glencore Zinc, (AIG).
- SVDM = Shaun Van Der Merwe, Manager Technical Services, Glencore Zinc, (AusIMM).

North America Mineral Resources (Kidd Creek, Matagami, PD1, Errington, Vermilion, Hackett River, Bell, Granisle)

Granisle)				Measure	d Mineral	Indicated	d Mineral	Measu	red and	Inferred	Mineral	
	Attributable	Mining		Reso	urces	Reso	urces	Indicated I	Resources	Reso	urces	Competent
Name of operation	interest	method	Commodity		31.12.15		31.12.15		31.12.15		31.12.15	Person
Kidd Creek	100%	UG	(Mt)	11.7	13.4	1.2	1.1	13	15	3.5	3.4	BD
			Zinc (%)	4.32	4.50	5.6	6.1	4.4	4.6	7	7	
			Copper (%)	1.99	2.01	1.6	1.6	2.0	2.0	2	2	
	4000/		Silver (g/t)	51	52	41	44	50	51	60	60	
Matagami	100%		(* *)									
Bracemac-McLe	od	UG	(Mt)	2.24	2.38	4.2	4.1	6.4	6.5	-	0.16	JD
			Zinc (%)	5.59	5.55	5.9	7.3	5.8	6.7	-	7	
			Copper (%)	1.03	1.28	1.1	1.2	1.1	1.2	-	2	
			Silver (g/t)	18.0	17.4	29	30	25	25	-	20	
			Gold (g/t)	0.36	0.35	0.6	0.8	0.5	0.6	-	0.6	
PD-1	100%	OC/UC	· · ·	0.55	0.55	1.0	1.0	1.6	1.6	-	-	GR
			Zinc (%)	4.22	4.22	5.0	5.0	4.7	4.7	-	-	
			Copper (%)	0.82	0.82	1.3	1.3	1.1	1.1	-	-	
			Silver (g/t)	19.7	19.7	20	20	20	20	-	-	
			Gold (g/t)	0.13	0.13	-	-	-	-	-	-	
Errington	100%	UG	(Mt)	6.70	6.70	2.3	2.3	9.0	9.0	-	-	AC
			Zinc (%)	3.94	3.94	4.3	4.3	4.0	4.0	-	-	
			Lead (%)	1.10	1.10	1.3	1.3	1.2	1.2	-	-	
			Copper (%)	1.15	1.15	1.1	1.1	1.1	1.1	-	-	
			Silver (g/t)	52.0	52.0	55	55	53	53	-	-	
			Gold (g/t)	0.84	0.84	0.8	0.8	0.8	0.8	_	-	
Vermilion	100%	UG	(Mt)	2.80	2.80	0.4	0.4	3.2	3.2	-	-	AC
			Zinc (%)	4.22	4.22	5.3	5.3	4.4	4.4	-	-	
			Lead (%)	1.16	1.16	1.3	1.3	1.2	1.2	-	-	
			Copper (%)	1.34	1.34	1.1	1.1	1.3	1.3	-	-	
			Silver (g/t)	52.6	52.6	56	56	53	53	-	-	
			Gold (g/t)	0.91	0.91	1.1	1.1	0.9	0.9	-	-	
Hackett River	100%	OC/UC	G (Mt)	-	-	27	27	27	27	60	60	AC
			Zinc (%)	-	-	4.5	4.5	4.5	4.5	3.5	3.5	
			Lead (%)	-	-	0.6	0.6	0.6	0.6	0.5	0.5	
			Copper (%)	-	-	0.5	0.5	0.5	0.5	0.4	0.4	
			Silver (g/t)	-	-	130	130	130	130	150	150	
			Gold (g/t)	-	-	0.3	0.3	0.3	0.3	0.2	0.2	
Total Zinc Nort	h America		(Mt)	24.0	25.8	36	36	60	62	60	60	
			Zinc (%)	4.32	4.42	4.7	4.9	4.6	4.7	4	4	
			Lead (%)	0.44	0.41	0.5	0.5	0.5	0.5	0.5	0.5	
			Copper (%)	1.56	1.62	0.7	0.7	1.0	1.1	0.5	0.5	
			Silver (g/t)	47.7	48.2	107	107	83	83	140	140	
			Gold (g/t)	0.38	0.35	0.4	0.4	0.4	0.4	0.2	0.2	
Bell	100%	OC	(Mt)	57.0	57.0	200	200	257	257	100	100	BD
			Copper (%)	0.41	0.41	0.4	0.4	0.4	0.4	0.4	0.4	
			Gold (g/t)	0.18	0.18	0.2	0.2	0.2	0.2	0.1	0.1	
Granisle	100%	OC	(Mt)	18.0	18.0	55	55	73	73	20	20	BD
			Copper (%)	0.34	0.34	0.3	0.3	0.3	0.3	0.3	0.3	
			Gold (g/t)	0.11	0.11	0.1	0.1	0.1	0.1	0.1	0.1	
Total Copper N	orth Americ	са	(Mt)	75.0	75.0	255	255	330	330	120	120	
•••			Copper (%)	0.39	0.39	0.4	0.4	0.4	0.4	0.4	0.4	
			Gold (g/t)	0.16	0.16	0.2	0.2	0.2	0.2	0.1	0.1	

North America Ore Reserves (Kidd Creek, Matagami)

	Attributable	Mining		Proved Ore	Reserves	Probable Or	e Reserves	Total Ore	Reserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Kidd Creek	100%	UG	Ore (Mt)	6.62	6.99	1.8	3.3	8.4	10	AM
			Zinc (%)	3.76	4.06	5.2	4.7	4.1	4.3	
			Copper (%)	1.98	2.02	1.7	1.8	1.9	2.0	
			Silver (g/t)	49	50	44	46	48	49	
Matagami	100%									
Bracemac-McLeod		UG	Ore (Mt)	1.04	1.21	3.0	4.2	4.0	5.4	AC
			Zinc (%)	7.05	6.24	6.8	6.8	6.8	6.7	
			Copper (%)	0.93	1.29	1.2	1.1	1.1	1.1	
			Silver (g/t)	18.7	16.9	30	27	27	25	
			Gold (g/t)	0.41	0.37	0.7	0.6	0.6	0.6	
Total North America			(Mt)	7.66	8.20	4.8	7.5	12	16	
			Zinc (%)	4.21	4.38	6.2	5.9	5.0	5.1	
			Copper (%)	1.84	1.91	1.4	1.4	1.7	1.7	
			Silver (g/t)	44.9	45.1	35	35	41	40	
			Gold (g/t)	0.06	0.05	0.4	0.4	0.2	0.2	

Notes

- Kidd Creek: Kidd Creek is a VMS Cu-Zn-Ag deposit. Mineralisation occurs within a rhyolitic volcanic/volcaniclastic sequence as massive sulphide lenses of dominantly pyrite-pyrrhotite-sphalerite-galena-rich ores that are underlain by copper (chalcopyrite) stringer zones.
 - Ore Reserves are based on the approved mining plan to 9600ft depth.
 - Mineral Resources and Ore Reserves changes are the result of mining drawdown, with some adjustments due to updated mine design, cost reductions, and commodity pricing changes.
 - January to December 2016 production totalled 2.3Mt at 4.0% Zn, 1.8% Cu and 44g/t Ag.
 - The majority of Ore Reserve in the Probable category reflects geotechnical and economic uncertainty during the latter years of the mine plan, rather than geological uncertainty.
 - Mine life is anticipated to be 5.3 years (Early 2022). There are no known land tenure issues that could affect the production plan.
- **Bracemac-McLeod:** The Bracemac-McLeod deposits comprise a cluster of polymetallic VMS lenses similar to other deposits mined historically in the Matagami mining camp, except for their generally thinner and complex morphology. The deposits are located 4.5km southeast of Glencore's Matagami concentrator, at depths ranging from 100m to 1,400m below the surface. The deepest deposit, the McLeod Deep lens, was delineated in 2013.

The geological models and the resource block models for all Bracemac lenses (Main, KT and Upper) and McLeod A, B, C and West lenses are revised regularly based on new information derived from underground delineation diamond drilling and geological mapping. The residual Mineral Resources are updated on a monthly basis as mining progresses. For most massive to semi-massive sulphide lenses, there is significant lower grade stringer-type mineralisation in the immediate footwall. Several of these stringer zones are now categorised as Measured Resources. The geological model and the resource block model for the McLeod Deep lens is now based on 56 intersections drilled in 2010 and 2013 on 50m spacing. The Mineral Resource for the McLeod Deep lens is now classified as Indicated. The Mineral Reserve of McLeod Deep was generated through the completed feasibility study.

Mine production for year 2016 was 0.94Mt grading 5.9% Zn, 1.1% Cu, 19g/t Ag and 0.37g/ Au.

The Bracemac-McLeod mine is contained on two mining leases expiring in April 2033.

PD-1: The PD1 deposit is a polymetallic VMS deposit of the same age and derived from the same ore-forming hydrothermal system as the rest of the Matagami camp deposits. It is located 40km west of Glencore's Matagami concentrator and offices.

The PD1 deposit was discovered in 1974. A total of 50 historical drill holes were drilled between 1974 and 1984. In 2010, 25 additional holes were drilled in the upper portion of the deposit above 100m vertical depth, including 3 duplicate holes to validate the historical data. The deposit is located on a mining claim owned by Glencore Canada Corporation.

Errington: The Errington deposit is a polymetallic massive sulphide located in the Sudbury Basin, Ontario. The 5 lenses that make up this deposit are hosted by sedimentary rocks of the Vermilion Formation at the contact of the Onaping and Onwatin formations. The deposits formed by replacement of carbonate mounds and carbonaceous tuffs fuelled by heat from the Sudbury Igneous Complex. Additional enrichment and concentration of metals was provided by deformation from the South Range shear zone.

Discovered in the 1920's, the Errington underground development began in 1924 and ended in 1928. A total of 129,713t of ore were produced from Errington.

In order to confirm both the morphology and grade of these historical resources based on 1,070 historical diamond drill holes, 175 supplementary diamond drill holes were carried out for a total of 50,000m. The resources have been interpolated by ID2.

Vermilion: The Vermilion deposit is a polymetallic massive sulphide located in the Sudbury Basin, Ontario. The 17 lenses that make up this deposit are hosted by sedimentary rocks of the Vermilion Formation at the contact of the Onaping and Onwatin formations. The deposits formed by replacement of carbonate mounds and carbonaceous tuffs fuelled by heat from the Sudbury Igneous Complex.

Additional enrichment and concentration of metals was provided by deformation from the South Range shear zone.

Although discovered in the 1920's, the Vermilion underground development only started between 1952 and 1957. A total of 22,172t of ore were hoisted at Vermillion and stockpiled circa since 1958. The stockpile was shipped to Kidd Creek for processing in 1992.

In order to confirm both the morphology and grade of these historical resources based on 609 historical diamond drill holes, 35 supplementary diamond drill holes were carried out for a total of 10,000m. The resources have been interpolated by ID2.

Hackett River Project: The Hackett River project is located in Nunavut, Canada, approximately 480km northeast of Yellowknife and 105km south–southwest of the community of Bathurst Inlet, which is located on the Arctic Ocean.

The Hackett River deposits are situated within the Slave Structural Province, a predominantly Archaean granitegreenstone-sedimentary terrain that lies between Great Slave Lake and Coronation Gulf. The deposits are typical of VMS deposits. Sulphide mineralisation occurs as tabular semi-massive to massive lenses. Stringer sulphide minerals are developed beneath the lower massive lenses in stratiform to pipe-like configurations. Stratiform disseminated sulphides envelop the massive sulphide and stringer zones.

The four principal sulphide occurrences from west to east are the East Cleaver, Boot Lake, Main Zone, and Jo Zone deposits. These deposits were defined as economically viable mineable resources, following boundaries of open cut vs underground mining, through a Preliminary Economic Assessment prior to Glencore's acquisition (2010). A Pre-Feasibility study is now underway and is revisiting both the mining methods and boundaries between open cut and underground; for these reasons the Mineral Resources are only distinguished through their categories instead of exploitation method.

Following the exploration drilling campaign of 2013, which added 114 drill holes totalling 39,000m, we have undertaken the re-interpretation of an *in situ* resource using Zn equivalent values instead of considering a block dollar value. The resources have been interpolated by ID2.

Bell/Granisle: Bell and Granisle are porphyry copper-gold deposits located at Babine Lake. The Babine deposits are associated with calc-alkaline magmatic rocks. They were formed in the roots of Eocene volcanoes built upon continental crust. Erosion has removed most of the poorly consolidated volcanic piles, exposing the mineral deposits.

Recorded past production from the Bell mine from 1972 to 1992 totalled 77.2Mt averaging 0.47% Cu with an average waste to ore ratio of 0.95:1.

Past production for Granisle from 1966 to 1982 totalled 52.7Mt averaging 0.47% Cu with an average waste to ore ratio of 1.37.

The latest Mineral Resource estimate does not include the additional 25 holes (12,260m) drilled in 2012. The results of the Pre-Feasibility Study delivered in May 2013 are still under review.

A life of mine of 19 years has been estimated, at a processing rate of 50,000 tonnes per day.

There are no known land tenure issues and the mining leases are renewed yearly.

Competent Persons

AC = Aline Côté, Project Manager, Glencore Zinc, (OGQ).

- AM = Adrianus Moerman, Principal Mine Engineer, Glencore Zinc, (APEO).
- BD = Benoit Drolet, Senior Resource Geologist, Glencore Zinc (APGO).
- GR = Gilles Roy, District Geologist, Glencore Zinc, (OGQ).
- JD = Julie Drapeau, Chief Mine Geologist, Glencore Zinc (OGQ).

	Attributable		_	Measureo Resou	urces	Indicated Resou	urces	Measure Indicated R	esources	Inferred Resou	irces (Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Los Quenuales	97.6%											
Iscaycruz		UG/OC	(Mt)	0.88	0.88	4.3	3.7	5.2	4.6	2	0.8	AC
			Zinc (%)	6.64	6.64	6.2	6.4	6.3	6.5	5	7	
			Lead (%)	0.13	0.13	0.7	0.7	0.6	0.6	0.8	0.9	
			Copper (%)	0.71	0.71	0.4	0.4	0.5	0.4	0.4	0.5	
			Silver (g/t)	23.5	23.5	38	39	35	36	40	20	
Yauliyacu		UG	(Mt)	4.69	-	15	15	20	15	0.5	3	AC
			Zinc (%)	3.89	-	3.7	2.5	3.8	2.5	2	2	
			Lead (%)	2.28	-	1.8	0.4	1.9	0.4	2	0.5	
			Copper (%)	0.52	-	0.4	0.3	0.5	0.3	0.3	0.3	
			Silver (g/t)	302	-	240	100	260	100	70	170	
Illapa	45%											
Bolivar		UG	(Mt)	1.62	1.56	0.9	2.8	2.5	4.4	4.3	1.5	AC
			Zinc (%)	16.5	7.75	15	9.0	16	8.5	9	8	
			Lead (%)	1.8	0.83	1.6	1.1	1.7	1.0	0.9	2	
			Silver (g/t)	396	151	313	214	366	192	180	210	
Porco		UG	(Mt)	0.83	0.74	0.60	0.43	1.4	1.2	1.0	1.0	AC
			Zinc (%)	9.94	9.81	10	10	10	10	10	10	
			Lead (%)	0.66	0.61	0.7	0.8	0.7	0.7	1	1	
			Silver (g/t)	101	104	101	100	101	103	80	80	
Sinchi Wayra	100%			-	-	-		-				
Caballo Blanco		UG	(Mt)	0.36	0.47	0.4	3.7	0.8	4.2	1.7	1.5	AC
			Zinc (%)	13.1	11.8	13	9.1	13	9.4	12	9	
			Lead (%)	3.9	1.91	3.5	1.2	3.7	1.3	2	1	
			Silver (g/t)	392	127	350	79	370	84	190	80	
Aguilar	100%		0.1101 (3.1)									
Aguilar Pit		OC	(Mt)	0.28	0.37	0.7	1.0	1.0	1.4	0.3	0.02	AC
, iguilar i it		00	Zinc (%)	2.77	2.43	2.8	2.5	2.8	2.5	2	1	7.0
			Lead (%)	2.38	2.04	2.4	2.5	2.4	2.4	2	4	
			Silver (g/t)	69.1	58.1	60	62	63	61	60	80	
Aguilar		UG	(Mt)	1.58	2.46	0.8	0.8	2.4	3.3	0.1	0.1	AC
Aguilai		00	Zinc (%)	8.33	7.96	10	11	8.9	8.8	13	11	70
			Lead (%)	9.23	8.59	8.2	10	8.9	9.0	7	7	
			Silver (g/t)	9.23	187	160	175	179	184	140	160	
Perkoa	90%	UG/OC	(Mt)	3.04	2.10	1.2	2.7	4.3	4.8	140	2.3	AC
	5070	00,00	Zinc (%)	15.5	17.4	1.2	12	4.3	4.8	1.0	2.3	70
Rosh Pinah	80.1%	UG	(Mt)	3.35	2.02	6.6	6.7	10	8.7	2.9	4.4	SK/AC
	50.170	00	Zinc (%)	8.74	7.88	7.4	8.3	7.9	8.2	2.9	4.4	
			Lead (%)	1.65	1.72	1.4	0.3 1.6	1.5	0.2 1.6	1	1	
			Silver (g/t)	27.0	29.1	23	25	24	26	26	26	
Pallas Green	100%				29.1		- 25		- 20	44	44	AH
			(Mt)	-						44	44	АП
Tobermalug Zon	6	UG	Zinc (%)	-	-	-	-	-	-		1	
Total Other 7:			Lead (%)	46.6	-	- 21	- 27	-	-	1		
Total Other Zind	•		(Mt)	16.6	10.6	31	37	48	48	58	59	
			Zinc (%)	9.27	9.78	5.9	6.1	7.1	6.9	7	7	
			Lead (%)	2.19	2.65	1.7	1.0	1.8	1.4	1	1	
			Copper (%)	0.18	0.06	0.3	0.2	0.2	0.1	0.1	0.1	
			Silver (g/t)	163	88.0	150	80	154	82	24	20	

Other Zinc Mineral Resources (Los Quenuales, Illapa, Sinchi Wayra, AR Zinc, Perkoa, Rosh Pinah, Pallas Green)

Nome of operation	Attributable	Mining	Corrected	Proved Ore		Probable Ore		Total Ore R		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Los Quenuales	97.6%	00	One (N4t)	0.40	0.40	0.0	0.7	0.4	0.0	10
Iscaycruz pit		OC	Ore (Mt)	0.18	0.18	2.9	2.7	3.1	2.9	AC
			Zinc (%)	5.25	4.90	6.1	6.1	6.1	6.0	
			Lead (%)	0.50	0.49	0.8	0.8	0.8	0.8	
			Copper (%)	0.32	0.30	0.2	0.2	0.2	0.2	
1			Silver (g/t)	28.2	28.3	44	45	43	44	
Iscaycruz		UG	Ore (Mt)	-	_	0.5	_	0.5	-	AC
			Zinc (%)	-	-	6.0	-	6.0	-	
			Copper (%)	-	-	0.9	-	0.9	-	
			Silver (g/t)	-	_	20	_	20		
Yauliyacu		UG	Ore (Mt)	1.63	-	5.1	3.8	6.7	3.8	AC
			Zinc (%)	2.30	-	2.7	2.0	2.6	2.0	
			Lead (%)	1.47	-	1.3	0.3	1.4	0.3	
			Copper (%)	0.32	-	0.3	0.2	0.3	0.2	
			Silver (g/t)	239	-	170	130	187	130	
Illapa	45%									
Bolivar		UG	Ore (Mt)	0.82	-	0.5	0.93	1.3	0.93	AC
			Zinc (%)	11.2	_	12	9.2	12	9.2	
			Lead (%)	1.3	-	1.3	1.1	1.3	1.1	
			Silver (g/t)	295	_	295	234	295	234	
Porco		UG	Ore (Mt)	0.23	0.16	0.45	0.70	0.67	0.86	AC
			Zinc (%)	6.6	6.24	6.2	6.7	6.3	6.6	
			Lead (%)	0.5	0.45	0.4	0.5	0.4	0.5	
			Silver (g/t)	84	118	75	75	78	83	
Sinchi Wayra	100%									
Caballo Blanco		UG	Ore (Mt)	0.16	0.07	0.26	0.32	0.42	0.39	AC
			Zinc (%)	7.71	8.98	8.3	11	8.1	11	
			Lead (%)	2.26	1.46	1.7	1.4	1.9	1.4	
			Silver (g/t)	181	104	146	92	160	94	
Aguilar	100%	UG	Ore (Mt)	1.02	1.12	0.46	0.41	1.5	1.5	AC
			Zinc (%)	6.71	7.87	9.4	10	7.6	8.5	
			Lead (%)	8.60	8.78	10	11	9.1	9.2	
			Silver (g/t)	169	194	177	151	171	182	
Perkoa	90%	UG/OC	Ore (Mt)	1.70	1.10	0.78	2.2	2.5	3.3	AC
			Zinc (%)	15.8	18.0	14	14	15	15	
Rosh Pinah	80.1%	UG	Ore (Mt)	1.6	0.83	3.5	4.9	5.1	5.7	DC/AC
			Zinc (%)	9.8	10.0	8.3	8.8	8.8	9.0	
			Lead (%)	1.0	1.52	1.7	1.7	1.5	1.6	
			Silver (g/t)	17	32	22	24	20	25	
Total Other Zinc			(Mt)	7.34	3.46	14	16	22	20	
			Zinc (%)	8.99	11.4	6.2	7.4	7.1	8.1	
			Lead (%)	1.96	3.28	1.4	1.1	1.6	1.5	
			Copper (%)	0.08	0.02	0.2	0.1	0.1	0.1	
			Silver (g/t)	120	79.5	96	69	104	71	

Other Zinc Ore Reserves (Los Quenuales, Illapa, Sinchi Wayra, AR Zinc, Perkoa, Rosh Pinah)

Notes

Iscaycruz: Zinc, lead and copper mineralisation are exposed as subvertical massive sulphide orebodies; described as skarn, breccias and carbonate replacement type along 12km corridor hosted in clay-rich limestone and dolomite rocks. Hydrothermal mineralisation assemblages are mainly composed of sphalerite, galena, pyrite and chalcopyrite distributed in five production zones named Limpe Centro, Chupa, Tinyag II, Tinyag I and Santa Este from north to south. Changes in Resource grades and tonnages come from the completed 3D model, which contains geostatistical sub-domains which enable grade enhancement.

Underground Reserves were generated at Chupa and Limpe Sur following a positive economic assessment. A feasibility study to determine underground viability at Santa Este and Olga Norte is planned for 2017/18.

The mine was on care and maintenance throughout 2016.

The expected life of Iscaycruz is 2 to 3 years based on Ore Reserves and 3 to 5 years based on Mineral Resources.

Expiry date of relevant mining/concession licences: permanent.

Yauliyacu: Main mineralisation occurs as sphalerite, galena, tetrahedrite and chalcopyrite in 60° to 80° northwest dipping narrow veins, stockwork and minor replacement massive orebodies exposed in about 5km length extension and +2km depth extension. This hydrothermal mineralisation is strongly structurally controlled and hosted in folded rock units as calcareous sandstones (red beds), conglomerates, volcanic tuffs, andesites and limestones.

Mineral Resource estimate methodology for two zones of Yauliyacu was changed from 2D polygonal to 3D kriging for the main zones and 3D ID2 for the secondary zones. The grade of the Reserve estimate reflects the Resource grade increase but is dampened by appropriate mining factors, in accordance with the mineralisation and various mining methods.

Production for 2016 was 1.4Mt grading 2.3% Zn, 0.8% Pb and 73g/t Ag.

The expected life of Yauliyacu is 3 to 4 years based on Ore Reserves and 10 to 12 years based on Mineral Resources.

Expiry date of relevant mining/concession licences: permanent.

Illapa and Sinchi Wayra: The majority of the deposits within the Illapa and Sinchi Wayra portfolio are epigenetichydrothermal base metal type vein and fault filled mineralisation hosted within a variety of lithologies from volcanic tuffs to sedimentary packages. The main mineral assemblages are composed of sphalerite, marmatite, galena, silver rich galena and silver sulfosalts. The resources are usually based on multiple structures with Porco containing over 100 different veins. The typical dimensions of these structures is +500m in length and +450m depth profile with mineralisation open at depth; average vein widths from 0.2 - 4.0m.

Caballo Blanco operational unit consists of three mines: Colquechaquita, Reserva and Tres Amigos, supplying the central plant "Don Diego" situated close to Potosi.

Regarding Bolivar, ore zones have been fully converted to 3D and reported; the Mineral Resource grades have increased as a result of new interpolation parameters, as dictated by variography. The increased Reserve grades and tonnages reflect the Resource update. Production for 2016 for each of the operations was:

Bolivar - 0.26Mt grading 9.0% Zn, 0.9% Pb and 229 g/t Ag.

Porco - 0.19Mt grading 7.6% Zn, 0.7% Pb and 101g/t Ag.

Caballo Blanco – 0.19Mt grading 7.8% Zn, 2.1% Pb and 214g/t Ag.

The expected life of the mines as a group, considering current production capacities, is an average of 2.5 years based on Ore Reserves and 7 to 10 years based on Mineral Resources.

According to the new Bolivian Constitution enacted in 2009, natural resources belong to the Bolivian people. The Bolivian State can enter into mining contracts with private investors to operate them. As with all private investors in Bolivia, Illapa and Sinchi Wayra do not hold property rights over mining resources in the country, but rather hold the right to exploit them pursuant to Bolivian legislation.

In May 2016, Glencore returned its exploitation rights in Poopo mine to the local co-operative.

Expiry date of relevant mining concessions / authorisations or contracts is different for each mine: Porco and Bolivar – July 2028 (joint venture agreement entered into in 2013); and permanent in respect of Caballo Blanco.

Aguilar: Mineralisation is classified as sedimentary exhalative type with sulphide layers in between siliciclastic and shale rocks with a post-secondary metasomatic over print between two intrusive stocks. Galena-rich, sphalerite, marmatite pyrite orebodies as lenses shape, locally brittle-style hydrothermal breccias, minor veinlets-stockworks and dissemination defines the economic portion of mineral inventories. Strike length extension of mineral geometries is variable and reaches up to 300m on north-south extension, about 55m in width and reaches up to 160m in depth.

The decrease in the underground Resources is a result of the sterilisation campaign on portions of Piqué Inferior and Capa that are impossible to access. There has been no effect on the Reserves in these areas as these zones were not considered part of the Reserve.

Production for 2016 from both underground and open pit was 0.58Mt grading 4.6% Zn, 5.1% Pb and 111g/t Ag.

The remaining mine life is approximately 3 years based on Ore Reserves and 4-5 years based on Mineral Resources.

Expiry date of relevant mining/concession licences: permanent.

Perkoa: Economic mineralisation at Perkoa occurs mainly as VMS lenses of sphalerite, galena, pyrite, and pyrrhotite. These massive sulphide lenses vary in width from 1m to 30m thick in places. These massive sulphide lenses dip at an average of 75°, striking northeast-southwest and consist of two main orebodies. Igneous intrusives have also caused endothermic and exothermic skarn like disseminated mineralisation of remobilised galena, pyrite, and to a lesser extent pyrrhotite and sphalerite.

Production for 2016 was 0.53Mt grading 15.2% Zn.

Current expected life of mine is approximately 5 years based on Ore Reserves and approximately 6 years based on Mineral Resources. Expiry date of relevant mining/concession licences: 20 March 2027.

Rosh Pinah: Sedimentary exhalative and remobilised zinc and lead sulphide ores are contained within the so-called Ore Equivalent Horizon, a stratiform horizon that is extensively

folded and thrusted, resulting in discreet, subvertical orebodies that vary in size from 0.4 - 8Mt.

Metallurgical test work is underway to improve throughput and metal recoveries from the WF3 orebody and produce a saleable Cu concentrate by 2017.

Mine production for the period January to December 2016 was 0.57Mt at 8.2% Zn and 1.7% Pb.

Expected life of mine is 7.6 years based on Ore Reserves. Rosh Pinah has previously been able to extend its expected life of mine through exploratory drilling in the area covered by its concession. Potential Life of Mine based on the conversion of Inferred Mineral Resources is 7.8 years. The expiry date of ML 39+AW (Mining Licence and Accessory Works) is 11 February 2020. EPL2616 (Exclusive Prospecting Licence) has been renewed for a period of 2 years, expiry date is 30 November 2017.

Pallas Green Project: The Pallas Green project is situated near Limerick in Southwestern Ireland. The Tobermalug zone consists of multiple, subhorizontal, stratiform lenses of Irish-type, breccia-hosted, sphalerite-galena-pyrite within a Carboniferous limestone. The lenses occur over an area 4,000m by 4,000m, and from 300m to 1,300m below surface.

The Inferred Mineral Resource is based on 370,000m of diamond drilling in 735 drill holes completed between 2005 and the end of 2014. Drill spacing is nominally 100m but 178 infill drill holes at 50m spacing have been completed. Mineralisation wireframes were built taking into account a cut-off of Zn + Pb and a minimum 3.0m true thickness, and constrain interpolation by Inverse Distance Weighting (IDW) in a block model.

No drilling was conducted in 2016, and there is no change in the Mineral Resource.

- AC = Aline Côté, Project Manager, Glencore Zinc, (OGQ).
- AH = Allan Huard, Senior Geologist, Glencore Zinc, (APGO).
- DC = Duan Campbell, Projects : Mining Engineer, Glencore Zinc, (ECSA).
- SK = Sheron Kaviua, Mineral Resources Manager, Glencore Zinc, (SACNASP).

Metals and Minerals Nickel

Integrated Nickel Operations (INO) Mineral Resources (Raglan, Sudbury)

Name of Attributable Mining operation Interest Method			Commodity	Measured Resou 31.12.16		Indicated Reso 31.12.16		Measur Indicated F 31.12.16	Resources	Inferred Reso 31.12.16	urces	Competent Person
Raglan	100%	UG	(Mt)	6.22	5.77	12.72	12.69	18.9	18.5	17	18	DP
			Nickel (%)	3.75	3.92	3.32	3.25	3.46	3.46	3.0	3.0	
			Copper (%)	0.95	0.94	0.98	0.97	0.97	0.96	0.9	0.9	
			Cobalt (%)	0.08	0.08	0.07	0.07	0.08	0.07	0.1	0.1	
			Platinum (g/t)	1.01	0.98	0.97	0.96	0.98	0.97	0.9	0.9	
			Palladium (g/t)	2.45	2.29	2.39	2.41	2.41	2.37	2.3	2.3	
Sudbury	100%	UG	(Mt)	8.70	9.86	17.19	18.67	25.9	28.5	21	15	SEK
			Nickel (%)	1.75	1.81	2.33	2.42	2.13	2.21	1.3	1.9	
			Copper (%)	1.79	1.92	0.99	1.15	1.26	1.42	3.6	3.0	
			Cobalt (%)	0.04	0.04	0.06	0.06	0.05	0.05	-	0.1	
			Platinum (g/t)	0.95	1.01	0.51	0.50	0.65	0.67	1.3	1.1	
			Palladium (g/t)	1.00	1.10	0.54	0.54	0.69	0.73	1.6	1.3	
Total INO			(Mt)	14.9	15.6	29.9	31.4	44.8	47.0	38	33	
			Nickel (%)	2.58	2.59	2.75	2.76	2.70	2.70	2.1	2.5	
			Copper (%)	1.44	1.56	0.99	1.08	1.14	1.24	2.4	1.8	
			Cobalt (%)	0.06	0.05	0.06	0.06	0.06	0.06	0.1	0.1	
			Platinum (g/t)	0.98	1.00	0.71	0.69	0.80	0.79	1.1	1.0	
			Palladium (g/t)	1.60	1.54	1.33	1.30	1.42	1.37	1.9	1.8	

Integrated Nickel Operations Ore Reserves (Raglan, Sudbury)

Name of	Attributable	Mining		Proved Ore Reserves Probable Ore Reserve			e Reserves	Total Ore Reserves Competent			
operation	Interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person	
Raglan	100%	UG	Ore (Mt)	3.60	3.75	6.43	6.68	10.03	10.43	RC	
			Nickel (%)	3.17	3.01	3.02	3.05	3.07	3.04		
			Copper (%)	0.73	0.73	0.78	0.77	0.76	0.76		
			Cobalt (%)	0.07	0.07	0.06	0.06	0.07	0.06		
			Platinum (g/t)	0.82	0.76	0.88	0.89	0.86	0.84		
			Palladium (g/t)	1.94	1.81	2.05	2.04	2.01	1.96		
Sudbury	100%	UG	Ore (Mt)	6.16	7.11	2.88	3.89	9.04	11.00	SEK	
-			Nickel (%)	1.34	1.39	1.62	1.47	1.43	1.42		
			Copper (%)	1.88	2.06	0.62	0.79	1.47	1.61		
			Cobalt (%)	0.03	0.03	0.05	0.04	0.04	0.03		
			Platinum (g/t)	1.02	1.16	0.16	0.30	0.75	0.85		
			Palladium (g/t)	1.09	1.25	0.18	0.31	0.80	0.92		
Total INO			(Mt)	9.8	10.9	9.3	10.6	19.1	21.4		
			Nickel (%)	2.02	1.95	2.59	2.47	2.29	2.21		
			Copper (%)	1.46	1.60	0.73	0.78	1.10	1.20		
			Cobalt (%)	0.04	0.04	0.06	0.05	0.05	0.04		
			Platinum (g/t)	0.95	1.02	0.66	0.67	0.81	0.85		
			Palladium (g/t)	1.40	1.44	1.47	1.40	1.44	1.43		

Metals and Minerals Nickel

Notes

For the purposes of this statement, the term 'Ore Reserves' as defined by the JORC Code 2012 has the same meaning as 'Mineral Reserves' as defined in the CIM Standards 2014. The resource totals have been restated in compliance with the JORC Code.

There are no known environmental, permitting, legal, taxation, political or other relevant issues that would materially affect the estimates of the Mineral Reserves.

Depending on when production is scheduled, Mineral Reserves and Resources are calculated using a blend of short, medium, or long term metal price assumptions and exchange rates.

- **Raglan:** Ni-Cu-Co-PGE mineralisation is located at or near the base of subvolcanic mafic-ultramafic intrusive complexes referred to as the "Raglan Formation". Economic Ni-Cu-Co-PGE mineralisation is composed of disseminated, nettextured, and massive pyrrhotite-pentlandite-chalcopyrite rich sulphides contained within more than 135 individual sulphide lenses, extending from surface to more than 750m vertical depth. The size of these high-grade sulphide lenses varies significantly from 0.01Mt to 5.2Mt, averaging 0.2Mt. Remaining life of mine is in excess of 20 years. Expiry date of relevant mining leases and exploration licenses: depending on the mine/project, range from 30 July 2017 to 17 August 2026.
- Sudbury: Sulphide deposits sit on broadly defined trends of mineralisation along basal brecciated rocks of the Sudbury Igneous Complex as pentlandite-pyrrhotite-chalcopyrite rich concentrations as well as within the underlying footwall in fractured pathways as chalcopyrite dominated polymetallic (Cu, Ni, Au, Ag, Pt, Pd) vein-style sulphides. The total Mineral Reserve and Resource tonnage decreased from 2015 due to production. The increase in Inferred Resources is based on the discovery of the Norman West footwall deposit. Cut-off grades are calculated for each individual mine site or resource based on a metal equivalent or net smelter return value taking into account all recoverable metals. The current LOM plan based on Mineral Reserves extends to the end of 2021. Significant undeveloped Mineral Resources and exploration potential provide an opportunity to extend mine life well beyond 2021. All Land holdings in Sudbury covering existing Mineral Reserves and Resources are patented and 100% owned by Glencore with the exception of an area covered by two licences of occupation which are held in perpetuity and one mining lease which expires in 2033.

Competent Persons

DP = Daniel Patry, Glencore Nickel, P.Geo (OGQ).

RC = Richard Caumartin, Glencore Nickel, P.Eng. (OIQ).

SEK = Steve Kormos, Glencore Nickel, P. Geo. (APGO).

Metals and Minerals Nickel

Murrin Murrin Mineral Resources

Name of	Attributable	Attributable	Mining		Measured Resou		Indicated Resor		Measu Indicated	red and Resources	Inferred Reso	Mineral urces	Competent
operation	Interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person	
Murrin Murrin	100%	00	(Mt)	186.1	167.3	88.3	79.0	274.4	246.3	21	18	SK	
			Nickel (%)	0.97	1.01	0.98	0.99	0.97	1.00	0.9	0.9		
			Cobalt (%)	0.071	0.071	0.078	0.083	0.073	0.075	0.06	0.07		

Murrin Murrin Ore Reserves

Name of	Attributable	Mining		Proved Ore Reserves		Probable Ore Reserves		Total Ore Reserves		Competent
operation	Interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Murrin Murrin	100%	OC	Ore (Mt)	183.7	151.6	54.2	39.9	237.9	191.6	KG/PW
			Nickel (%)	0.94	0.97	0.92	0.96	0.94	0.97	
			Cobalt (%)	0.064	0.070	0.061	0.069	0.064	0.069	

Koniambo Mineral Resources

Name of	Attributable	Mining		Measured Resou		Indicated Reso		Measur Indicated F		Inferred Reso		Competent
operation	Interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Koniambo	49%	00	(Mt)	13.1	18.1	46.3	29.1	59.6	47.2	88	95	HD
			Nickel (%)	2.50	2.48	2.44	2.42	2.46	2.44	2.5	2.5	

Koniambo Ore Reserves

Name of	Attributable	Minina		Proved Ore Reserves		Probable Ore Reserves		Total Ore Reserves		Competent
operation	Interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Koniambo	49%	00	Ore (Mt)	9.7	14.2	26.2	20.9	35.9	35.1	HD
			Nickel (%)	2.30	2.33	2.28	2.29	2.29	2.30	

Other Nickel Mineral Resources

Name of operation Kabanga	Attributable Interest	e Minina		Measured Resou			d Mineral urces		red and Resources		Mineral urces	Competent
		Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
	50%	UG	(Mt)	13.8	13.8	23.4	23.4	37.2	37.2	21	21	SEK
			Nickel (%)	2.49	2.49	2.72	2.72	2.63	2.63	2.6	2.6	
			Copper (%)	0.34	0.34	0.36	0.36	0.35	0.35	0.3	0.3	
			Cobalt (%)	0.21	0.21	0.19	0.19	0.20	0.20	0.2	0.2	
			Platinum (g/t)	0.16	0.16	0.42	0.42	0.32	0.32	0.3	0.3	
		P	alladium (g/t)	0.19	0.19	0.28	0.28	0.25	0.25	0.3	0.3	

Metals and Minerals Nickel

Notes

For the purposes of this statement, the term 'Ore Reserves' as defined by the JORC Code 2012 has the same meaning as 'Mineral Reserves' as defined in the CIM Standards 2014. The Mineral Resource totals have been restated in compliance with the JORC Code.

There are no known environmental, permitting, legal, taxation, political or other relevant issues that would materially affect the estimates of the Mineral Reserves.

Depending on when production is scheduled, Mineral Reserves and Resources are calculated using a blend of short, medium, or long term metal price assumptions and exchange rates.

Murrin Murrin: Nickel and cobalt mineralisation at Murrin is hosted within a laterite formed from the weathering of ultramafic rocks. The resources are hosted in multiple deposits over three main project areas (North, South and East). Mineral Resource and Ore Reserve figures as at 31 December 2016 are generated by depletion of the resource models by using end-of-period surface surveys as at 30 September 2016, with adjustments applied for October to December production. Resources are determined at a 0.76% Ni cut-off.

The Murrin 2016 Ore Reserve estimate is based on the optimised Base Case pit shells for Measured and Indicated Mineral Resources, and includes scats and stockpiles. Updates to process modelling, 4-yearly shutdown costs and operating costs have been included. The Ore Reserve and Resource tonnages increased in 2016 due to lower operating costs and revised optimised pit shells.

Ore Reserve grades have been subject to the application of grade modifying factors. These have been derived from analysis of the previous two years mine-to-mill grade performance and result in grade modifying factors of 96.7% and 86.9% for nickel and cobalt respectively.

At the forecast throughput capacity of 4.5 million tonnes per annum, the project's operating life is in excess of 30 years. Expiry dates for relevant tenements differ for each tenement and range from 2017 to 2037.

Koniambo: Nickel rich laterite deposits are developed on variably serpentinized ultramafic rocks. The Ore Reserve estimate for the deposit was established from a new Life of Mine Resource Model established in 2015 and updated in 2016. It has been adjusted to incorporate changes in classification, and actual completed production up to 31 October 2016, with adjustments applied for forecast production between November and December. Mineral Resources for the Life of Mine area have been calculated by 3D modelling and Conditional Simulation geostatistical methods and for those areas outside of this footprint by the plan polygonal method. The reserve cut-off grade used is 2.0% Ni. Expected mine life is 18 years. The expiry date of relevant mining property licences range from April 2017 to 31 December 2048. Production began in April 2013 and mining is supporting a one line operation at the Metallurgical Plant as at 31 December 2016. Ore Reserves stated include an estimated 256kt at 2.24% of stockpiles between the mine and process plant. Ore Reserves and Mineral Resources exclude an additional 34 million tonnes of medium grade material at 1.9% Ni below the current cutoff grade (< 2.0% Ni).

Kabanga: The current delineated Mineral Resource is comprised of 88% of contact-style massive sulphide and 12% of ultramafic-hosted disseminated to semi-massive sulphide mineralisation. The ultramafic bodies are hosted in a sequence of metamorphosed pelitic sediments that are overturned, steeply dipping. All resource estimates are done using Ordinary Kriging and are based on block models with appropriate variography. A cut-off grade of 1% Niequivalent is used for all Mineral Resources except ultramafic-hosted disseminated to semi-massive sulphide (UMIN) mineralisation at Tembo (1.1% Ni cut-off grade applied). The contribution to the Ni-equivalent value is provided by copper, cobalt and platinum group elements. The last Mineral Resource drilling campaign was done in 2009 and the latest Mineral Resource estimate dates from June 2010. The delineated Kabanga Mineral Resources are sufficient to support a 30 year mine life as currently studied. All resources are held under a Retention Licence which was renewed for an additional 5 year period on 2 May 2014.

Competent Persons

HD = Hubert Dumon, Koniambo Nickel SAS, (AusIMM).

- KG = Kellie Gill, Minara Resources Pty Ltd., (mining and metallurgical costs, reserve optimisation), (AusIMM).
- PW = Paul Wiltshire, Minara Resources Pty Ltd., (process plant modelling assumptions), (AusIMM).
- SK = Stephen King, Minara Resources Pty Ltd., (geostatistical analysis, modelling/estimation and resource classification), (AusIMM).
- SEK = Steve Kormos, Glencore Nickel, P. Geo (APGO).

Metals and Minerals Ferroalloys

Chrome Mineral Resources

	Attributable	Mining		Measured Resou		Indicated Resou		Measure Indicated R		Inferred Resou		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Waterval Mine	79.5%	UG	Ore (Mt)	16.199	16.331	0.98	0.98	17.18	17.31	0.6	0.6	DR/MM
			$Cr_2O_3(\%)$	41.3	41.3	42.6	42.6	41.4	41.4	43	43	
Marikana West	79.5%	UG	Ore (Mt)	2.807	2.909	1.60	1.60	4.41	4.50	_	-	DR/MM
			Cr ₂ O ₃ (%)	42.46	42.41	42.6	42.6	42.5	42.5	_	_	
Kroondal Mine	79.5%	UG/OC	Ore (Mt)	8.521	7.750	2.15	2.16	10.67	9.91	_	_	DR/MM
			Cr ₂ O ₃ (%)	42.84	42.75	41.6	41.5	42.6	42.5	_	_	
Kroondal	79.5%	UG/OC	Ore (Mt)	10.566	10.799	5.03	4.98	15.59	15.78	_	_	DR/MM
Gemini			Cr ₂ O ₃ (%)	42.58	42.47	41.6	41.6	42.3	42.2	_	-	
Marikana East	79.5%	UG	Ore (Mt)	4.196	4.178	0.78	0.75	4.98	4.92	_	_	DR/MM
			Cr ₂ O ₃ (%)	42.25	42.25	41.8	41.8	42.2	42.2	_	_	
Boshoek Mine	79.5%	UG/OC	Ore (Mt)	_	_	17.09	17.09	17.09	17.09	_	_	DR/MM
			Cr ₂ O ₃ (%)	_	_	40.5	40.5	40.5	40.5	_	_	
Thorncliffe Mine	79.5%	UG/OC	Ore (Mt)	36.104	38.906	13.05	13.75	49.16	52.65	47.4	47.3	DR/SV
			Cr ₂ O ₃ (%)	40.63	40.61	41.3	41.0	40.8	40.7	38	38	
Helena Mine	79.5%	UG/OC	Ore (Mt)	23.621	23.786	11.42	11.17	35.04	34.96	134.4	133.5	DR/SV
			Cr ₂ O ₃ (%)	39.88	39.75	39.0	38.2	39.6	39.2	38	38	
Extension 9	79.5%	UG	Ore (Mt)	-	_	12.94	13.85	12.94	13.85	-	-	DR/MM
			Cr_2O_3 (%)	-	_	41.4	41.4	41.4	41.4	-	-	
De Grooteboom	79.5%	UG/OC	Ore (Mt)	1.039	0.812	0.51	0.65	1.54	1.46	_	_	DR/SV
			Cr_2O_3 (%)	40.23	40.28	40.3	40.3	40.3	40.3	-	-	
Klipfontein/	79.5%	UG	Ore (Mt)	11.820	12.433	16.95	15.79	28.77	28.23	114.9	116.0	DR/MM
Waterval			Cr ₂ O ₃ (%)	42.12	42.11	42.0	42.0	42.0	42.0	42	42	
Total Chrome			(Mt)	115	118	83	83	197	201	297	297	
			Cr ₂ O ₃ (%)	41.2	41.1	41.0	40.9	41.1	41.0	40	40	

Chrome Ore Reserves

	Attributable	Mining		Proved Ore	Reserves	Probable Or	e Reserves	Total Ore	Reserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Waterval Mine	79.5%	UG	Ore (Mt)	8.727	10.534	0.89	0.91	9.61	11.45	DR/MM
			Cr ₂ O ₃ (%)	31.40	30.94	26.8	26.0	31.0	30.5	
Marikana West	79.5%	UG	Ore (Mt)	0.110	0.584	_	-	0.11	0.58	DR/MM
			Cr ₂ O ₃ (%)	28.93	28.92	-	-	28.8	28.9	
Kroondal Mine	79.5%	UG/OC	Ore (Mt)	2.142	1.895	1.92	1.94	4.07	3.83	DR/MM
			Cr ₂ O ₃ (%)	29.27	28.44	28.5	28.1	28.9	28.3	
Kroondal Gemini	79.5%	UG/OC	Ore (Mt)	6.178	6.942	4.72	4.68	10.89	11.63	DR/MM
			Cr ₂ O ₃ (%)	31.42	31.05	28.5	28.3	30.2	29.9	
Marikana East	79.5%	UG	Ore (Mt)	-	0.306	-	0.20	_	0.51	DR/MM
			Cr ₂ O ₃ (%)	-	29.16	-	28.1	-	28.7	
Boshoek Mine	79.5%	UG/OC	Ore (Mt)	-	-	0.58	0.58	0.58	0.58	DR/MM
			Cr ₂ O ₃ (%)	-	-	26.1	26.1	26.1	26.1	
Thorncliffe Mine	79.5%	UG/OC	Ore (Mt)	22.267	22.413	8.15	7.80	30.41	30.21	DR/SV
			Cr ₂ O ₃ (%)	33.76	36.94	34.2	37.9	33.9	37.2	
Helena Mine	79.5%	UG/OC	Ore (Mt)	3.250	3.640	-	0.18	3.25	3.82	DR/SV
			Cr ₂ O ₃ (%)	33.45	34.73	-	30.6	33.5	34.5	
Klipfontein/	79.5%	UG	Ore (Mt)	-	0.232	0.29	0.29	0.28	0.53	DR/MM
Waterval			Cr ₂ O ₃ (%)	-	25.94	27.3	27.3	28.0	26.7	
Total Chrome			(Mt)	43	47	17	17	59	63	
			Cr ₂ O ₃ (%)	32.7	34.0	31.1	32.6	32.2	33.6	

Notes

Tonnages are quoted as million metric tonnes.

Grades are quoted as %Cr₂O₃.

The Measured and Indicated Mineral Resources are inclusive of those Mineral Resources modified to produce Ore Reserves.

The chromitite assets include those owned by Glencore and Merafe in different ownership percentages, the attributable interest in such assets remain as reflected.

All Glencore Ferroalloys' chrome operations mine the chromitite deposits developed within the world renowned Bushveld Complex of South Africa. The 2060 Ma year-old Bushveld Complex is the largest known deposit of chrome, vanadium and platinum group elements (PGEs) in the world. The Bushveld Complex stretches 350km east-west and 450km north-south. The chrome ore is mined from shallow dipping $(10^{\circ} - 14^{\circ})$ tabular orebodies. Although there are numerous chromitite layers developed in the Bushveld Complex, the chromitite layers targeted for economic exploitation are the LG6/LG6A Chromitite Layer package and the MG1 Chromitite Layer.

No cut-off grades are applied to the chromitite layers currently being mined. The chromitite layer grades show exceptional regional grade consistency and continuity.

The chromitite layers are mostly mined underground using trackless mechanised mining methods on a board and pillar mine layout design.

The Mineral Resources are estimated as chromitite tonnages and grades to reflect the grades of the various individual chromitite layers. Both the LG6 and MG1 Chromitite Layers which Glencore currently mine are discrete solid chromitite layers with sharp contacts.

Changes in the year on year Mineral Resource tonnage and grade estimates are mainly due to mining depletion and changes due to additional geological information gained through exploration. These changes reflect in the tonnage and grade reports from the grade block models. The most significant changes are noted below.

- **Kroondal Mine:** Previously considered un-mineable areas (white areas) have been added back to the resources. These resources amount to 0.86Mt. The net change in the Kroondal Mine resources is 1.23Mt after mining depletion has been discounted.
- **Kroondal Gemini:** Previously considered un-mineable areas (white areas) have been added back to the resources. These resources amount to 0.25Mt. The net change in the Kroondal Gemini resources is 0.77Mt after mining depletion has been discounted.

The tonnage and grade estimations are initiated by the geostatistical analysis of the exploration drill hole data. The outcomes of this analysis are used in the construction of block models for each and every mine and project area. The geostatistical analysis of the chromitite data indicates a high degree of continuity both in grade and thickness of the chromitite layers. The block model estimates are verified using geostatistical parameters such as Kriging Efficiency to test the stability of the variograms used and the suitability of the selected cell sizes and Kriging parameters. Tonnages and grades are reported from these block models for each mine and project. There is a high degree of confidence in the tonnage and grade estimations derived from the block models. This is confirmed by the monthly and yearly reconciliation between the block model estimates, the monthly survey measurements and the actual mine production for each operating mine.

The LOM for the operating chrome mines vary between 1 year and 5 years based on the declared Ore Reserves. The LOM periods for the various operating mines, based on all the Mineral Resources converted to Ore Reserves vary between 5 years and 38 years. The Mining Right expiry dates vary from 2022 to 2039 for the operating chrome mines. All the chrome mining rights were granted for an initial period of 30 years. The production rates for the various chrome mines vary from 37kt ROM per month to 110kt ROM per month.

The Mining Rights of Horizon and Wonderkop have been disposed of through a Section 11 process and will no longer be reflected in the resource and reserve statement. The combined resources of these two properties amounted to 21.703Mt. The declared reserves for these properties were 2.102Mt. The Klipfontein/Waterval Prospecting Right has successfully been converted to a Mining Right and incorporated into the Kroondal Mining Right. The Klipfontein/Waterval resources and reserves are still declared separately.

A Prospecting Right has been granted to Glencore on the farms of St George and Richmond. These properties form the downdip extension of the Thorncliffe and Helena chrome Mining Right areas. The next resource and reserve statement will include an initial resource statement on these properties. The Extension 9 Prospecting Right is in its final 3 year cycle.

- PJG = Pieter-Jan Gräbe, Glencore Ferroalloys, (SACNASP); overall responsibility for Mineral Resources and Ore Reserves.
- SV = Solly Vaid, Glencore Ferroalloys, (PLATO); responsibility for Mineral Resources and Ore Reserves.
- DR = Dean Richards, Obsidian Consulting Services (SACNASP); responsible for geostatistical analysis of data, Mineral Resource classification and construction of tonnage and grade block models and reporting of tonnage and grades from block models.
- MM = Mogomotsi Maputle, Glencore Ferroalloys, (SACNASP), Responsible for Mineral Resources and Ore Reserves.

Vanadium Mineral Resources

		Mining		Measured Mineral Resources		Indicated Mineral Resources		Measured and Indicated Resources		Inferred Mineral Resources		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Rhovan	74.0%	00	(Mt)	49.18	48.85	39.4	38.7	88.5	87.6	95	95	DR/SM
			V ₂ O ₅ (%)	0.48	0.48	0.5	0.5	0.5	0.5	0.5	0.5	

Vanadium Ore Reserves

	Attributable	Mining		Proved Ore	Reserves	Probable Ore	Reserves	Total Ore F	Reserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Rhovan	74.0%	00	Ore (Mt)	26.83	26.66	13.0	12.9	39.8	39.6	DR/SM
			V ₂ O ₅ (%)	0.47	0.48	0.5	0.5	0.5	0.5	

Notes

Tonnages are quoted as million metric tonnes.

Grades are quoted as %V₂O₅.

The Measured and Indicated Mineral Resources are inclusive of those Mineral Resources modified to produce Ore Reserves.

Glencore Ferroalloys' vanadium mining operations mine the vanadiferous magnetite deposits developed within the Bushveld Complex, South Africa. The 2060 Ma year-old Bushveld Complex is the largest known deposit of chrome, vanadium and platinum group elements (PGE's) in the world.

The magnetite ore is mined from shallow dipping ($6^{\circ} - 25^{\circ}$) stratified magnetite orebodies. Various ore zones with varying grades can be identified in the orebody. The ore zones are defined based on their magnetite and vanadium content.

The magnetite ore is mined using open cast mining methods.

No material changes were recorded compared with the 2015 resource and reserve estimation.

Rhovan: There was a net increase of 1.66Mt in the Mineral Resource estimate after mining depletions have been discounted.

The Ore Reserves had a net increase of 1.14Mt after mining depletions have been discounted.

Obsidian Consulting Services has estimated the Mineral Resources for Pit 1 and Pit 6. An updated grade block model with new data was constructed during September 2016, following the addition of blast hole data for the 2015-2016 reporting period.

The tonnage and grade estimations were done using geostatistical analysis of the exploration drill hole and blast hole data. From this analysis the most appropriate parameters for the construction of block models for the various pits were derived. The block model estimates are verified using geostatistical parameters such as Kriging Efficiency and Slope of Regression to test the stability of the variograms used and the suitability of the selected cell sizes. A final geospatial validation is done by means of swath and QQ plots. Other validations included a comparison of distributions of the source data versus estimated results as well as swath plots. Tonnages and grades are reported from the block models for each pit. The degree of confidence in the tonnage and grade estimations derived from the block models is reflected in the classified Mineral Resource classes.

The Rhovan LOM based on the declared Ore Reserves is 16 years. The LOM based on all the Mineral Resources converted to Ore Reserves is 33 years. Rhovan is mining from various open cast pits at a mining rate averaging 200kt of ROM per month. The Mining Right expiry date is 2027.

- PJG = Pieter-Jan Gräbe, Glencore Ferroalloys, (SACNASP); overall responsibility for Mineral Resources and Ore Reserves.
- DR = Dean Richards, Obsidian Consulting Services, (SACNASP); responsible for data validation, geo-statistical analysis of data, construction of tonnage and grade block models and reporting of tonnage and grades from block models for Mineral Resource and Ore Reserve estimates.
- SM = Sydney Maseti, Glencore Ferroalloys, (SACNASP); Responsible for Mineral Resources and Ore Reserves.

PGM Mineral Resources

	Attributable	Mining		Measured Resou			d Mineral ources		red and Resources		Mineral urces	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Mototolo JV	37%	UG/OC	(Mt)	24.294	29.191	-	-	24.29	29.19	-	-	CL/DR/FF
		3PGE	+ Gold (g/t)	3.89	4.15	-	-	3.9	4.2	-	-	
Eland Platinum	73.99%	UG/OC	(Mt)	55.957	55.957	16.36	16.36	72.32	72.32	76.3	76.3	DR
		3PGE	+ Gold (g/t)	4.49	4.49	3.5	3.5	4.3	4.3	4.5	4.5	
Zilkaatsnek	73.99%	UG/OC	(Mt)	_	-	2.78	2.78	2.78	2.78	-	_	DR
		3PGE	+ Gold (g/t)	-	-	2.5	2.5	2.5	2.5	-	-	
Schietfontein	73.99%	UG/OC	(Mt)	_	_	6.41	6.41	6.41	6.41	7.0	7.0	DR
		3PGE	+ Gold (g/t)	-	-	2.3	2.3	2.3	2.3	2.3	2.3	
Total PGM			(Mt)	80	85	26	26	106	111	83	83	
		3PGE	+ Gold (g/t)	4.31	4.37	3.1	3.1	4.0	4.1	4.3	4.3	

PGM Ore Reserves

	Attributable	Minina		Proved Ore	Reserves	Probable Ore	Reserves	Total Ore R	leserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Mototolo JV	37%	UG/OC	Ore (Mt)	12.795	18.731	-	-	12.80	18.73	FF/DR
			3PGE + Gold (g/t)	3.58	3.82	-	-	3.6	3.8	

Notes

Tonnages are quoted as million metric tonnes.

Grades are quoted as 3PGE + Au (Platinum, Palladium, Rhodium and Gold).

The Measured and Indicated Mineral Resources are inclusive of those Mineral Resources modified to produce Ore Reserves.

Glencore Ferroalloys' platinum mining operations mine the platinum bearing UG2 Chromitite Layer of the Bushveld Complex, South Africa. The 2060 Ma year-old Bushveld Complex is the largest known deposit of chrome, vanadium and platinum group elements (PGEs), in the world.

The PGE ore at Mototolo Mine is mined from a shallow dipping $(10^{\circ} - 14^{\circ})$ tabular ore-body referred to as the UG2 Chromitite Layer. The chromitite layer is mined underground using a trackless mechanised mining method on a bord and pillar mine layout design.

The Eland Platinum Mine has been put on care and maintenance. No Ore Reserves have been declared.

Changes in the year on year tonnage and grade estimates are mainly due to mining depletion, reclassification and changes in the Mineral Resource and Ore Reserve tonnages and grades due to additional geological information gained through exploration.

The tonnage and grade estimations are made using geostatistical analysis of the exploration drill hole data as well as underground channel sample data, where available. From this analysis the most appropriate parameters are derived for the construction of block models for the various orebodies. Tonnages and grades are reported from these block models for each mine and project. Confidence limit curves are derived from fitted distributions and used to classify the Mineral Resources at various confidence levels. A final geospatial validation is done by means of swath and QQ plots.

Mototolo JV: The structural and grade block model was updated with exploration drill hole data drilled during the 2015 – 2016 reporting period.

The net change in the year on year Mineral Resources is -2.0Mt. This change is mainly due to the removal of resources that are considered unmineable, represented by slump and pothole areas as well as the St George Fault

Zone. The change is after the mining depletions have been discounted.

The Ore Reserve changed year on year on a net basis, after mining depletions have been discounted, by -3.145Mt. The change was mainly due to a decrease in the mining footprint that was reduced from 7.9 years back to an approximate 5 year footprint.

The LOM period for Mototolo Platinum Mine is 5.6 years based on the declared Ore Reserves. The LOM period based on all the Mineral Resources converted to Ore Reserves is 9.6 years. This includes remnant areas. The Mining Right expiry date is 2039. The planned production rate is 191kt ROM per month.

Eland, Zilkaatsnek and Schietfontein: Eland Platinum Mine is currently on care and maintenance and there is therefore no change in the year-on-year resource estimate.

Eland, Zilkaatsnek and Schietfontein are contiguous Mining and Prospecting Right areas, targeting the same orebody and together constitute the Eland Platinum Mine.

The current Mineral Resource estimate contained in this report is based only on mining depletions during 2015. No exploration activities were carried out during the reporting period.

The Mineral Resources are constrained by lithological facies. The Mining Right expiry date is 2039.

- PJG = Pieter-Jan Gräbe, Glencore Ferroalloys (SACNASP); overall responsibility for Mineral Resources.
- FF = Frikkie Fensham, Glencore Ferroalloys, (SACNASP); Responsible for Mineral Resources and Ore Reserves.
- DR = Dean Richards, Obsidian Consulting Services (SACNASP); responsible for data validation, construction of tonnage and grade block models and reporting of tonnage and grades from block model.
- CL = Carina Lemmer, Geological & Geostatistical Services (SACNASP); responsible for geostatistical analysis of data and classification of Mineral Resources.

Silica Mineral Resources

	Attributable	Mining		Measure Reso	d Mineral urces	Indicated Reso		Measu Indicated	red and Resources	Inferred Reso		Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Rietvly	79.5%	00	(Mt)	-	-	23.27	23.81	23.27	23.81	-	_	СМ
			SiO ₂ (%)	-	-	91	91	91	91	_	_	

Silica Ore Reserves

	Attributable	Minina		Proved Ore F	Reserves	Probable Ore	Reserves	Total Ore Re	eserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Rietvly	79.5%	OC	Ore (Mt)	-	-	2.05	2.15	2.05	2.15	СМ
			SiO ₂ (%)	_	_	91	91	91	91	

Notes

Tonnages are quoted as thousand metric tonnes.

Grades are quoted as %SiO₂.

Glencore Ferroalloys' silica mining operation is an open cast mining operation mining a massive quartzite deposit of the Magaliesberg Formation of the Pretoria Group, which is a subgroup of the Transvaal Super Group, South Africa. The Magaliesberg Formation forms a prominent quartzite ridge striking north-south along the Magaliesberg Mountain Range. The formation is a few hundred meters thick and dips towards the east at over 20°. The Rietvly quartzite orebody is a very pure quartzite with a silica content of over 90% SiO₂.

The ore is mined through open cast mining methods and the ROM ore is crushed, washed and sized on site to produce a final sized and quality graded product. No silica cut-off grades are applied to the Mineral Resource estimation. The quartzite is mainly mined to supply the Glencore Ferroalloys furnaces with silica.

No significant changes have been recorded in the year on year Mineral Resource estimation.

No significant changes have been recorded in the year on year Ore Reserve estimation.

The Ore Reserves for Rietvly Silica Mine is based on a 5 year mining footprint with a production rate of 39.94kt ROM per month. The LOM period based on all the Mineral Resources converted to Ore Reserves is 38.3 years. The Mining Right expiry date is 2037.

Competent Persons

- PJG = Pieter-Jan Gräbe, Glencore Ferroalloys (SACNASP); overall responsibility for Mineral Resources and Ore Reserve.
- CM = Mogomotsi Maputle, Glencore Ferroalloys (SACNASP); responsible for Mineral Resources and Ore Reserves.

Competent Person for Ore Reserve / Competent Person for Mineral Resource; where only one set of initials is listed, the same Competent Person is responsible for all categories quoted. Unless otherwise noted all Competent Persons are full time employees of Glencore plc subsidiaries.

Metals and Minerals Iron Ore

Iron Ore Mineral Resources

	ral Resol			Measu		Indica Mineral D		Measure Indicated R		Inferr Mineral De		
Name of operation	Attributable interest	Minin		Mineral Re 31.12.16		Mineral Re 31.12.16		Indicated R 31.12.16		Mineral Re 31.12.16		Competent Person
El Aouj Mining			,									
Guelb el Aouj	50%	OC	(Mt)	400	400	1,170	1,170	1,570	1,570	300	300	AM/SvdM
East			Iron (%)	36	36	36	36	36	36	36	36	
			DTC wt (%)	45	45	45	45	45	45	45	45	
			DTC Iron (%)	69.8	69.8	69.2	69.2	69.3	69.3	69.5	69.5	
			Oxidised (Mt)	70	70	80	80	150	150	30	30	AM/SvdM
			Iron (%)	34	34	35	35	35	35	35	35	
Guelb el Aouj	50%		(Mt)	_	_	185	185	185	185	615	615	AM/SvdM
Centre			Iron (%)	_	-	34	34	34	34	35	35	
			DTC wt (%)	_	-	43	43	43	43	44	44	
			DTC Iron (%)	-	-	69.6	69.6	69.6	69.6	69.8	69.8	
			Oxidised (Mt)	_	_	_	_	_	_	45	45	AM/SvdN
			Iron (%)	_	_	_	_	_	_	33	33	
Bou Derga	50%		(Mt)	_	_	_	_	_	_	510	510	AM/SvdN
			Iron (%)	_	_	_	_	_	_	36	36	
			DTC wt (%)	-	-	-	-	-	-	43	43	
			DTC Iron (%)	-	-	-	-	-	-	69.7	69.7	
			Oxidised (Mt)	-	-	-	-	-	-	130	130	AM/SvdN
			Iron (%)	-	-	-	-	-	-	35	35	
Tintekrate	50%		(Mt)	_	-	_	-	_	_	710	710	AM/SvdN
			Iron (%)	-	-	-	-	-	-	36	36	
			DTC wt (%)	-	-	-	-	-	-	44	44	
			DTC Iron (%)	-	-	-	-	-	-	69.4	69.4	
			Oxidised (Mt)	-	-	-	-	-	-	180	180	AM/SvdN
			Iron (%)	-	-	-	-	-	-	34	34	
Total El Aouj M	ining Com	pany	S.A. (Mt)	470	470	1,435	1,435	1,905	1,905	2,520	2,520	
			Iron (%)	36	36	36	36	36	36	35	35	
Sphere Maurita	nia S.A.											
Askaf North	90%	OC	(Mt)	200	200	160	160	360	360	45	15	AM/SvdM
	0070	00	(IVIL)	200	200	100		500	500	-10	40	Alvi/Svulv
	0070	00	Iron (%)	36	36	35	35	36	36	36	36	AW
	0070	00	. ,									Alli/Ovalv
	0070	00	Iron (%)	36	36	35	35	36	36	36	36	AW/Svulv
		00	Iron (%) DTC wt (%)	36 47	36 47	35 45	35 45	36 46	36 46	36 45	36 45 69.2	
		00	Iron (%) DTC wt (%) DTC Iron (%)	36 47 69.8	36 47 69.8	35 45 69.4	35 45 69.4	36 46 69.6	36 46 69.6	36 45 69.2	36 45 69.2	
Askaf Centre	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt)	36 47 69.8 15	36 47 69.8 15	35 45 69.4 30	35 45 69.4 30	36 46 69.6 45	36 46 69.6 45	36 45 69.2 15	36 45 69.2 15 35	AM/SvdN
Askaf Centre			Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%)	36 47 69.8 15 35	36 47 69.8 15 35	35 45 69.4 30 35	35 45 69.4 30 35	36 46 69.6 45 35	36 46 69.6 45 35	36 45 69.2 15 35	36 45 69.2 15 35	AM/SvdN
Askaf Centre			Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) (Mt)	36 47 69.8 15 35 -	36 47 69.8 15 35 –	35 45 69.4 30 35 -	35 45 69.4 30 35 -	36 46 69.6 45 35 -	36 46 69.6 45 35 -	36 45 69.2 15 35 95	36 45 69.2 15 35 95	AM/SvdN
Askaf Centre			Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%)	36 47 69.8 15 35 –	36 47 69.8 15 35 -	35 45 69.4 30 35 -	35 45 69.4 30 35 -	36 46 69.6 45 35 -	36 46 69.6 45 35 -	36 45 69.2 15 35 95 36	36 45 69.2 15 35 95 36 42 69.9	AM/SvdM AM/SvdM
Askaf Centre			Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) DTC wt (%)	36 47 69.8 15 35 – –	36 47 69.8 15 35 - - -	35 45 69.4 30 35 - -	35 45 69.4 30 35 - -	36 46 69.6 45 35 – –	36 46 69.6 45 35 - -	36 45 69.2 15 35 95 36 42	36 45 69.2 15 35 95 36 42 69.9	AM/SvdM AM/SvdM
Askaf Centre		OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) DTC wt (%) DTC Iron (%)	36 47 69.8 15 35 - - - -	36 47 69.8 15 35 - - - - -	35 45 69.4 30 35 - -	35 45 69.4 30 35 - - - -	36 46 69.6 45 35 - - - -	36 46 69.6 45 35 - - - -	36 45 69.2 15 35 95 36 42 69.9	36 45 69.2 15 35 95 36 42 69.9 13 37	AM/SvdM AM/SvdM AM/SvdM
Askaf Centre Askaf East			Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) (Mt)	36 47 69.8 15 35 - - - - - -	36 47 69.8 15 35 - - - - - -	35 45 69.4 30 35 - - - - -	35 45 69.4 30 35 - - - - - -	36 46 69.6 45 35 - - - - -	36 46 69.6 45 35 - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70	36 45 69.2 15 35 95 36 42 69.9 13 37 70	AM/SvdM AM/SvdM AM/SvdM
	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) 0TC wt (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%)	36 47 69.8 15 35 - - - - - - - - -	36 47 69.8 15 35 - - - - - - - -	35 45 69.4 30 35 - - - - -	35 45 69.4 30 35 - - - - - - - -	36 46 69.6 45 35 - - - - - - - -	36 46 69.6 45 35 - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35	AM/SvdM AM/SvdM AM/SvdM
	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) (Mt)	36 47 69.8 15 35 - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70	36 45 69.2 15 35 95 36 42 69.9 13 37 70	AM/SvdM AM/SvdM AM/SvdM
	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) 0TC wt (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35	AM/SvdM AM/SvdM AM/SvdM
	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) OTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) Oxidised (Mt) Iron (%) Other (%) DTC wt (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3	AM/SvdM AM/SvdM AM/SvdM
	90%	OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) OTC Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) DTC Iron (%) DTC Iron (%) DTC Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31	AM/SvdM AM/SvdM AM/SvdM
Askaf East	90%	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) (Mt)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - 215	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251	AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M	90% 90% auritania S	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) OTC Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) DTC Iron (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) DTC Iron (%) Iron (%) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31	AM/SvdM AM/SvdM AM/SvdM
Askaf East	90% 90% auritania S	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) (Mt)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - 215	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei	90% 90% auritania S	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) (Mt) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - - - - - - - 215	35 45 69.4 30 35 - - - - - - - - - - - - - - - 190 35 2,180	35 45 69.4 30 35 - - - - - - - - - - - - - - - - 190 35 2,180	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - 405 36 2,180	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei Lebtheinia	90% 90% auritania S nia S.A.	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) Iron (%) Iron (%) Iron (%) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - 215 36	36 47 69.8 15 35 - - - - - - - - - - - 2 5 36	35 45 69.4 30 	35 45 69.4 30 35 - - - - - - - - - - - - 190 35	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - 405 36	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 350 32	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei Lebtheinia	90% 90% auritania S nia S.A.	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) (Mt) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - 215 36	36 47 69.8 15 35 - - - - - - - - - - - - 215 36	35 45 69.4 30 35 - - - - - - - - - - - - - - - 190 35 2,180	35 45 69.4 30 35 - - - - - - - - - - - - - - - 190 35 2,180	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - 405 36 2,180	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei Lebtheinia	90% 90% auritania S nia S.A.	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) Iron (%) Iron (%) Iron (%) Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - 215 36	36 47 69.8 15 35 - - - - - - - - - - - 215 36 -	35 45 69.4 30 	35 45 69.4 30 190 35 2,180 32	36 46 69.6 45 405 36 2,180 32	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35 350 32	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 350 32	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei	90% 90% auritania S nia S.A.	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%) DTC Iron (%) DTC Iron (%) DTC Iron (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) Iron (%) Iron (%) DTC wt (%) DTC wt (%)	36 47 69.8 15 35 - - - - - - - - - - - - 215 36	36 47 69.8 15 35 - - - - - - - - - 215 36 - - - 215	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - 405 36 2,180 32 27	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35 350 32 27	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 350 350 32 27 68.1	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei Lebtheinia	90% 90% auritania S nia S.A.	000	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Iron (%) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - 215 36 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - 215 36 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35 350 32 27 68.1	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 350 350 32 27 68.1	AM/SvdM AM/SvdM AM/SvdM AM/SvdM
Askaf East Total Sphere M Sphere Lebthei Lebtheinia	90% 90% auritania S inia S.A. 100%	OC OC	Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC Iron (%) Oxidised (Mt) Iron (%) Oxidised (Mt) Iron (%) DTC wt (%) DTC wt (%) DTC Iron (%)	36 47 69.8 15 35 - - - - - - - - - - - - - - 215 36 - - - - - - - - - - - - - - - - - -	36 47 69.8 15 35 - - - - - - - - - - - 215 36 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	35 45 69.4 30 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 46 69.6 45 35 - - - - - - - - - - - - - - - - - -	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 35 350 32 27 68.1 210	36 45 69.2 15 35 95 36 42 69.9 13 37 70 35 42 70.3 13 31 251 350 350 32 27 68.1 210	AM/SvdM AM/SvdM AM/SvdM AM/SvdM

Metals and Minerals Iron Ore

	Attributable		Mining			sured Resources		cated Resources		red and Resources		erred Resources	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person	
Jumelles Limite	ed												
Zanaga	50%	OC	(Mt)	2,300	2,300	2,500	2,500	4,800	4,800	2,100	2,100	МТ	
			Iron (%)	34	34	30	30	32	32	31	31		

Iron Ore Reserves

	Attributable	Mining		Proved Ore	Reserves	Probable Ore	Reserves	Total Ore	Reserves	Competent
Name of operation	interest	method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
El Aouj Mining Compa	ny S.A.									
Guelb el Aouj East	50%	OC	Ore (Mt)	380	380	551	551	931	931	JS
			Iron (%)	35	35	35	35	35	35	
			DTC wt (%)	44	44	43	43	44	44	
			DTC Iron (%)	69.6	69.6	69.0	69.0	69.2	69.2	
Jumelles Limited	50%									
Zanaga		OC	Ore (Mt)	770	770	1,290	1,290	2,070	2,070	GB
			Iron (%)	37	37	32	32	34	34	

Notes

All Mineral Resources are considered suitable for open cut extraction.

DTC wt (%) - Davis Tube Concentrate mass recovery.

DTC Iron (%) - Davis Tube Concentrate assay %Fe.

Davis Tube test work has been conducted at a grind size of 95% passing 80 micron.

The rounding used for the values in this report reflects the confidence in the different levels of Mineral Resource and Ore Reserve classifications.

During the year, Glencore acquired the remaining shares in Sphere Minerals that it did not already own.

No exploration activities were carried out in the reporting period.

- **EI Aouj Mining Company:** Glencore holds a 50% interest in the EI Aouj Mining Company through a Joint Venture arrangement with SNIM.
- **Guelb el Aouj East:** The "Guelb" deposits are hosted in Banded Iron Formations (BIF) within the Dorsale Reguibat, an uplifted part of the Archaean West African Craton, which dominates the northern third of Mauritania's surface geology. Recrystallisation and aggregation of the magnetite grains in BIF has resulted in the partial to total destruction of the original banded (bedding) texture to produce the Guelb el Aouj magnetite-quartzite deposits. The geological sequence is overprinted by a reasonably uniform, approximately 80m thick weathered zone in which much of the magnetite has oxidised to hematite.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

The Guelb el Aouj East Ore Reserve Statement is based on a Feasibility Study and uses a 20% DTC wt% cut-off. No oxidised material has been included in the Ore Reserves.

Guelb el Aouj Centre: The El Aouj Centre magnetite-quartzite (MQ) deposit is a highly metamorphosed banded iron formation (meta-BIF) unit that ranges in true thickness from 50m to over 200m. The geometry of the deposit is defined by a tight synformal structure with a sub-vertical axial plane. The synform outcrops over a strike length of about 2.4km. The thickest accumulation of magnetite-quartzite is found along the western limb of the synform, pinching out towards the east. A series of stacked recumbent isoclinal folds probably controlled the overall geometry of the deposit. The original bedding has been partially to completely obliterated by recrystallisation, resulting in a coarse-grained texture with aggregated magnetite grains. The weathered zone, though variable, has an average vertical thickness of approximately 40m. In this zone partial to complete oxidation of magnetite to hematite has occurred.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

Bou Derga: The Bou Derga deposit forms part of a larger scale synformal structure defined by an Archean magnetitequartzite (MQ) unit that ranges in true thickness from approximately 20m to 200m. The thicker parts of the deposit are considered to be a result of isoclinal folding. Drilling was restricted to the western fold closure. The deposit dips towards the northeast at about 60°. The deposit contains a number of internal waste bands (typically 5m to 50m thick) which have been modelled separately and excluded from the Mineral Resource estimation. A northwest-southeast trending fault displaces the mineralisation in the south-eastern part of the deposit.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

The Bou Derga Mineral Resource Statement has been prepared in accordance with the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (the JORC Code, 2004 Edition).

Tintekrate: The Tintekrate deposit is hosted within the Dorsale Reguibat, an uplifted part of the Archaean West African Craton, which dominates the northern third of Mauritania's surface geology. Recrystallisation and aggregation of the magnetite grains in the meta-banded iron formation (BIF) units has resulted in partial to total destruction of the original banded (bedding) texture to produce the Tintekrate and other similar magnetite-quartzite deposits. The Tintekrate deposit is a circular structure defined by a steep dipping MQ unit with dips of 50° to 80° (locally overturned) with true mineralised thicknesses of 100m to 150m on the western side of the structure to 50m to 100m on the eastern side. The weathered zone averages 70m to 75m vertical depth below natural surface and its base tends to mirror the

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natural surface profile. In this zone, magnetite has been partially to completely oxidised to hematite.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

Askaf North: Askaf North Deposit is an east-west striking synformal structure defined by a magnetite-quartzite (MQ) unit that ranges in true thickness from approximately 140m in the western hinge zone to approximately 30m along the eastern part of the southern limb. The synformal axis plunges at between 20° to 30° towards the east in the western part of the synform, and at about 35° to 45° towards the west at the eastern fold closure, producing a double plunging synform. A dolerite dyke has been emplaced along an east-west fault zone that displaces the northern part of the deposit in a dextral shear sense. The disruption and emplacement of the dolerite along the northern limb of the synform has not affected the quality of the mineralisation. The MQ unit represents a metamorphosed banded iron-formation (BIF). The precursor BIF was subjected to high-grade metamorphic conditions during the Archaean, which resulted in complete recrystallisation of the original fine-grained BIF. In most cases the primary textures have been destroyed by the recrystallisation. Coarse-grained (>1mm) MQ is produced as a result, with good Davis Tube liberation characteristics and concentrate grades at a liberation grind size of 95% passing 80 micron.

The Askaf North Mineral Resource Statement uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

Askaf Centre: The Askaf Centre deposit comprises a northern body that is exposed over a strike length of 3.5km and a southern body that is exposed over a strike length of 1.7km. Both bodies form part of a regional scale antiformal structure and each body is also duplicated within itself by outcrop-scale tight isoclinal folding. The northern and southern bodies are separated and displaced in a dextral shear sense by a regional scale fault/fracture system. The northern body is generally sub-vertical striking roughly northwest-southeast. The magnetite-quartzite unit ranges in thickness from approximately 50m in the west to approximately 70m in the east, with the magnetite-quartzite mineralisation being thinnest in the steep dipping middle portion (±10 m). The multiple layers reported is the result of tight isoclinal folding. The southern body comprises an open synformal structure with an undulating sub-horizontal fold axis that plunges at approximately 25° towards the southwest at the southern part of the deposit. At this locality the mineralisation is still open-ended at depth. The two limbs of the synform are exposed over a strike length of approximately 1km. The northern part of the synform is tighter than is the case in the south, with the eastern limb almost being overturned in some places. Magnetitequartzite ranges in thickness from approximately 30m to 35m in the limbs to approximately 45m to 55m in the synformal keel as a result of structural thickening with thicknesses of up to 90m reported. The magnetite-quartzite unit is embedded within an Archaen granitic/gneiss sequence. The weathered zone which, though variable, has an average vertical thickness of approximately 40m and in this zone partial to complete oxidation of magnetite to hematite has occurred. Oxidation significantly reduces the

Davis Tube mass recovery (wt%) in mineralised drill samples.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

Askaf East: The Askaf East deposit occupies the southern limb of an apparent east-west striking synformal structure defined by an Archaean magnetite-quartzite unit that ranges in true thickness from approximately 20m at the western end of the limb to approximately 140m in the central part of the limb. The hinge zone is at the eastern end of the deposit. The synformal axis plunges about 40° towards the west in the eastern part of the synform. The thickening of the sequence in the eastern part of the deposit is probably as a result of isoclinal folding within the sequence. The MQ unit is embedded within an Archaen granitic/gneiss sequence. The weathered zone, though variable, has an average vertical thickness of approximately 40m. Partial to complete oxidation of magnetite to hematite has occurred in this zone and this significantly reduces the Davis Tube mass recovery (wt%) in mineralised drill samples.

This resource uses a cut-off grade of 20% DTC wt% for fresh (unoxidised) mineralisation and a cut-off grade of 20% head Fe for oxidised mineralisation. All reported concentrate grades were weighted by DTC wt%.

Lebtheinia: The magnetite-rich Banded Iron Formations (BIF) at Lebtheinia form part of the Archaean Lebzena Group. The BIF units in EL264 are exposed over a total strike length of approximately 24km, of which Lebtheinia Centre has a strike length of 11.5km. Parts of the main BIF units at Lebtheinia Centre deposit are covered by laterite and colluvium consisting of BIF fragments.

The magnetite-BIF at Lebtheinia Centre averages about 240m thick. The BIF is characterised by a well-defined banding pattern, with individual bands ("mesobands") averaging 5-10mm thick. Drilling shows that mineralisation extends to at least 400m vertically below natural surface and is open at depth. The deposit is intruded by a series of sub-vertical dolerite dykes, striking NE-SW to NNE-SSW. Lebtheinia Centre has a hanging wall of (variously) quartzite, amphibolite, rhyolite, clay/saprolite (altered amphibolite) and a footwall of quartzite or amphibolite.

The depth of weathering (oxidation) of the BIF averages around 50m. In the lower two thirds of the oxidised zone (the Lower Oxidised Zone, "LOZ") the degree of oxidation is less than in the more oxidised upper third.

The Lebtheinia Centre Mineral Resource Statement for fresh mineralisation uses 20% DTC wt% cut-off. For the LOZ unit the cut-off is 14 SI x 10-3 units of magnetic susceptibility.

The Lebtheinia Mineral Resource Statement has been prepared in accordance with the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (the JORC Code, 2004 Edition).

Zanaga Project: The Zanaga ELs are located within a northsouth oriented greenstone belt which extends for over 47km in length, and vary between 0.5km and 3km in width. The mineralisation is hosted by metamorphosed volcanosedimentary itabirites, and is interbedded with amphibolites and mafic schists. The contact with the crystalline basement is typically faulted and sheared. The principal ore lithologies consist of itabirites, interbedded with basic lavas, which are later altered to amphibolites. Typically, the itabirites consist

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of layers of iron-rich and quartz rich meta-sediments, on a millimetre to centimetre scale. The orebody lithologies are crosscut by late intrusions and dolerite dykes, oriented northeast-southwest. The deposit comprises a sequence of weathering domains, which overlay an un-weathered protore comprising itabirite. The weathered sequence observed at Zanaga is typical of iron ore deposits, where the surficial material demonstrates enrichment in iron above the protore due to a mass reduction and associated leaching of the silicate layers. The Mineral Resource is reported at a 0% Fe cut-off.

The Zanaga Mineral Resource Statement has been prepared in accordance with the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (the JORC Code, 2004 Edition).

The Zanaga Project Ore Reserve Statement was prepared by SRK Consulting (UK) Limited as part of a Feasibility Study.

The full release of this Mineral Resource and Ore Reserve update is available on the Zanaga Iron website (www.zanagairon.com).

- AM = Alan Miller, Independent Consultant (MAusIMM (CP)), responsible for the construction of the geological block model, the grade interpolation and the Mineral Resource estimation (tonnage and grade) and classification.
- JS = Jean-François St-Onge, Eng. formerly of BBA (OIQ).
- GB = Gabor Bacsfalusi, SRK Consulting (UK) Limited (MAusIMM (CP)).
- MT = Malcolm Titley, CSA Global (UK) Ltd (MAusIMM (CP)).
- SvdM = Schalk van der Merwe, Independent Consultant (SACNSP), responsible for the geological interpretation for the Mineral Resource estimation (wireframe model), and the drill hole data set used in these resource estimation.

Australia Coal Resources - New South Wales

	Attributable	Mining		Meas Coal Re	sured sources	Indic Coal Re		Infe Coal Re		Competent
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Oakbridge Group			Thermal Coal (Mt)	1,184	1,184	673	673	1,030	1,030	
Bulga Complex	68.3%	OC/UG	Thermal Coal (Mt)	1,150	1,150	600	600	800	800	CS
			CV (kcal/kg)	5,800	5,800	5,750	5,750	5,600	5,600	
Baal Bone	74.1%	OC/UG	Thermal Coal (Mt)	15	15	13	13	80	80	JMB
			CV (kcal/kg)	5,800	5,800	6,350	6,350	5,000	5,000	
Running Stream	78%	OC/UG	Thermal Coal (Mt)	19	19	60	60	150	150	JMB
			CV (kcal/kg)	5,050	5,050	5,050	5,050	5,150	5,150	
Macquarie Coal JV	80%		Thermal Coal (Mt)	62	62	97	97	30	30	
West Wallsend		UG	Thermal Coal (Mt)	60	60	5	5	_	_	CFP
			CV (kcal/kg)	5,500	5,500	5,100	5,000	-	_	
Cardiff Borehole		UG	Thermal Coal (Mt)	-	_	12	12	30	30	CFP
			CV (kcal/kg)	_	_	5,700	5,700	5,500	5,500	
Teralba		UG	Thermal Coal (Mt)	2	2	. 80	80		-	CFP
			CV (kcal/kg)	6,000	6,000	6,100	6,000	_	_	-
Mitchells Flat	100%	OC/UG	Thermal Coal (Mt)	_	_	120	120	350	400	CFP
			CV (kcal/kg)	_	_	5,750	5,300	5,900	5,000	
Liddell	67.5%	OC/UG	Thermal Coal (Mt)	170	150	180	150	350	400	ST/DS
			CV (kcal/kg)	6,200	6,200	6,100	5,950	6,150	6,150	
Mount Owen Complex	100%		Thermal Coal (Mt)	265	280	155	166	191	241	
Mount Owen		OC	Thermal Coal (Mt)	130	140	80	90	100	150	DS
			CV (kcal/kg)	6,300	6,250	6,050	6,100	6,000	6,050	20
Ravensworth East		OC	Thermal Coal (Mt)	65	65	25	26	1	1	DS
		00	CV (kcal/kg)	5,750	5,700	5,700	5.700	5,950	5,750	20
Glendell		OC	Thermal Coal (Mt)	70	75	50	50	90	90	DS
		00	CV (kcal/kg)	5,900	5,900	5,900	5,950	5,850	5,850	20
			Thermal/Coking	0,000	0,000	0,000	0,000	0,000	0,000	
Integra	100%	UG	Coal (Mt)	15	_	55	_	40	_	DS
j			CV (kcal/kg)	5,400	_	5,450	_	5,400	_	
United	95%	OC/UG	Thermal Coal (Mt)	240	250	230	220	600	600	ML
			CV (kcal/kg)	6,000	6,200	6,000	6,200	5,550	5,850	
Ulan Complex	90%		Thermal Coal (Mt)	265	285	373	413	720	720	
Ulan UGs		UG	Thermal Coal (Mt)	220	240	360	400	700	700	ML
			CV (kcal/kg)	6,600	6,600	4,750	4,700	5,000	5,000	
Ulan Waratah OC		OC	Thermal Coal (Mt)	45	45	13	13	20	20	ML
		00	CV (kcal/kg)	4,950	4,950	5,200	5,200	4,900	4,900	IVIL
Ravensworth Group			Thermal Coal (Mt)	434	434	240	<u>220</u>	<u>100</u>	<u>150</u>	
Narama	100%	OC	Thermal Coal (Mt)	24	24		0.1	_	-	ML
Indiana	10070	00	CV (kcal/kg)	5,600	6,000	_	5,500	_	_	
Ravensworth North	90%	OC	Thermal Coal (Mt)	410	410	240	220	100	150	ML
	3078	00	CV (kcal/kg)	5,900	6,000	6,000	6,100	5,700	5,700	IVIL
Mangoola	100%	00/110	Thermal Coal (Mt)	<u> </u>	140	260	250	1,300	1,200	ML
Mangoola	100 /6	00/00			5,150	4,650	4,650	4,400		IVIL
Tahmoor Complex	100%		CV (kcal/kg) Coking Coal (Mt)	5,200 60	5,150 65	4,850 340	4,850 340	4,400 250	4,300 250	
Tahmoor North	100 /0	UG	Coking Coal (Mt)	40	45	100	100	150	150	N //
Tahmoor South										ML
Ravensworth UG	70%	UG	Coking Coal (Mt)	20	20	240	240	100	100	KP
Navensworth UG	70%	UG	Thermal Coal (Mt)	320	320	220	220	250	250	PH
			CV (kcal/kg)	5,800	5,800	5,400	5,400	5,350	5,350	
Coal Resources New So	outh Wales	Coking	Thermal Coal (Mt)	3,145	3,170	2,943	2,869	5,211	5,271	

Australia Coal Reserves – New South Wales

	Attributable	Mining		Coal Ro Proved	eserves Probable		etable eserves Probable	Total Ma Coal Re		Competen
Name of operation	interest	method	Coal type	31.12.16	31.12.16	31.12.16	31.12.16	31.12.16	31.12.15	Persor
Oakbridge Group	68.3%		Thermal Coal (Mt)	280	34	190	22	212	225	
Bulga OC		OC	Thermal Coal (Mt)	160	15	100	8	108	118	DI
			CV (kcal/kg)			6,450	6,650	6,450	6,450	
Bulga UG		UG	Thermal Coal (Mt)	120	19	90	14	104	107	M
			CV (kcal/kg)			6,400	6,450	6,400	6,450	
Macquarie Coal JV	80%		Thermal Coal (Mt)	_	-	-	_	_	1	
West Wallsend		UG	Thermal Coal (Mt)	-	-	-	-	-	1	AF
			CV (kcal/kg)	-	_	_	_	-	5,500	
Liddell	67.5%	00	Thermal Coal (Mt)	24	13	15	8	23	27	CG
			CV (kcal/kg)			6,700	6,700	6,700	6,700	
Mount Owen Complex	100%		Thermal Coal (Mt)	110	11	62	6	68	80	
Mount Owen		OC	Thermal Coal (Mt)	75	11	40	6	46	51	STH
			CV (kcal/kg)			6,450	6,500	6,450	6,450	
Ravensworth East		OC	Thermal Coal (Mt)	11	_	7	_	7	11	STH
			CV (kcal/kg)		-	6,100	-	6,100	6,400	
Glendell		OC	Thermal Coal (Mt)	24	-	15	-	15	18	STH
			CV (kcal/kg)		_	6,450	_	6,450	6,500	
Integra	100%	UG	Coking Coal (Mt)	2	0.5	1	0.4	1.4	_	KE
			Ash (%)			8.1	8.2	8.1	_	
United	95%		Thermal Coal (Mt)	-	70	-	50	50	54	
United OC		OC	Thermal Coal (Mt)	-	70	-	50	50	30	STH
			CV (kcal/kg)			-	6,450	6,450	6,850	
United A444		UG	Thermal Coal (Mt)	_	_	-	_	_	24	STH
			CV (kcal/kg)			-	-	-	6,850	
Ulan Complex	90%		Thermal Coal (Mt)	161	10	149	10	159	163	
Ulan #3 UG		UG	Thermal Coal (Mt)	55	5	50	5	55	54	EMcG
			CV (kcal/kg)			6,550	6,550	6,550	6,450	
Ulan West UG		UG	Thermal Coal (Mt)	100	5	95	5	100	105	HE
			CV (kcal/kg)			6,400	6,350	6,400	6,400	
Ulan Waratah OC		OC	Thermal Coal (Mt)	6	0.4	4	0.2	4	4	DL
			CV (kcal/kg)			5,000	5,000	5,000	5,000	
Ravensworth Group			Thermal Coal (Mt)	220	12	150	7	157	167	
Ravensworth North	90%	OC	Thermal Coal (Mt)	220	12	150	7	157	167	CG
			CV (kcal/kg)			6,200	6,250	6,200	6,250	
Mangoola	100%	OC	Thermal Coal (Mt)	91	5	76	3	79	90	MM
			CV (kcal/kg)			5,350	5,050	5,350	5,300	
Tahmoor Complex	100%	Cokin	g/Thermal Coal (Mt)	16	41	11	31	42	44	
Tahmoor North		UG	Coking Coal (Mt)	3	11	2	7	9	12	RC
			Ash (%)			9.2	9.3	9.3	9.3	
		UG	Thermal Coal (Mt)	_	_	0.1	1.2	1	1	
			CV (kcal/kg)			5,500	5,600	5,600	5,600	
Tahmoor South		UG	Coking Coal (Mt)	13	30	8	21	29	28	RC
			Ash (%)			9.5	9.5	9.5	9.5	
		UG	Thermal Coal (Mt)	_	_	0.8	2	3	3	
			CV (kcal/kg)			5,850	5,850	5,850	5,850	
Ravensworth UG	70%	UG	Thermal Coal (Mt)	40	_	28	_	28	28	KE
			CV (kcal/kg)			6,950	_	6,950	6,600	
Coal Reserves New So	uth Wales		Thermal Coal (Mt)	926	155	671	109	780	839	
Joan 112351 463 146W 30	uni wales			520	155	071	109	700	039	

Note: Hunter Valley Semisoft coking coal is included in Thermal Coal (Mt)

Notes

The Australian Coal Resources and Reserves are contained within the Sydney Basin (New South Wales), the Bowen Basin (Queensland), the Surat Basin (Queensland) and the Galilee Basin (Queensland).

Coal Resource tonnes have been reported on an in situ moisture basis while Coal Reserves are reported on an as received moisture basis. Coal Resources are reported inclusive of those Coal Resources modified to produce Coal Reserves.

Coal Resources have been re-estimated in 2016 for inclusion in this summary table except where otherwise stated. Revision of the totals includes changes to classifications of Coal Resource status due to exploration, geological reinterpretation remodelling, and changes to lease holdings and depletion by mining.

Thermal Coal Resource qualities are reported at an *in situ* moisture basis and Thermal Coal Reserve qualities are reported on a gross as received basis. Coal Resource qualities for coking coal are not reported, however Coking Coal Reserve qualities are reported on a gross as received basis.

Unless otherwise stated, the product yields used to estimate Marketable Coal Reserves are derived from a simulation package applied to sizing and washability data from each deposit, commonly the "Resource Mastor" software developed by A&B Mylec. The plant simulation package forecasts future plant performance based on historical plant data and exploration drill core analysis.

Changes and issues material to the estimation of Coal Resources and Reserves are noted below for specific projects. Reference to production changes between 31 December 2015 and 31 December 2016 are detailed for each producing mine site.

Tonnages are quoted as million metric tonnes.

Oakbridge Group

Bulga Complex: Overall Coal Resource decrease (-13Mt).

Bulga open cut: Coal Resource and Reserve depletion due to mining (-9Mt).

Coal Reserves for Bulga Open Cut operations are sufficient to support a mine life of 17 years.

Bulga underground: Coal Resource and Reserve depletion due to mining (-4.2Mt).

Coal Reserve decrease due to mine plan changes (-4.4Mt).

Tenements for the Bulga Complex expire between April 2016 and September 2036. Some tenements have expired and Renewal Applications have been submitted.

Baal Bone: Coal Resources were not re-estimated in 2016.

Extraction of the remaining Lithgow Seam and Irondale Seam resources expected to be by underground methods except where the Lithgow, Irondale and Glen Davis Seams are potentially extractable by open cut methods.

Tenements for Baal Bone expire between May 2017 and August 2032.

Running Stream: Running Stream is an undeveloped thermal coal project. Coal Resources were not re-estimated in 2016.

Potential mining methods are open cut for the shallow Coal Resources (less than 60m depth of cover) and underground mining for Coal Resources in excess of 60m depth of cover.

Assessment Lease expires in May 2020. There is no mine plan current for this deposit.

Macquarie JV

West Wallsend: Coal Resource and Reserve depletion due to mining (-1.9Mt).

West Wallsend Mine completed operations in 2016 and mine closure work has commenced. Tenements for West Wallsend, Cardiff Borehole and Teralba expire between January 2020 and January 2030.

Cardiff Borehole: Coal Resources were not re-estimated in 2016. There is currently no mine plan at Cardiff Borehole.

Cardiff Borehole is a potential underground Coal Resource possibly accessible via the existing Teralba Colliery infrastructure.

Teralba: Coal Resources were not re-estimated in 2016.

Teralba was a longwall underground that ceased operations in 2001. The Coal Resource remains suitable for a potential underground mine.

Mitchells Flat: Inferred Coal Resource decrease due to removal of plies deeper than 300m with thin interburden and ply thicknesses (-42Mt).

Increase in Indicated and Inferred Coal Resource CV due to review of coal quality model (Indicated 450kcal/kg and Inferred 900kcal/kg).

Tenements for Mitchells Flat expire in February 2019.

Liddell Open Cut: Coal Resources include both the current Liddell Open Cut Operations and the Glendell North Project. Each area has been assessed by a different Competent Person.

Coal Resource and Reserve depletion due to mining at Liddell Open Cut (-6Mt).

Tenements for Liddell expire between October 2023 and November 2028. Coal Reserves for Liddell operations are sufficient to support a mine life of 6 years.

Coal Resources have increased due to exploration programme (19Mt) and there has been a reclassification of Inferred Resources to Measured and Indicated (37Mt).

Tenements for Glendell North Project expire October 2023.

Mt Owen Complex

Mount Owen: Coal Resource decrease due to transfer of lease areas to the newly acquired Integra Underground (Measured and Indicated -10Mt, Inferred -35Mt). Coal Resource depletion due to mining (-6.5Mt). Figures in table have been reported in accordance with rounding procedures.

Coal Reserve increase due to geological model & dilution factor adjustments (3.4Mt). Coal Reserve depletion due to mining (-9.6Mt).

Tenements for Mt Owen expire between October 2015 and July 2036. Some tenements have expired and Renewal Applications have been submitted.

Coal Reserves for Mt Owen Complex operations are sufficient to support a mine life of 13 years.

Ravensworth East: Coal Resource depletion due to mining (-1.3Mt) and dilution (0.7Mt).

Coal Reserve decrease due to a life of mine plan change to pit limits (-5.7Mt). Coal Reserve depletion due to mining and sterilisation (-1.1Mt).

Decrease in Marketable Reserves CV due to quality model and life of mine plan changes (-300kcal/kg).

Tenements for Ravensworth East expire between July 2019 and October 2034.

Coal Reserves for Ravensworth East operations are sufficient to support a mine life of 6 years.

Glendell: Coal Resource depletion due to mining (-4Mt) and sterilisation (-1Mt).

Coal Reserve depletion due to mining (-4.5Mt) and dilution (-1.3Mt).

Tenements for Glendell expire between October 2018 and March 2032. Coal Reserves for Glendell operations are sufficient to support a mine life of 6 years.

Integra: The assets associated with the Integra underground mine including coal tenements, land, equipment and infrastructure, adjacent to the Mt Owen Complex, were acquired from the Integra JV in December 2015. Open cut mine and coal preparation assets were concurrently acquired by others.

Initial reporting of Coal Resources and Reserves has been completed during 2016. Coal Resources contained within the leases were acquired (65Mt) and, in addition, adjacent Resources from Mt Owen have been transferred (45Mt).

Underground operations are to recommence in 2017 to extract the next two longwall blocks. Reserves reflect these two blocks only.

Tenements for the area acquired by Glencore expire between March 2016 and November 2033. Some tenements have expired and Renewal Applications have been submitted.

United: No material change to Coal Resources since 31 December 2015. Tenements for United expire between January 2015 and March 2033. Some tenements have expired and Renewal Applications have been submitted.

Decrease of Resource coal quality (CV) due to calculation adjustments since the 2015 estimation (Measured - 250kcal/kg, Indicated -200kcal/kg, Inferred -250kcal/kg).

United Open Cut: Coal Reserve has increased due to the expansion of the open cut mine plan (24Mt).

Decrease in Marketable Reserves coal quality (CV) due to open cut mine plan changes (-400kcal/kg).

Coal Reserves for United Open Cut operations are sufficient to support a mine life of 16 years following commencement of operations which is currently under consideration.

United A444 Underground: Coal Reserves have decreased (-36Mt) due to the underground mine project being discontinued in favour of expanding the open cut project above.

Ulan Complex: Coal Resource decrease due to removal of sterilised mining section and re-correlation of coal tops (-38Mt). Underground Coal Resource depletion due to mining (-11.6Mt). Open Cut Coal Resource depletion due to mining (-0.8Mt). Figures in table have been reported in accordance with rounding procedures

Tenements for Ulan expire between February 2017 and January 2036.

Ulan Waratah Open Cut: Coal Reserve depletion due to mining (-0.8Mt).

Ulan West Underground: Coal Reserve depletion due to mining (-7.1Mt).

Coal Reserves for Ulan West underground operations are sufficient to support a mine life of 13 years.

Ulan #3 Underground: Coal Reserve increase due to change in reported moisture basis and review of working section (6.1Mt). Coal Reserve depletion due to mining (-5Mt).

Coal Reserves for Ulan #3 underground operations are sufficient to support a mine life of 12 years.

Ravensworth Group

Narama: No material change to Coal Resources since 31 December 2015. Mining Operations in Narama ceased at the end of 2014 upon completion of the Life of Mine Plan. The area is currently being used by the adjacent Ravensworth North operations for overburden, water management and infrastructure.

Decrease in Measured Resource CV due to 2015 calculation corrections (-400kcal/kg).

Tenements for Narama expire between December 2023 and August 2036.

Ravensworth North: Coal Resource and Reserve depletion due to mining (-11.4Mt).

Coal Reserve increase due to parting thickness changes in some seams (4.9Mt).

Tenements for Ravensworth North expire between May 2014 and February 2034. Some tenements have expired and Renewal Applications have been submitted. Coal Reserves for Ravensworth North operations are sufficient to support a mine life of 18 years.

Mangoola: Coal Resource increase due to additional exploration drill data within Mangoola North allowing reclassification of resources and addition of new Inferred Resources. Coal Resource and Reserve depletion due to mining (-13.5Mt).

Tenements for Mangoola expire between November 2019 and November 2029. Coal Reserves for Mangoola operations are sufficient to support a mine life of 10 years.

Ravensworth Underground: No material change in Coal Resource or Reserve estimations since 31 December 2015.

Production was suspended in September 2014 with the mine currently on care and maintenance.

Increase in Marketable Reserves CV due to 2015 CV calculation corrections (350kcal/kg).

Tenements for Ravensworth Underground expire between September 2012 and August 2036. Some tenements have expired and Renewal Applications have been submitted.

Tahmoor Complex:

Tahmoor North: Coal Resource depletion due to mining (-2.8Mt). Coal Reserve depletion due to mining (-2.5Mt).

Tenements for Tahmoor North expire between August 2016 and March 2035. Some tenements have expired and Renewal Applications have been submitted. Coal Reserves for Tahmoor North underground operations are sufficient to support a mine life of 1 year.

Tahmoor South: Coal Resources were not re-estimated in 2016.

Tenements for Tahmoor South expire in November 2025.

- AF = Alison Freeman, Senior Mining Engineer, West Wallsend Colliery (AusIMM);
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- CG = Chris Gerard, Technical Services Manager, Liddell Open Cut (AusIMM);
- CS = Christopher Smith, Geologist, Bulga Underground Operations (AusIMM);
- DC = David Cahill, Superintendent Technical Services, Ravensworth Open Cut (AusIMM);
- DL = David Lennard, Xenith Consulting/Manager Hunter Valley (AusIMM);
- DS = Daniel Saunders, Geologist, Glendell Mine (AusIMM);
- EMcG = Edward McGonigle, Senior Mining Engineer Ulan Underground Mine (AusIMM);
- HE = Heath Evans, Technical Services Manager, Ulan West Operations (AusIMM);
- JMB = Janet Bartolo, Senior Geologist, McElroy Bryan Geological Services Pty Ltd (AusIMM);
- KB = Konrad Bawelkiewicz, Mining Engineer/Analyst, Glencore Coal Assets Australia (AusIMM);
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- MW = Mark Williams, Technical Services Manager, Mangoola (AusIMM);
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Australia Coal Resources – Queensland

	Attributable	Mining		Meas Coal Re		Indic Coal Re		Infe Coal Re		Competent
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Oaky Creek	55%	OC/UG	Coking Coal (Mt)	260	260	340	350	80	80	RJH
Red Rock	75%	OC/UG	Coking/Thermal Coal (Mt)	0.1	-	150	150	150	150	RJH
			CV (kcal/kg)	6,900	7,450	6,000	5,950	5,700	6,200	
NCA	100%		Coking/Thermal Coal (Mt)	473	513	544	542	1,010	1,018	
Newlands, Suttor		OC/UG	Coking Coal (Mt)	-	13	-	7	-	3	JT/BL
Eastern (RCM)			Thermal Coal (Mt)	310	330	130	120	400	400	JT/BL
			CV (kcal/kg)	5,700	5,700	5,400	5,350	5,100	5,200	
Wollombi (MCM)		OC/UG	Coking Coal (Mt)	10	11	26	27	70	70	BL
			Thermal Coal (Mt)	18	19	60	60	90	90	BL
			CV (kcal/kg)	5,250	4,800	5,050	4,650	5,100	4,550	
Sarum		OC/UG	Coking Coal (Mt)	30	30	8	8	60	60	JT
			Thermal Coal (Mt)	-	-	65	65	250	250	JT
			CV (kcal/kg)	-	-	5,450	5,450	4,650	4,650	
Collinsville		OC/UG	Coking Coal (Mt)	35	35	75	75	40	45	RJH
			Thermal Coal (Mt)	70	75	180	180	100	100	RJH
			CV (kcal/kg)	5,850	5,750	6,250	6,200	5,850	5,800	
Cook	95%	OC/UG	Coking/Thermal Coal (Mt)	_	_	210	100	800	900	DP
			CV (kcal/kg)	-	-	6,650	6,650	6,500	6,850	
Rolleston	75%		Thermal Coal (Mt)	220	225	210	215	370	370	
Rolleston ML		OC	Thermal Coal (Mt)	220	140	210	45	350	20	JB
			CV (kcal/kg)	5,750	5,750	5,550	5,550	5,800	5,450	
Rolleston MDL &		OC	Thermal Coal (Mt)	-	85	-	170	20	350	JB
EPCs			CV (kcal/kg)	-	5,700	-	5,550	5,550	5,550	
Togara North	70%	OC/UG	Thermal Coal (Mt)	370	370	250	250	700	730	DP
			CV (kcal/kg)	6,350	6,350	6,000	6,000	6,000	6,000	
Wandoan	75%		Thermal Coal (Mt)	1,500	1,450	1,750	1,700	4,300	4,400	RJ
			CV (kcal/kg)	5,200	5,200	5,050	5,150	5,200	5,200	
Milray	75%	OC/UG	Thermal Coal (Mt)	-	-	170	-	600	600	RJH
			CV (kcal/kg)	-	-	6,150	-	5,800	4,900	
Pentland	75%	OC/UG	Thermal Coal (Mt)	100	100	40	40	10	10	RJ
			CV (kcal/kg)	4,400	4,400	4,050	4,050	4,100	4,100	
Clermont	25%	OC	Thermal Coal (Mt)	110	110	21	26	10	11	JT
			CV (kcal/kg)	6,200	6,250	6,050	6,050	5,750	5,750	
Coal Resources (Juconclond		Coking/Thermal Coal (Mt)	3,033	3,028	3,685	3,373	8,030	8,269	

Australia Coal Reserves – Queensland

				Coal R	eserves		etable eserves	Total Ma Coal Re		
	Attributable	Mining		Proved	Probable	Proved	Probable			Competent
Name of operation	interest	method	Coal type	31.12.16	31.12.16	31.12.16	31.12.16	31.12.16	31.12.15	Person
Oaky Creek	55%		Coking Coal (Mt)	68	28	45	18	63	71	
Oaky No. 1		UG	Coking Coal (Mt)	2	-	1	-	1	4	SW
			Ash (%)			9.4	_	9.4	9.4	
Oaky North		UG	Coking Coal (Mt)	66	28	44	18	62	67	SW
			Ash (%)			9.4	9.4	9.4	9.4	
NCA	100%		Coking/Thermal Coal (Mt)	80	89	64	67	131	136	
Newlands OC		OC	Coking Coal (Mt)	5	4	3	3	6	15	AC
			Ash (%)			9.2	9.9	9.5	9.3	
		OC	Thermal Coal (Mt)	35	25	27	17	44	37	AC
			CV (kcal/kg)			6,450	6,150	6,350	6,100	
Newlands UG		UG	Thermal Coal (Mt)	_	_	_	_	_	1	POG
			CV (kcal/kg)			-	_	-	6,300	
Collinsville OC		OC	Coking Coal (Mt)	10	10	7	7	14	22	AC
			Ash (%)			10.3	9.7	10	9.9	
		OC	Thermal Coal (Mt)	30	50	27	40	67	61	AC
			CV (kcal/kg)			5,800	5,950	5,900	5,900	
Rolleston	75%	OC	Thermal Coal (Mt)	160	75	160	75	235	230	RM
			CV (kcal/kg)			5,700	5,500	5,600	5,650	
Togara North	70%	OC	Thermal Coal (Mt)	_	28		28	28	28	PJ
			CV (kcal/kg)			_	6,300	6,300	6,300	
Wandoan	75%	OC	Thermal Coal (Mt)	570	200	490	170	660	730	PJ
			CV (kcal/kg)			5,600	5,600	5,600	5,900	
Clermont	25%	OC	Thermal Coal (Mt)	95	23	95	23	118	133	BM
	10,0		CV (kcal/kg)	50	_0	6,200	5,950	6,150	6,150	2.00
Coal Reserves 0	Queensland		Thermal Coal (Mt)	890	401	799	353	1,152	1,220	
			Coking Coal (Mt)	83	42	55	28	83	108	

Notes

Oaky Creek Complex

Oaky Creek: Coal Resource depletion due to mining (-8.6Mt).

Tenements for the Oaky Creek Complex expire between June 2020 and August 2035.

Oaky Creek No. 1: Coal Reserve increase due to minor mine plan changes (1.2Mt). Coal Reserve depletion due to mining (-4Mt).

Coal Reserves are sufficient to support a mine life for a further year.

Oaky Creek North: Coal Reserve depletion due to mining.

Coal Reserves are sufficient to support a mine life for 18-19 years.

Red Rock: Addition of new Inferred Coal Resources due to exploration data (30Mt).

Tenements for Red Rock expire between September 2018 and September 2020.

NCA

- Attributable interest was increased from 55% to 100% in 2016.
- **Newlands Open Cut:** Overall, Coal Reserve decrease (-3Mt) via pit shell adjustments, inclusion of previously unclassified tonnes and depletion from mining.

Increase in Marketable Reserves CV due to greater portion of lower energy coals & depletion of high energy coals (250kcal/kg).

Coal Reserves for Newlands Open Cut operations are sufficient to support a mine life of 8 years.

Newlands, Suttor, Eastern (RCM – Rangal Coal Measure): Newlands Coal Resource decrease due to the closure of the Northern Underground (-33.3Mt). Addition of new Inferred Coal Resources in Newlands North area due to exploration data (30.4Mt). Coal Resource depletion due to mining (-2.6Mt) and sterilisation (-0.7Mt).

Eastern Creek Coal Resource depletion due to mining (-2.1Mt) and dilution (0.3Mt).

Suttor Creek Coal Resource no material change since 31 December 2015.

Tenements for Newlands Complex expire between October 2011 and August 2037. Some tenements have expired and Renewal Applications have been submitted.

Wollombi (MCM – Moranbah Coal Measures): Coal Resource depletion due to mining (-2.4Mt) and sterilisation (-0.1Mt).

Newlands Underground: Newlands Coal Reserve depletion due to mining (-1.3Mt) and dilution changes (0.3Mt).

Sarum: No change in the Coal Resource estimation since 31 December 2012.

The Sarum Project is inclusive of the Sarum and Gattonvale deposits. Tenements at the Project expire between September 2015 and May 2018. An application for the grant of a new, singular, Mineral Development Licence that covers the entire project area has been lodged.

Collinsville: Coal Resource decrease due to sterilisation (Measured and Indicated -2.4Mt, Inferred -4.4Mt) and geology model adjustments (Inferred -5.1Mt).

Coking Coal Reserve decrease due to reclassification as thermal reserves (-6.8Mt) and removal of marginal reserves (-4.3Mt).

Coal Reserves for Collinsville are sufficient to support a mine life for 20 years.

Tenements for Collinsville expire between March 2018 and September 2035.

Cook (Blackrock): Coal Resource reclassification of Inferred Resources to Indicated (123.6Mt). Figures in table have been reported in accordance with rounding procedures.

Tenements for Cook expire between April 2021 and September 2028.

Rolleston: Coal Resource and Reserve depletion due to mining (-13.3Mt) and sterilisation (-1.7Mt). Additional MLs granted during 2016, resulting in transference of resources from EPCs to MLs.

Coal Reserve increase due to addition of Rolleston West in mine plan (12.7Mt) and alterations to software aggregation logic (7Mt).

Tenements for Rolleston expire between May 2016 and May 2043. Some tenements have expired and Renewal Applications have been submitted. Coal Reserves for Rolleston are sufficient to support a mine life of 20 years.

Togara North: Coal Resources were not re-estimated in 2016. The Inferred Coal Resource rounding has been corrected for 2016.

Tenements for Togara North expire between September 2018 and December 2046.

Wandoan: No material change to Coal Resources since 31 December 2015.

Coal Reserve decrease due to change in mining strategy and reassessment of mining layout (-247Mt).

Decrease in Marketable Reserves CV due to change in product composition with additional bypass coal (-300kcal/kg).

Tenements for Wandoan expire between October 2016 and August 2020. Some tenements have expired and Renewal Applications have been submitted. Coal Reserves for Wandoan are sufficient to support a mine life greater than 30 years.

Milray: Coal Resource increase due to interpretation review in light of additional 2015/16 exploration data (152.8Mt).

Tenement for Milray expires in January 2021.

Pentland: Coal Resources were not re-estimated in 2016.

Tenements for Pentland expire between October 2019 and September 2021.

Clermont: Clermont Coal Resources estimated for the extraction of thermal coal via open cut methods.

Coal Resource and Reserve depletion due to mining (-12.5Mt) and sterilisation (-1.9Mt).

Coal Resource increase due to re-correlation (3.7Mt).

Tenements for Clermont expire between April 2020 and November 2027. Coal Reserves at Clermont are sufficient to support a mine life for 10 years.

- AC = Andrew Connell, Principal Mining Engineer, Glencore Coal (AusIMM).
- BL = Bronwyn Leonard, Senior Resource Geologist, Newlands Coal QLD (AusIMM).
- BM = Ben Myall, Production Manager, Clermont (AusIMM).
- DP = Doyle Pryde, Senior Geologist, McElroy Bryan Geological Services Pty Ltd (AusIMM).
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- PJ = Paul Jones, Principal Mining Engineer, Glencore Coal (AusIMM).
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South Africa Coal Resources

	Attributable	Mining		Meas Coal Re		Indic Coal Re		Infer Coal Re		Competent
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Tweefontein	79.8%		Thermal Coal (Mt)	940	940	65	65	40	40	
Tweefontein North		OC/UG	Thermal Coal (Mt)	730	730	-	-	10	10	MS
			CV (kcal/kg)	5,250	5,250			5,500	5,500	
Tweefontein South		OC/UG	Thermal Coal (Mt)	210	210	65	65	30	30	MS
			CV (kcal/kg)	5,350	5,350	4,350	4,350	4,600	4,600	
Goedgevonden	74%	OC/UG	Thermal Coal (Mt)	540	540	13	28	-	-	MS
Complex			CV (kcal/kg)	4,800	4,800	5,000	5,000		_	
iMpunzi	79.8%		Thermal Coal (Mt)	410	420	14	14	3	3	
iMpunzi North		OC	Thermal Coal (Mt)	270	280	4	4	3	3	MS
_ ·			CV (kcal/kg)	5,250	5,250	5,500	5,500	5,600	5,600	
iMpunzi East		OC	Thermal Coal (Mt)	140	140	10	10	_		MS
1			CV (kcal/kg)	5,400	5,400	5,250	5,250			
Zonnebloem Project	100%	OC	Thermal Coal (Mt)	200	200	40	40	_	_	MS
			CV (kcal/kg)	5,150	5,150	4,850	4,850			
Oogiesfontein	100%	UG	Thermal Coal (Mt)	60	60	20	20	_	_	MS
ooglestontein	10070	00	CV (kcal/kg)	4,950	4,950	4,950	4,950			inio
Paardekop	100%	UG	Thermal Coal (Mt)	<u>130</u>	<u>130</u>		<u>-,530</u> 640	90	90	MS
i aaiuekop	100 /8	00	CV (kcal/kg)	5,350	5,350	5,400	5,400	5,350	5,350	1010
Naaitgadaabt	100%	UG	Thermal Coal (Mt)	<u> </u>	<u>3,330</u> 22	<u> </u>	<u> </u>	<u> </u>	<u> </u>	MS
Nooitgedacht	100%	00	()							IVIS
Us david an ed	4000/	00/110	CV (kcal/kg)	4,850	4,850	4,850	4,850	4,850	4,850	
Undeveloped	100%	OC/UG	Thermal Coal (Mt)	-	_	13	13	250	250	MS
Resources			CV (kcal/kg)	-	-	4,750	4,750	5,400	5,400	
Izimbiwa	49.99%		Thermal Coal (Mt)	188	190	110	110	16	16	
Graspan		OC	Thermal coal (Mt)	7	9	-	-	-	-	MS
			CV (kcal/kg)	5 350	5,350					
Townlands		OC	Thermal coal (Mt)	14	15	-	-	-	-	MS
			CV (kcal/kg)	4 850	4,850					
Steelcoal		OC	Thermal coal (Mt)	10	10	-	-	-	-	MS
			CV (kcal/kg)	4,650	4,650					
Lakeside		OC/UG	Thermal coal (Mt)	5	5	-	-	-	-	MS
			CV (kcal/kg)	4,500	4,500					
Leeuwfontein		OC	Thermal coal (Mt)	5	5	-	-	-	-	MS
			CV (kcal/kg)	4,600	4,600					
Springlake		UG/OC	Thermal coal (Mt)	17	17	10	10	6	6	MS
			CV (kcal/kg)	6,300	6,300	6,300	6,300	6,300	6,300	
Argent		OC	Thermal coal (Mt)	29	29	_	_	_	-	MS
			CV (kcal/kg)	5,100	5,100					
Springboklaagte*		UG/OC	Thermal coal (Mt)	86	86	100	100	10	10	MS
			CV (kcal/kg)	4,500	4,500	4,500	4,500	4,500	4,500	
Corobrik		OC	Thermal coal (Mt)	15	16		-			MS
			CV (kcal/kg)	5,100	5,100					

* Springboklaagte is held as a Joint Venture between Izimbiwa and Umcebo, 100% of the Springboklaagte resources are included in the table above under Izimbiwa and excluded from Umcebo.

South Africa Coal Resources (continued)

	Attributable	Mining		Meas Coal Re	sured sources	Indic Coal Re		Infe Coal Re		Competent
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Umcebo	48.67%		Thermal Coal (Mt)	405	410	181	181	30	30	
Klippan		OC/UG	Thermal Coal (Mt)	4	4	1	1	-	-	KD
			CV (kcal/kg)	5,800	5,800	5,800	5,800			
Kleinfontein Jicama			Thermal Coal (Mt)	11	11	-	-	10	10	KD
			CV (kcal/kg)	5,200	5,200			5,200	5,200	
Wonderfontein		OC/UG	Thermal Coal (Mt)	95	100	-	-	-	-	KD
			CV (kcal/kg)	5,350	5,350					
Norwesco		OC	Thermal Coal (Mt)	1	1	-	-	-	_	GC
			CV (kcal/kg)	5,000	5,000					
Doornrug		OC	Thermal Coal (Mt)	4	4	-	-	-	_	GC
			CV (kcal/kg)	5,000	5,000					
Hendrina		UG	Thermal Coal (Mt)	180	180	180	180	10	10	GC
			CV (kcal/kg)	4,250	4,250	4,250	4,250	4,250	4,250	
Belfast	24.34%	UG	Thermal Coal (Mt)	110	110	-	-	10	10	GC
			CV (kcal/kg)	5,100	5,100			5,100	5,100	
Wildfontein	23.26%	UG	Thermal Coal (Mt)	_	2	-	_	_	-	AB
			CV (kcal/kg)		5,000					
Coal Resources Sou	th Africa		Thermal Coal (Mt)	2,895	2,914	1,136	1,151	435	435	

South Africa Coal Reserves

				Extrac Coal Re	eserves	Sale Coal Re	eserves	Total Sa Coal Re	eserves	
Name of operation	Attributable interest	Mining method	Coal type	Proved 31.12.16	Probable 31.12.16	Proved 31.12.16	Probable 31.12.16	31.12.16	31.12.15	Competent Person
Tweefontein	79.8%	metriou	Thermal Coal (Mt)	219	11	142	7	149	149	1 01301
Tweefontein	. 0.070	UG/OC	Thermal Coal (Mt)	210	11	135	7	143	143	TH
North		00,00	Export (Mt)	210		85	4	89	94	
			CV (kcal/kg)			5,900	5,900	5,900	5,900	
			Domestic (Mt)			50	3,000	53	48	
			CV (kcal/kg)			5,100	5,100	5,100	5,100	
Tweefontein		UG/OC	Thermal Coal (Mt)	9		<u> </u>		<u> </u>	<u> </u>	TH
South		00,00	Export (Mt)	5		7		7	7	
South			CV (kcal/kg)			5,900		5,900	5,900	
Coodeeveedee	74%	OC	Thermal Coal (Mt)	290	11		6			СТ
Goedgevonden	7470	00		290	11	180	6	186	206	CI
			Export (Mt)			80	4	84	89	
			CV (kcal/kg)			6,000	6,000	6,000	6,000	
			Export (Mt)			65	-	65	80	
			CV (kcal/kg) Domestic (Mt)			5,100	0	5,100	5,100	
			. ,			35	2	37	37	
iMpunzi	79.8%		CV (kcal/kg) Thermal Coal (Mt)	170	6	5,100 92	5,100 2	5,100 94	5,100 96	
iMpunzi North	13.070	OC	Thermal Coal (Mt)	50	3	92 27	1	94 28	30	TH
		00	Export (Mt)	50	5	24	1	25	27	
			CV (kcal/kg)			5,700	5,700	5,700	5,700	
			Domestic (Mt)						,	
						3	-	3	3	
		OC	CV (kcal/kg) Thermal Coal (Mt)	400	0	5,100		5,100	5,100	
iMpunzi East		00		120	3	65	1	66	66	TH
			Export (Mt)			55	1	56	61	
			CV (kcal/kg)			5,700	5,700	5,700	5,700	
			Domestic (Mt)			10	-	10	5	
7	100%	OC	CV (kcal/kg) Thermal Coal (Mt)		400	5,100		5,100	5,100	
Zonnebloem	100%	00		-	160	-	80	80	80	RR
			Export (Mt)				40	40	40	
			CV (kcal/kg)				6,000	6,000	6,000	
			Domestic (Mt)				40	40	40	
	4000/	~~	CV (kcal/kg)				5,100	5,100	5,100	
Oogiesfontein	100%	OC	Thermal Coal (Mt)	-	8	-	5	5	5	RR
			Export (Mt)				4	4	4	
			CV (kcal/kg)				5,900	5,900	5,900	
			Domestic (Mt)				2	2	2	
			CV (kcal/kg)				5,100	5,100	5,100	
Nooitgedacht	100%	UG	Thermal Coal (Mt)	-	35	_	22	22	22	MS
			Export (Mt)				11	11	11	
			CV (kcal/kg)				5,900	5,900	5,900	
			Domestic (Mt)				11	11	11	
			CV (kcal/kg)				5,100	5,100	5,100	

South Africa Coal Reserves (continued)

				Extrac Coal Re	eserves	Sale Coal Re	eserves	Total S Coal Re		
Name of operation	Attributable interest	Mining method	Coal type	Proved 31.12.16	Probable 31.12.16	Proved 31.12.16	Probable 31.12.16	31.12.16	31.12.15	Competent Person
Izimbiwa Coal	49.99%	moulou	Thermal Coal (Mt)	37	76	26	56	<u>82</u>	<u>82</u>	1 013011
Graspan		OC	Thermal Coal (Mt)	0.7	-	0.5	-	0.5	1	MC
Chuopan			Domestic (Mt)	0.1		0.2		0.2	0.5	ine
			CV (kcal/kg)			5,900		5,900	5,900	
			Domestic (Mt)			0.3		0.3	0.7	
			CV (kcal/kg)			4,500		4,500	4,500	
Townlands		OC	Thermal Coal (Mt)	5	_	3	_	3	3	МС
			Export (Mt)			0.8		0.8	1	
			CV (kcal/kg)			5,800		5,800	5,800	
			Domestic (Mt)			2		2	2	
			CV (kcal/kg)			4,500		4,500	4,500	
Steelcoal		OC	Thermal Coal (Mt)	4	_	4	_	4	4	МС
			Domestic (Mt)			0		_	_	
			CV (kcal/kg)			5,800		5,800	5,800	
			Domestic (Mt)			4		4	4	
			CV (kcal/kg)			4,500		4,500	4,500	
Springlake		UG/OC	Anthracite (Mt)	12	_	6	_	6	6	MC
1 0			Export (Mt)			5		5	5	
			CV (kcal/kg)			6,500		6,500	6,500	
			Domestic (Mt)			. 1		. 1	. 1	
			CV (kcal/kg)			6,500		6,500	6,500	
Argent		OC	Thermal Coal (Mt)	_	26	_	24	24	24	MC
			Export (Mt)				19	19	19	
			CV (kcal/kg)				4,500	4,500	4,500	
			Domestic (Mt)				5	5	5	
			CV (kcal/kg)				4,500	4,500	4,500	
Springboklaagte*		UG/OC	Thermal Coal (Mt)	_	50	_	33	33	33	MC
			Export (Mt)				18	18	18	
			CV (kcal/kg)				6,000	6,000	6,000	
			Domestic (Mt)				15	15	15	
			CV (kcal/kg)				4,500	4,500	4,500	
Corobrik		OC	Thermal Coal (Mt)	15	-	12	_	12	12	МС
			Domestic (Mt)			4		4	4	
			CV (kcal/kg)			5,800		5,800	5,800	
			Domestic (Mt)			8		8	8	
			CV (kcal/kg)			4,500		4,500	4,500	

* Springboklaagte is held as a Joint Venture between Izimbiwa and Umcebo, 100% of the Springboklaagte reserves are included in the table above under Izimbiwa and excluded from Umcebo.

South Africa Coal Reserves (continued)

				Extrac Coal Re		Sale Coal Re		Total Sa Coal Re		
Name of operation	Attributable	Mining			Probable	Proved	Probable	04 40 40	04 40 45	Competent
Umcebo	interest 48.67%	method	Coal type Thermal Coal (Mt)	31.12.16 40	31.12.16	31.12.16 28	31.12.16	31.12.16 28	31.12.15 28	Person
Wonderfontein		OC/UG	Thermal Coal (Mt)	40	_	28	_	28	28	HG
			Export (Mt)			22		22	22	
			CV (kcal/kg)			5,700		5,700	5,700	
			Domestic (Mt)			6		6	6	
			CV (kcal/kg)			4,500		4,500	4,500	
Norwesco		OC	Thermal Coal (Mt)	0.3	-	0.2	-	0.2	0.2	HG
			Export (Mt)			0.2		0.2	0.2	
			CV (kcal/kg)			5,600		5,600	5,600	
			Domestic (Mt)			0.05		0.05	0.05	
			CV (kcal/kg)			4,500		4,500	4,500	
Wildfontein	23.26%	OC	Thermal Coal (Mt)	-	-	-	-	_	2	AB
			Domestic (Mt)						2	
			CV (kcal/kg)						5,000	
Coal Reserves So	outh Africa		Thermal Coal (Mt)	756	307	468	179	646	670	

Notes

Coal Resources have been re-estimated in 2016 for inclusion in this summary table except where otherwise stated. Revision of the totals includes changes to classifications of Coal Resource status due to exploration, geological reinterpretation and remodelling, and changes to lease holdings.

Coal Resources and Reserve qualities are reported at an air dried moisture basis and Export Saleable Coal Reserves are reported at a net as received moisture basis. Coal Resources are reported inclusive of Coal Reserves.

Product yields used to estimate Saleable Coal Reserves were derived from the "Limn Model" software. Inputs to this model are coal ply and in-seam dilution data, processed in the model. The model takes into account plant efficiencies to calculate practical yields. The model is calibrated to historical plant performance and where applicable, large diameter borehole data is used.

Changes and notes relevant to the estimation of Coal Resources and Reserves are listed below for specific projects. Changes reported are exclusive of production from 31 December 2015 to 31 December 2016. Depletion due to mining is based on the actual depletion from January to September, and a forecast for October to December. This forecast number is reconciled each year to the actual and an adjustment is made accordingly.

Coal Resource and Reserve totals are rounded to appropriate levels of accuracy in accordance with the 2007 SAMREC Code (as amended July 2009) and the Glencore Coal rounding process. In summary, Measured and Indicated Coal Resources are rounded to 1 significant figure if less than 10Mt and 2 significant figures if greater than 10Mt; calorific values are rounded to the nearest 50kcal/kg.

Tweefontein Complex

Tweefontein North: Coal Resource depletion due to mining: (-12Mt). Re-evaluation of pillars in the opencast areas (+6.5Mt). Resource gain in Makoupan Pit after additional drilling and remodelling increased the resource footprint (+4.2Mt).

The Tweefontein North development includes all five seams present in the Vryheid Formation. The 1, 2, 4 and 5 seams form part of the mineable and economic Coal Resources. The resources have the potential to be extracted via both opencast truck and shovel or dragline, and underground bord and pillar mining methods.

Coal Reserve depletion due to mining: (-1.8Mt). Probable reserves of 9.9Mt were reclassified as proven reserves in the TUGO 2S highwall area.

Changes in the mine plan resulted in a number of changes in the reserves: review of economic pitshells resulted in an increase in UG reserves (2.5Mt) and OC reserves (3.6Mt). A change in the practical pitshell design in BM Central 2 seam UG area resulted in a loss of reserves (-2.0Mt). Reevaluation of BM North 2 seam reallocated reserves to UG from OC, giving access to additional UG reserves (2.5Mt). No-Coal areas were identified by sub-outcrop drilling in the Klipplaat OC (-1.6Mt).

Mining Tenement for Tweefontein North expires in August 2020. Coal Reserves for Tweefontein North are sufficient to support a mine life of 18 years.

Tweefontein South: Tweefontein South Complex is contained in the Tavistock and Klippoortje old order mining authorisations, situated within the Witbank Coalfield. The complex development includes all five seams, however, only the 1, 2, 4 and 5 seams form part of the potentially mineable and economic Coal Resources.

No mining was conducted in 2016 and the 5 Seam Addcar Coal Reserves remain available for future extraction.

Tenements for Tweefontein South expire between August 2024 and August 2040. Coal Reserves for Tweefontein South are sufficient to support a mine life of 10 years.

Goedgevonden: Coal Resource depletion due to mining (-11.5Mt). Resource loss due to spoils covering coal seams in cut 17, North Pit (-2.9Mt).

The Goedgevonden Complex is situated within the Witbank coal field. Opencast dragline mining operations in the area are extracting the 2, 4 and 5 seams. The 3 seam is too thin for practical extraction and the 1 seam is not considered an economic Coal Resource in the area.

Coal Reserve depletion due to mining (-10.5Mt).

Mine planning changes have reduced Coal Reserves (-5.2Mt) and Marketable Reserves (-3.4Mt). Product optimisation has resulted in a reduction in ROM Coal Reserves (-1.6Mt), an increase in saleable export coal (1.7Mt) and reduction in saleable low grade coal (5.3Mt). Removal of fines flotation recovery from the current plan has resulted in a reduction in low grade saleable reserves (-11.6Mt).

Tenements for Goedgevonden expire between February 2038 and May 2038. Coal Reserves for Goedgevonden are sufficient to support a mine life of 25 years.

iMpunzi

iMpunzi North Coal Resource depletion due to mining (-11.3Mt). Loss of 4 Lower Seam Resources in North East Pit based on the extent of a coal wash out area after additional drilling (-0.14Mt).

iMpunzi consists of iMpunzi Opencast (opencast dragline and truck and shovel operations) and iMpunzi minipits (truck and shovel operations). The Opencast resources include the 1, 2 and 4 seams, whilst the minipits includes only the 4 seam.

Coal Reserve depletion due to mining: (-9.8Mt), Top coal situated immediately above the 2 seam is included in the mining horizon, (previously seen as spoils), after new drilling has proved that the qualities are acceptable, increasing the reserves by 1.9Mt.

Mining tenement for iMpunzi North expires in August 2040. Coal Reserves for iMpunzi North and East are sufficient to support a mine life of 24 years.

iMpunzi East: Resource depletion due to mining (-0.8Mt).

A large portion of the 2 seam and a small area of the 4 seam have been previously mined via underground bord and pillar method. The full seam is extracted through opencast mining methods – the lower zone of each seam was previously partially extracted by underground mining and the upper zone remains intact.

Reserves depletion due to mining (-0.7Mt). Thin 4 upper seam of between 0.5 and 1.0m thick in the VDD West area, which satisfies the quality requirements and overlies 4 lower seam reserves, has been included in the mining horizon increasing the reserves by 1.9Mt

Zonnebloem: The 1 and 2 seams are developed at Zonnebloem and will be extracted by opencast truck and shovel or dragline.

All mining and environmental licensing and permitting has been finalised. Mining tenement for Zonnebloem expires in September 2039. Coal Reserves for Zonnebloem are sufficient to support a mine life of 24 years.

Oogiesfontein: There have been no changes to the Coal Resource or Reserve. A Section 102 application has been approved to incorporate Oogiesfontein into the Goedgevonden Complex. Awaiting environmental licensing and permitting to be finalised.

Mining Tenement for Oogiesfontein expires in October 2018.

Paardekop: The project area comprises underground resources in a single-seam deposit. A mining right application was submitted in 2012. Awaiting approval of right and environmental licensing and permitting.

The only seam of economic importance in Paardekop project area is the Main seam which contains nearly 95% of the extractable coal. The seam has a mean thickness of 2.5m and is structurally almost flat. The upper zone is of poor CV whilst the lower zone has an average raw CV over 5,300kcal/kg.

Undeveloped Coal Resources: Applications for mining rights have been submitted for all the undeveloped Coal Resources. The mining right for Amersfoort was granted while the others are still outstanding. The Coal Resource estimation includes:

Amersfoort (contained in the southern portion of the Ermelo Coalfield, Mpumalanga province southwest of Breyten). Estimation for Amersfoort is based on the C seam which is at a depth of 200m and a thickness of 2.5m;

Boschmanspoort (located in the Witbank Coalfield of Mpumalanga southeast of Middleburg). Estimation for Boschmanspoort is based on the 2 seam which dips towards the east, therefore allowing some potential opencast resources in the west; and

Trichardsfontein (located north of Trichardt). Estimation for Trichardsfontein is based on the 4 seam lower at approximately 90m depth. The seam has potential for underground extraction.

Izimbiwa Coal: Please note the name change from Shanduka to Izimbiwa Coal (Pty) Ltd.

The remaining mine life for individual mining operations range from 2 to 12 years, based on the coal reserves except for the Springboklaagte deposit that extends lzimbiwa's expected life by approximately 20 to 25 years.

Expiry date of relevant mining/concession licenses are different for each mine, ranging from January 2017 to March 2027 in respect of Graspan, Townlands, Steelcoal, Lakeside and Springlake. An application for extension of the Graspan mining right that lapses in January 2017 has been submitted.

Springboklaagte is a prospecting right, which is granted for 5 year periods and renewable for a further three year period. The main prospecting right expired on 3 August 2011 and an application for the renewal of the prospecting right was lodged on 3 June 2011. A mining right was also lodged in April 2011. The mining right was granted in May 2016, awaiting finalisation of the environmental licensing and permitting.

Argent had its mining right granted in May 2016, awaiting finalisation of the environmental licensing and permitting.

Graspan: Coal Resource depletion due to mining (-2.5Mt).

Losses due to geological disturbances including wash-outs, pinching seams and geological structures (-0.4Mt).

Coal Reserve depletion due to mining (-2.5Mt). Coal gained outside mine layout and Portion 31 boundary pillar resulting in a 1.5Mt ROM reserve increase.

Townlands: No mining was conducted at Townlands and the reserves remains unchanged. It forms part of the Graspan LOM and the reserves will be mined as such in the future.

Steelcoal: No change in Coal Resources or Reserves, and the reserves have been earmarked as a future expansion to Graspan.

Lakeside and Leeuwfontein: No change to Coal Resources or Reserves, mines are on care and maintenance. Leeuwfontein remains an 'old order right' or mining licence, with applications pending for conversion into a 'new order right' or mining licence.

Springlake: Coal Resource depletion due to mining (-0.5Mt).

Coal Reserve depletion due to mining (-0.5Mt). Mine placed on care and maintenance.

Corobrik: Resource Depletion due to mining (1.1Mt). Thinning of 1 seam towards the north of North-West Pit and into the Corobrik block resulted in a loss of resources (-0.1Mt)

The reserve forms a natural extension to Graspan Colliery and North Pit was extended into the Corobrik block.

Coal Reserve depletion due to mining (-0.7Mt).

Umcebo: The remaining mine life of the individual mining operations range up to 10 years. Expiry date of relevant mining/concession licenses are different for each mine, ranging from October 2017 to December 2037. Renewals have been lodged for Doornrug, Klippan and Norwesco. The Wonderfontein mining right expires in August 2037.

Klippan: The mine is currently closed with 4.2Mt of Coal Resources remaining.

Kleinfontein Jicama: Mine on care and maintenance.

Wonderfontein: Coal Resource depletion due to mining (-3.3Mt).

Coal Reserve depletion due to mining (-3.2Mt). Change in seam thickness cut-off to from 1.0m to 0.5m resulted in an increase in reserves (2.9Mt).

Norwesco, Doornrug, Hendrina and Belfast: future projects and there are no changes in Coal Resources or Reserves for the current reporting period.

Wildfontein was held under a Joint Venture agreement between Izimbiwa Coal and Hlagisa. The coal reserve was depleted through mining in 2016(1.9Mt).

Competent Persons

- AB = Allen Bullock; N.H.Dip. Coal Mining; Pr Sc Nat (400059/98); Director, Hlagisa Mining (Pty) Ltd;
- CT = Chris Theart; ND, NHD Metal Mining; SAIMM (706513); Group Mining Engineer, Glencore Coal South Africa;
- GC = Gerrit Cronjé, BSc Hons Geology; Pr Sc Nat 400128/86, employed by Izimbiwa Coal (Pty) Ltd;
- HG = Hugo Grobler B Engineering Mining; MSc Engineering Mining; MCC, SAIMM; employed by Izimbiwa Coal (Pty) Ltd;

KD = Karin van Deventer; (MSc Geochemistry; Pr Sc Nat 400705/15, employed by Glencore Coal South Africa;

MC = Mark Cunney, BEng Hons Mining Engineering, MCC; Pr Cert Eng 2007 0114, employed by lzimbiwa Coal (Pty) Ltd;

- MS = Marius Smith; B Sc. Honours Geology; MBA; Pr Sc Nat 400075/03; Group Coal Geologist, Glencore Coal South Africa;
- RR = Rohan Roach; B Com; B Tech Mine Engineering; Coal Mine Managers Certificate of Competency, Metalliferous Mine Managers Certificate of Competency; SAIMM, ECSA; Group Mining Engineer, Glencore Coal South Africa;
- TH = Trevor Howard; B Eng. Mining; Coal Mine Managers Certificate of Competency; SAIMM (701062); Group Mining Engineer, Glencore Coal South Africa;

Colombia Coal Resources and Reserves

Prodeco Coal Resources

Coal Resources	Coal Resources Prodeco		Thermal Coal (Mt)	175	185	210	220	70	70	
			CV (kcal/kg)	7,100	7,100	7,050	7,050	-	-	
La Jagua	100%	OC	Thermal Coal (Mt)	65	75	30	30	-	-	KJW
			CV (kcal/kg)	6,300	6,300	6,200	6,200	6,250	6,250	
Calenturitas	100%	OC	Thermal Coal (Mt)	110	110	180	190	70	70	KJW
Name of operation	Attributable interest	Mining Method	Commodity		sured sources 31.12.15	Indic Coal Re 31.12.16	sources		rred sources 31.12.15	Competent Person

Prodeco Coal Reserves

				Coal R	eserves		etable eserves		arketable eserves	
Name of operation	Attributable interest	Mining method	Coal type	Proved 31.12.16	Probable 31.12.16	Proved 31.12.16	Probable 31.12.16	31.12.16	31.12.15	Competent Person
Calenturitas	100%	OC	Thermal Coal (Mt)	45	55	45	55	100	110	GL
			CV (kcal/kg)	6,200	6,100	6,200	6,100	6,150	6,100	
La Jagua	100%	OC	Thermal Coal (Mt)	55	25	55	25	80	90	GL
			CV (kcal/kg)	6,800	6,700	6,800	6,700	6,750	6,750	
Coal Reserves I	Prodeco		Thermal Coal (Mt)	100	80	100	80	180	200	

Cerrejón Coal Resources

	Attributable	Mining		Meas Coal Re	sured sources	Indic Coal Re	ated sources	Infe Coal Re		Competent
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
Carbones del	33.3%	00	Thermal Coal (Mt)	3,000	2,950	1,250	1,150	650	800	GH
Cerrejón			CV (kcal/kg)	6,550	6,550	6,600	6,550	6,450	6,550	

Cerrejón Coal Reserves

Cerrejón			CV (kcal/kg)	6,100	6,150	6,100	6,100	6,100	6,100	
Carbones del	33.3%	OC	Thermal Coal (Mt)	490	70	470	70	540	610	GH
Name of operation	interest	method	Coal type	31.12.16	31.12.16	31.12.16	31.12.16	31.12.16	31.12.15	Person
	Attributable	Mining		Proved	Probable	Proved	Probable			Competent
					ctable eserves		eable eserves		aleable eserves	

Notes

Glencore's Colombian coal interests are located in two different coal provinces: in the La Guajira Department (Cerrejón) and the Cesar Department (Prodeco).

Extractable Reserves are as mined Coal Reserves taking into account geological losses, mining losses, contamination and as mined moisture adjustments. Reserves are reported on a ROM moisture basis. Coal Resources are reported on an *in situ* moisture basis.

Saleable Reserves: As sold basis are Coal Reserves adjusted for yield losses in the preparation plant (if applicable) and converted to a saleable moisture basis. The Coal Resource and Coal Reserve estimates tabulated above are stated on a total mine basis as at 31 December 2016.

Coal Resource qualities are reported on an *in situ* moisture basis and Coal Reserve qualities are reported on a gross as received basis. Coal Resources are reported inclusive of those Coal Resources modified to produce Coal Reserves. Coal tonnages are quoted as million metric tonnes

Changes and issues material to the estimation of Coal Resources and Reserves are noted below for specific projects. Reference to production changes between 31 December 2015 and 31 December 2016 are detailed for each producing mine site. Coal Resource and Coal Reserve totals are rounded to appropriate levels of accuracy in accordance with the 2012 JORC Code and the Glencore Coal Assets rounding procedures.

Prodeco:

Calenturitas: Coal Resources decreased due to mining (-10Mt).

Marketable Coal Reserves depletion due to mining (-12Mt). Coal Reserves increases due to new geological model (1Mt).

Remaining mine life expected to be 11 years. Expiry date of relevant mining/concession licenses: 2035.

La Jagua: Coal Resource decreased due to mining (-7Mt), mining sterilisation of M45 seam for pit stability (-1.7Mt) and change in modelling technique of seams M17 and M27 from in-pit observations showing thinner thicknesses for these seams (-1.3Mt).

Marketable Coal Reserves depletion due to mining (-7Mt). Coal Reserves decreases due to new geological model (-1Mt).

Remaining mine life expected to be 12 years. Expiry date of relevant mining/concession licenses: Carbones El Tesoro (CET), Consorcio Minero Unido (CMU) and Carbones de La Jagua (CDJ) expire between 2027 and 2038.

Cerrejón: Coal Resources are reported as gross tonnes in situ, i.e. thin seams (<0.65m) excluded with no geological losses applied. The Coal Resources occur within a 'geoshell' constrained by the horizontal and vertical distribution of data within the drill hole (data limits) envelope. Resources include coal for which the continuity, quality and mineability are established but are outside the current LOM plan. There are approximately 220Mt of coal that occurs within 1 km of major towns. This coal has not been included in the Coal Resources in 2016. These Coal Resources comply with current and foreseen mining and marketing criteria and have economic potential.

Coal Reserves have reduced by 40Mt due principally to depletion by mining (-33.1Mt).

The estimates of Coal Resources and Coal Reserves presented in this table for Cerrejón have been prepared in accordance with the 2007 SAMREC Code (South African Code for Reporting of Mineral Resources and Mineral Reserves) (as amended July 2009).

Competent Persons

 $\mathsf{GL}=\mathsf{Guillermo}$ Leon, Superintendent Mine Planning, Prodeco, (AusIMM).

KJW = Kerry Whitby, Managing Director, McElroy Bryan Geological Services Pty Ltd (AusIMM).

GH = German Hernandez; BSc, BHPBilliton Certificate of Competent Person; GSSA; APS Geology Superintendent, Carbones del Cerrejón.

Canada Coal Resources

Coal Resources	S Canada		Coking/Thermal Coal (Mt)	45	45	113	113	130	130	
Sukunka	75%	UG/OC	Coking Coal (Mt)	45	45	100	100	40	40	KJW
			CV (kcal/kg)	-	-	6,100	6,100	6,100	6,100	
Suska	75%	OC	Coking/Thermal Coal (Mt)	-	-	13	13	90	90	KJW
Name of operation	interest	Method	Commodity	31.12.16	31.12.15	31.12.16	31.12.15	31.12.16	31.12.15	Person
	Attributable	Mining			sured sources		ated sources		rred sources	Qualified

Notes

Glencore's Canadian coal resources and reserves (Sukunka, Suska) occur in the Peace River area of the Province of British Columbia. Additional tenements adjacent to these Peace River projects are targeted for exploration. These include tenement areas identified as Central South, South Cirque and other tenements that extend north and south of the Pine River.

Coal Resource tonnage and quality are reported at an *in situ* moisture basis. Coal Resources are reported in accordance with the JORC Code 2012 edition.

No coal was mined from the Glencore Canadian coal assets in 2016.

Suska: Coal Resources have not been re-estimated since 2013.

Sukunka: Coal Resources have not been re-estimated since 2015.

Competent Person

KJW = Kerry Whitby, Managing Director, McElroy Bryan Geological Services Pty Ltd (AusIMM).

Energy Products Oil

Net Reserves (Proven and Probable)¹

				Workir	ng Interest Basi	5			
	Equatorial C	Guinea	Chad		Camero	on		Total	
	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Combined mmboe
31 December 2015	19	-	166	-	-	-	185	-	185
Revisions	(2)	-	(9)	-	-	-	(11)	_	(11)
Production	(4)	-	(5)	-	-	-	(9)	_	(9)
31 December 2016	13	-	152	-	_	-	165	_	165

Net Contingent Resources (2C)¹

	Working Interest Basis								
	Equatorial Guinea		Chad		Cameroon		Total		
	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Oil mmbbl	Gas bcf	Combined mmboe
31 December 2015	23	562	9	-	47	595	79	1,157	278
Revisions	2	(32)	_	_	_	_	2	(32)	(3)
Acquisitions/Divestments	-	_	_	-	(29)	(595)	(29)	(595)	(132)
31 December 2016	25	530	9	-	18	-	52	530	143

¹ "Net" Reserves or Resources are equivalent to Glencore's working interest in the asset/property.

Notes

Equatorial Guinea: Equatorial Guinea Reserves and Contingent Resources consist of Block O (Glencore 25% working interest ("WI")) and Block I (Glencore 23.75% WI) reserves and resources.

The Aseng field (Block I, 23.75% WI) came on stream in November 2011. The field is produced from subsea wells tied back to an FPSO. Average 2016 gross production was 27,800 barrels per day.

The Alen field (95% Block O, 25% WI and 5% Block I, 23.75% WI) came on stream in May 2013. Gross production in 2016 averaged 18,600 barrels per day. The field is produced from subsea wells tied back to a production platform where condensate is stripped and transported to the Aseng FPSO via a subsea pipeline. The produced gas is re-injected into the field.

The Aseng and Alen fields have a 25 year exploitation term from approval of a plan of development.

Reserves for Equatorial Guinea were independently assessed by Gaffney, Cline & Associates (GCA), have been prepared in accordance with the Petroleum Resources Management System (PRMS) and have been extracted without material adjustment from the GCA report dated 31 December 2016. Contingent Resources are based on Glencore and GCA estimates and have been prepared in accordance with PRMS.

Chad: Glencore holds a majority WI in the DOB/DOI, Doseo/Borogop and DOH production sharing contracts. Glencore holds an 85% WI in the Badila and Mangara oil field Exclusive Exploitation Authorisations (EXAs). In addition McDaniel have classified the Kibea, Baouda East and Krim discovered fields as reserves in which Glencore has a 75% WI. The Badila field is an onshore development which came on stream in September 2013. Oil is transported through an export pipeline to the Chad/Cameroon export pipeline (Totco/Cotco pipeline) with off-take at the Marine Terminal in Cameroon. Average gross 2016 production was 10,000 barrels per day.

The Mangara field is an onshore development that has been producing since late December 2014. Gross production in 2016 averaged 4,500 barrels per day. Oil is transported through an export pipeline to the Totco/Cotco pipeline with off-take at the Marine Terminal in Cameroon.

The EXA's have a 25 year exploitation term after the authorisation of the EXA.

Reserves for Chad were independently assessed by McDaniel & Associates (McDaniel), have been prepared in accordance with PRMS and have been extracted without material adjustment from the McDaniel report dated 31 December 2016. Contingent Resources are based on Glencore estimates and have been prepared in accordance with PRMS.

Cameroon: Glencore holds 100% WI of the Bolongo license. Development studies are currently underway on the Bolongo licence and as such there are no reserves currently attributable to the licence. Contingent resources are based on Glencore estimates and have been prepared in accordance with PRMS.

Glencore has agreed to assign its entire interest in the Matanda Block to Gaz du Cameroun Matanda S.A. Subject to government formalities, the effective date of the transfer will be backdated to March 2016 and as such there are no reserves or contingent resources attributable to the licence.