

ASX Announcement
2 February 2022

Significant Resource Upgrade at Twin Hills Project

HIGHLIGHTS

- Updated JORC 2012 Mineral Resource estimate for The Twin Hills deposits (Lone Sister and 309) of **12.9 million tonnes at 1.8 g/t Au for 760,700 ounces (previously 633,000 ounces)** of contained gold on granted mining leases (see Table 1).
- **Combined gold resources under GBM's ownership in the Drummond Basin now total approximately 1.6 Moz** across the Yandan, Mt Coolon and Twin Hills assets (see Figure 1). All are located within 70 km of GBM's proposed "Drummond Basin Processing Hub" centred on the Yandan ML's, which has significant infrastructure (i.e. water storage dams, tailings facilities, airstrip, leach pads and grid power).

Drummond Basin, QLD Total - 41 million tonnes at 1.2 g/t Au for 1,612,200 ounces

- Twin Hills is considered **highly prospective for the discovery of additional mineralisation**, with preliminary analysis suggesting high grade gold shoots at 309 and Lone Sister may be open at depth. **GBM is now aiming to commence drilling in the coming weeks at Twin Hills** to test these targets.
- **GBM's analysis of the deposit geometry and grade distribution suggests that the 309 and Lone Sister resources maybe be successfully mined via a combination of bulk minable open pit and bulk underground mining methods.**
- **Resource upgrade work will continue with new pit optimisation work commencing soon on the 309 deposit to evaluate opportunities to deepen the pit below the 1050 RL used in this resource update.**

GBM Managing Director and CEO, Peter Rohner, commented:

"GBM's geological and mining team is rapidly gaining further knowledge of the Twin Hills deposits to improve the resource model and evaluate mining scenarios which will assist in focusing drilling activities which will commence shortly. The planned drilling programs will focus on increasing the confidence of the resource, improving resource knowledge in the shallower likely open pit resources and extending the resources at depth. This transformational acquisition represents a significant step in the execution of our 'processing halo' strategy to build over two million ounces under ownership within the Drummond Basin, providing an entry into the development of a genuine mid-tier Australian gold company."

Twin Hills Resource Summary

The 309 and Lone Sister deposits are low sulphidation, epithermal gold deposits hosted within the western arm of the Drummond Basin in Queensland. The Drummond Basin is host to a number of significant gold deposits and is considered by GBM to hold potential for further discoveries.

The 309 and Lone Sister gold deposits are located 7 kilometres apart and linked by a major north-south structural lineament. Both deposits have previously been interpreted as intrusion related, high gold fineness, low sulphidation epithermal gold deposits, sometimes exhibiting bonanza gold grades (as evidenced by the peak gold value in the 309 deposit of 2,940 g/t Au, with 300 individual metre samples exceeding 30 g/t Au, and a peak gold value of 939 g/t Au at Lone Sister). Refer ASX: GBZ release 18 January 2019.

GBM considers that potential depth extensions and strike repetition of both the 309 and Lone Sister deposits have not been adequately tested.

No changes were made to the 2019 resource model, however metallurgical and mining costs were reviewed and as a result of the increase in the gold price the cut-off grade was reduced from 1.0 g/t Au to 0.4 g/t Au for the open pit resources.

The 309 Deposit has been re-estimated to comprise 10.8 Mt averaging 1.4 g/t Au containing 500,600 ounces of gold and 783,100 ounces of silver (assuming open pit mining to 1050 RL, or a depth of approximately 200 m, and underground mining below 1050 RL).

The Lone Sister Deposit is re-estimated at 2.0 Mt at an average grade of 4.0 g/t Au containing 260,100 ounces of gold and 604,800 ounces of silver (refer Table 1).

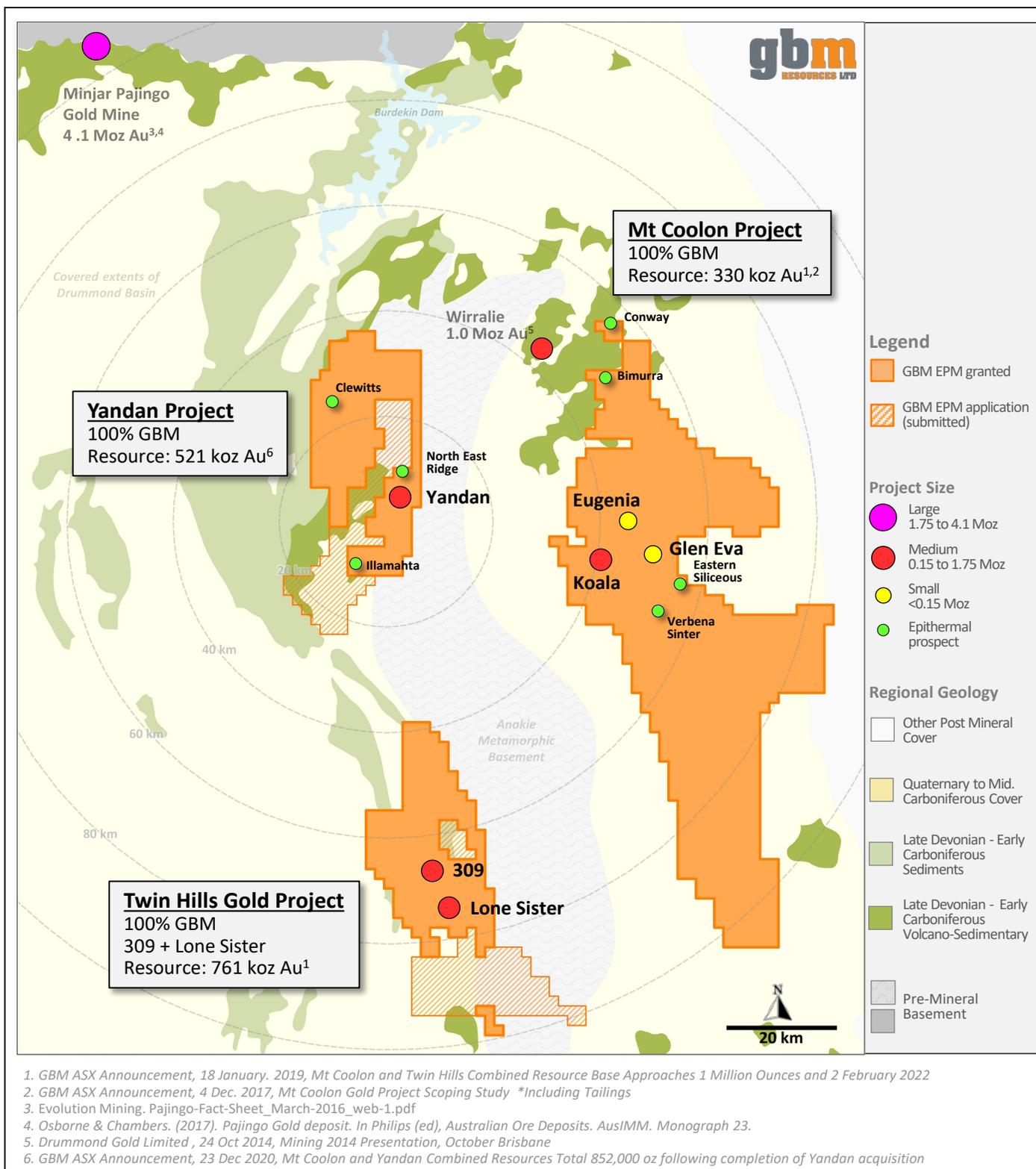
Table 1: Twin Hills Resource Summary for the 309 and Lone Sister Gold Deposits (rounded for reporting '000 tonnes, '00 ounces, 0.0 grade). See previous release ASX GBM 18 January 2019 'Mount Coolon and Twin Hills Combined Resource Base Approaches 1 Million Ounces' also.

Open Pit Resources (above 1050 RL) stated at 0.4 g/t Au cut-off and Underground Resources (below 1,050 RL) stated at 2.0 g/t Au cut-off.

Category	Cut-off	Tonnage (t)	Grade		Contained Metal	
	Au (g/t)		Au (g/t)	Ag (g/t)	Au (oz)	Ag (oz)
309 Deposit						
Open Pit (above 1050RL)						
Measured	0.4	586,000	2.7	4.8	50,300	85,500
Indicated	0.4	5,470,000	1.4	2.6	253,200	457,500
Inferred	0.4	4,165,000	0.9	1.5	120,200	199,400
Total open pit	0.4	10,220,000	1.3	2.3	423,500	742,400
Underground (below 1050 RL)						
Measured	2.0					
Indicated	2.0	110,000	4.8	3.4	16,800	11,900
Inferred	2.0	510,000	3.7	1.8	60,100	28,800
Total underground	2.0	620,000	3.9	2.0	76,900	40,700
Total 309 Deposit						
Measured	0.4 / 2.0	586,000	2.7	4.8	50,300	85,500
Indicated	0.4 / 2.0	5,580,000	1.5	2.6	270,000	469,400
Inferred	0.4 / 2.0	4,675,000	1.2	1.5	180,300	228,200
TOTAL	0.4 / 2.0	10,841,000	1.4	2.2	500,600	763,100
Lone Sister Deposit						
Measured	2.0					
Indicated	2.0					
Inferred	2.0	2,010,000	4.0	9.4	260,100	604,800
Total	2.0	2,010,000	4.0	9.4	260,100	604,800
Total Twin Hills						
Measured	0.4 / 2.0	586,000	2.7	4.5	50,300	85,500
Indicated	0.4 / 2.0	5,580,000	1.5	2.6	270,000	469,400
Inferred	0.4 / 2.0	6,685,000	2.0	3.9	440,400	833,000
TOTAL	0.4 / 2.0	12,851,000	1.8	3.4	760,700	1,387,900

Drummond Basin Consolidation and Processing Halo Strategy – now at ~1.6 Moz

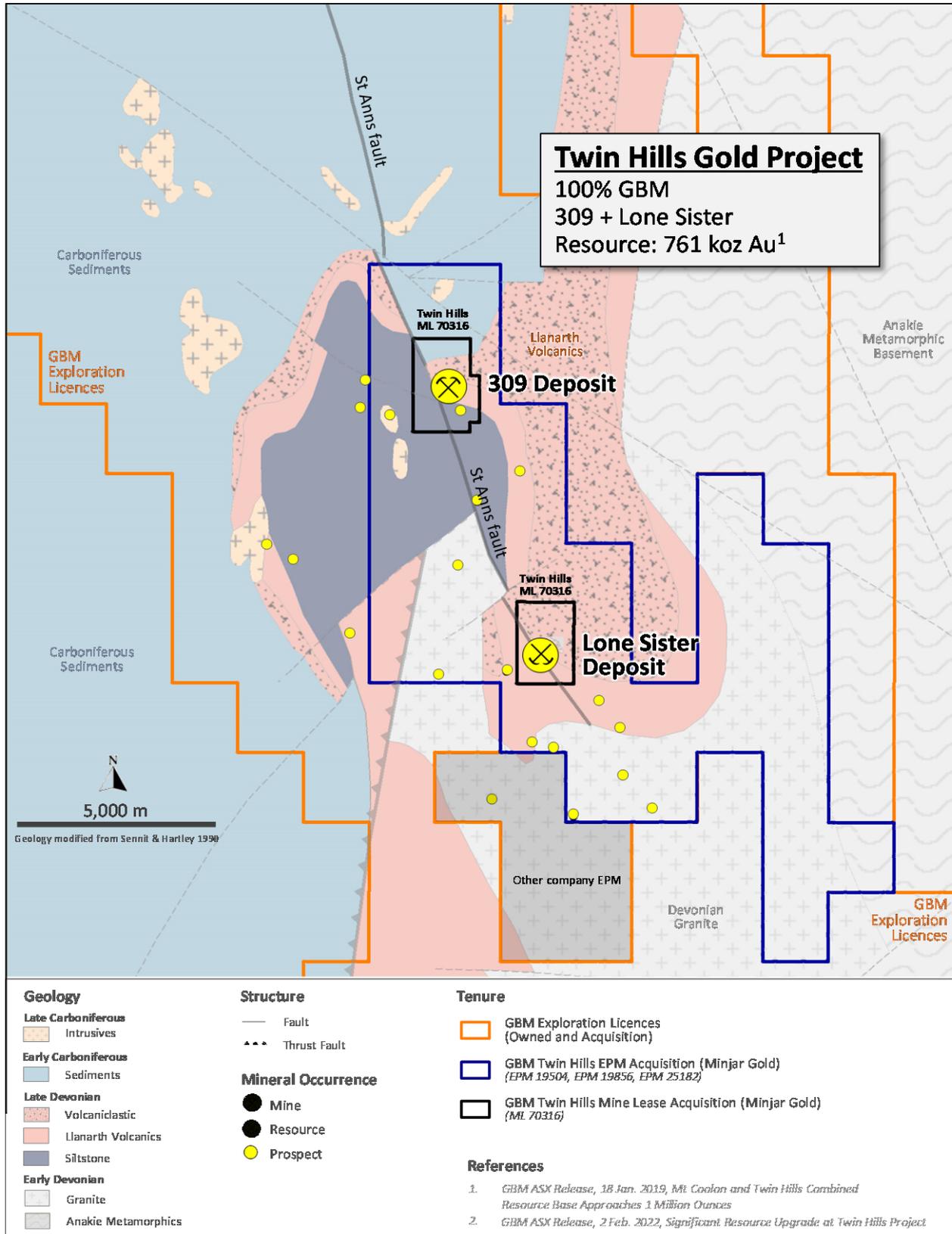
Figure 1: Drummond Basin Processing Halo and Resource Consolidation Summary



Twin Hills Geology and Exploration Potential – High Grade Shoots Open to Depth

Twin Hills is hosted by a sedimentary-volcanic package interpreted to have been deposited in a late Devonian age, structurally controlled, pull apart basin that formed along the margin of a Cambro-Ordovician age metamorphic basement high, the Anakie metamorphic inlier (refer Figure 2).

Figure 2: Geological setting of 309 and Lone Sister Deposits



Gold-silver mineralisation is temporally and probably genetically linked to subvolcanic to volcanic felsic domes and related breccia pipes. The age of mineralisation at Lone Sister and by association the related domes has been shown to be early Carboniferous (341 to 346 ma).

Mineralisation at the Twin Hills project, 309 and Lone Sister deposits (Figure 2) belongs to the felsic dome related, high gold fineness, low sulphidation quartz sulphide class of mineralisation that has produced a number of notable high value gold deposits including the high-grade Sleeper deposit and large bulk minable style deposits like Round Mountain in Nevada. This class of deposit usually develops an early phase of quartz-sulphide gold mineralisation followed by later stages of very high-grade often free gold quartz and or gold electrum chalcedony events, as is seen at Twin Hills, that are important to the deposit economics.

GBM's preliminary interpretation shows that the Twin Hills deposits are characterised by the 309 (phreatomagmatic to phreato-hydrothermal) milled matrix breccia body and the Lone Sister breccia and veinlet zone that is hosted within a rhyolite feeder dyke to a flow dome and the adjacent wall rock sediments. Better gold mineralisation in these deposits is strongly associated with epithermal quartz breccia matrix fill and cross cutting quartz fracture veinlet networks, forming discontinuous veinlet corridors that crosscut the host rock. GBM believes that the 309 and Lone Sister deposit characteristics are better suited to open pit or underground bulk mining approach compared to the selective underground mining previously used at Twin Hills.

Figure 3: Plan view of the Twin Hills 309 gold deposits. These show ore blocks in various grade ranges and previous underground development.

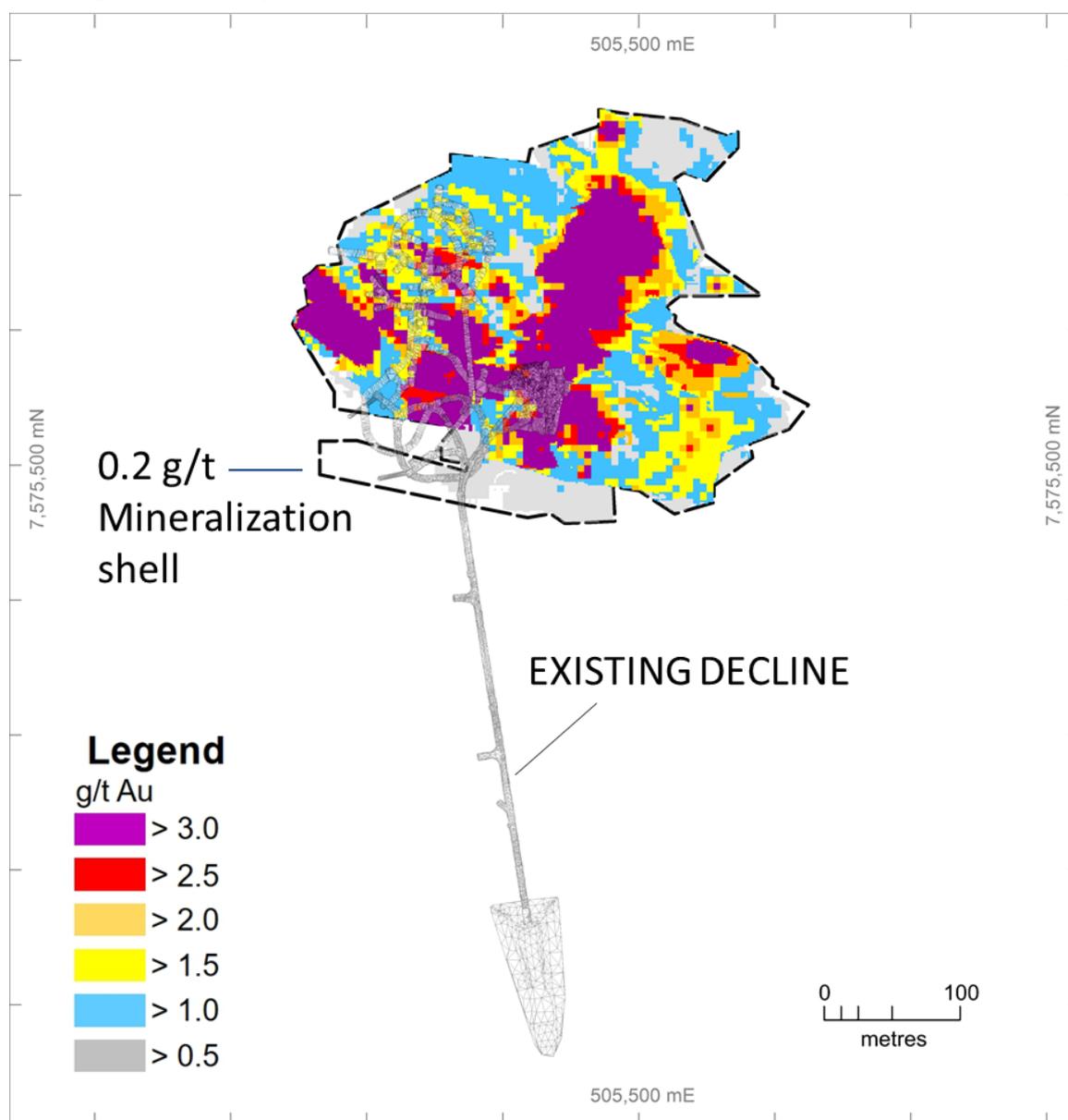


Figure 4: Long Projection of the Twin Hills 309 gold deposits. These show ore blocks in various grade ranges and previous underground development.

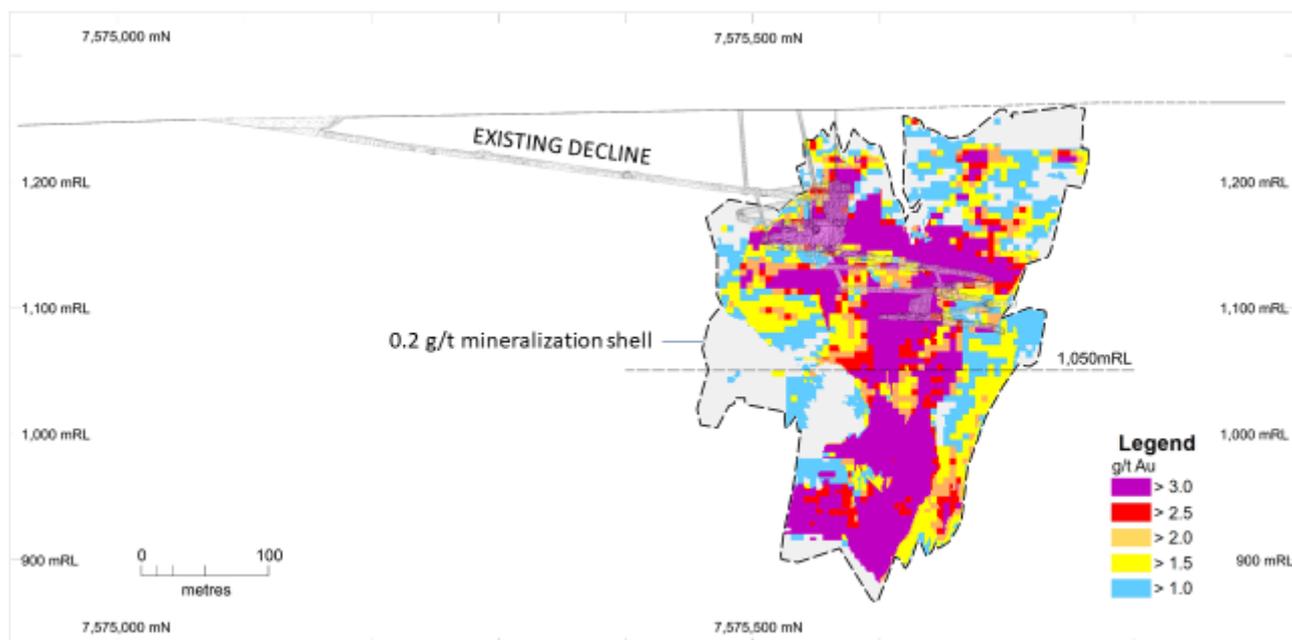


Table 2: 309 and Lone Sister Deposit Selected Historic Down Hole Gold Drill Intersections (Refer ASX: GBZ release 18 January 2019).

309 Deposit Selected Length Weighted Au g/t Intersections including High Grade 1.0 g/t Au Intercepts														
Length Weighted Au g/t Intersections Nominal 0.3 g/t Au cut off									Including Maximum Intervals at 1.0 g/t Au cut off					
Hole ID	From (m)	To (m)	Interval (m)	Au g/t	Ag g/t	Au Gram Metres	Lode		From (m)	To (m)	Interval (m)	Au g/t	Ag g/t	Au Gram Metres
THRCD875	154	294.6	140.6	5.1	3.2	723.8	a	incl.	177	185	8	81.8	43.0	654.7
THRCD816	32	181	149	4.6	22.3	687.8	a	incl.	54	76	22	27.4	142.0	603.6
THRCD828	5	174	169	2.9	11.3	494.3	b	incl.	101	111	10	21.7	133.2	217.1
THRC761	75	146	71	6.3	20.8	445.4	b	incl.	111	125	14	16.4	54.7	228.9
TRCD384	25	215	190	2.3	4.0	437.2	b	incl.	100	127	27	7.3	10.3	198.1
THRCD827	236	408	172	2.1	0.8	361.8	a	incl.	382	396	14	11.0	2.4	154.4
THRCD843	106	262	156	2.3	2.4	360.2	a	incl.	108	119	11	13.0	20.0	143.4
THRCD826	241	315	74	4.7	2.3	350.7	a	incl.	269	274	5	65.5	29.4	327.3
THDD885	25	134	109	2.5	10.0	268.2	b	incl.	45	56	11	6.0	30.5	66.3
THRCD861	154	233.9	79.9	3.2	3.9	256.3	a	incl.	183.5	188.5	5	28.7	26.8	143.6
THRC781	20	109	89	3.0	9.2	270.6	a	incl.	56	81	25	9.9	29.0	247.3
THRCD844	84	236	152	1.3	0.9	199.8	a	incl.	201	214	13	7.1	2.4	92.0
THRCD873	147	261	114	1.9	3.9	213.1	a	incl.	194	206	12	8.9	11.6	107.2

Lone Sister Deposit Selected Length Weighted Au g/t Intersections including High Grade 1.0 g/t Au Intercepts														
Length Weighted Au g/t Intersections Nominal 0.3 g/t Au cut off									Including Maximum Intervals at 1.0 g/t Au cut off					
Hole ID	From (m)	To (m)	Interval (m)	Au g/t	Ag g/t	Au Gram Metres	Lode		From (m)	To (m)	Interval (m)	Au g/t	Ag g/t	Au Gram Metres
LRCD015	104	250	146	9.8	8.7	1433.3	a	incl.	211	239	28	45.2	41.3	1266.8
LRCD154	216	242	26	10.3	13.8	267.5	a	incl.	218	229	11	23.4	31.4	257.9
LRCD057	121	217	96	5.5	0.0	529.3	a	incl.	177	189	12	27.1	0.0	324.9
LRCD063	158	268	110	3.6	4.8	395.6	a	incl.	247	267	20	14.3	26.6	285.3
LRCD157	168	248	80	4.7	10.6	373.8	a	incl.	219	231	12	13.8	18.9	165.9
LRCD152	243	359	116	2.8	5.0	321.7	a	incl.	245	263	18	6.0	12.8	108.7
LRCD012	222	375	153	2.0	0.4	309.7	a	incl.	343	352	9	5.8	0.5	51.9
LRCD140	128	382	254	1.2	3.9	306.8	a	incl.	265	277	12	4.1	5.4	49.5
LRCD143	124	216	92	3.3	4.8	301.1	a	incl.	139	157	18	12.4	10.1	224.0
LRCD147	80	210	130	2.3	3.5	294.4	a	incl.	172	187	15	4.8	8.5	71.9
LRC180	8	106	98	1.4	0.0	137.0	a	incl.	19	25	6	8.5	1.5	51.2
LRCD064	24	132	108	1.3	0.0	139.3	a	incl.	85	101	16	5.6	0.0	89.5
LRCD134	69	137.93	68.93	2.4	10.0	164.9	a	incl.	75	86	11	3.9	23.5	42.4

Length weighted downhole intercepts were manually selected using a combination of logged geology and Au grade above 0.3 g/t Au. Internal dilution was typically < 2 m but may include intervals of 5 to 10 m in some instances. 1 g/t Au composite calculated with 1 g/t Au cut off grade and a maximum 2 m internal dilution @ 0.1 g/t Au. No high-grade cut was applied. Selected intercepts at 309 Deposit do not include intersections which have passed through mining voids (have been mined out), or were drilled sub-parallel to the interpreted strike of the Lode. No underground drilling has been used in composite calculations (UG prefix holes, drilled primarily within the mining void).

Note: all intercepts quoted in Table 2 are outside any previously mined areas.

BMA established an underground mine at the 309 deposit and completed a decline to approximately 160 m below surface, plus the development of two ore lenses. Ore produced was trucked to the Rishton Mill east of Charters Towers, some 280 kilometres away. With gold trading at AUD\$572/oz in June 2005 and prices for consumable and labour rising with the mining boom, the operation was forced to model and selectively mine ore zones at a high cut-off grade (considered to be around 6 g/t Au). The result appears to be that mineralisation was not continuous at these high cut-off grades yielding disappointing head grades for the second ore zone and resulting in closure of operations around January 2007. Current economics suggest the potential for GBM to operate at much lower cut-off grades, resulting in more coherent ore zones delivering more predictable head grades for treatment at a proximal processing plant. Recent inspection by GBM indicates that the portal, ventilation shafts and decline appear to be in sound condition with service lines intact.

The current reported 309 resource is based on data from 429 drill holes and assumes a combination of open pit and underground mining with cut-off grades of 0.4 and 2.0 g/t Au respectively. This resource contains 500,600 ounces and represents approximately 66% of the total 760,700 ounces of gold contained within the current Twin Hills mineral resources which only extends to around 400 metres below surface. While the modelled resource lies within a larger broadly mineralised breccia zone refining the mining and processing costs along with metallurgical recoveries will be important in understanding what additional resources could be extracted, it does serve to demonstrate the overall scale of this deposit. It is also only part of the Twin Hills mineralising system, much of which remains to be thoroughly tested along strike and at depth.

Photograph: Decline portal at the Twin Hills 309 Gold Mine.



Figure 5: Grade Tonnage Curve for the Twin Hills 309 Deposit. This graph clearly shows that there is a large volume of lower grade material that may become extractable should deeper open pit mining with a lower cut-off grade prove feasible

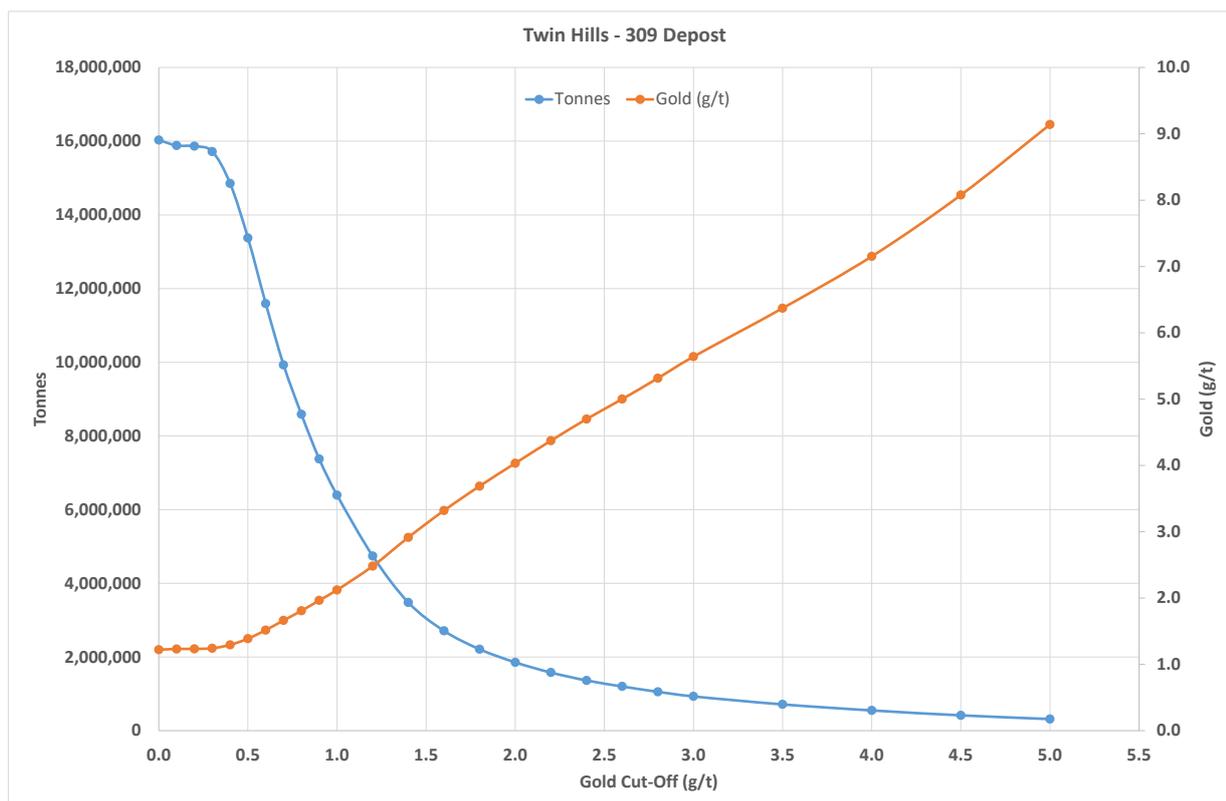
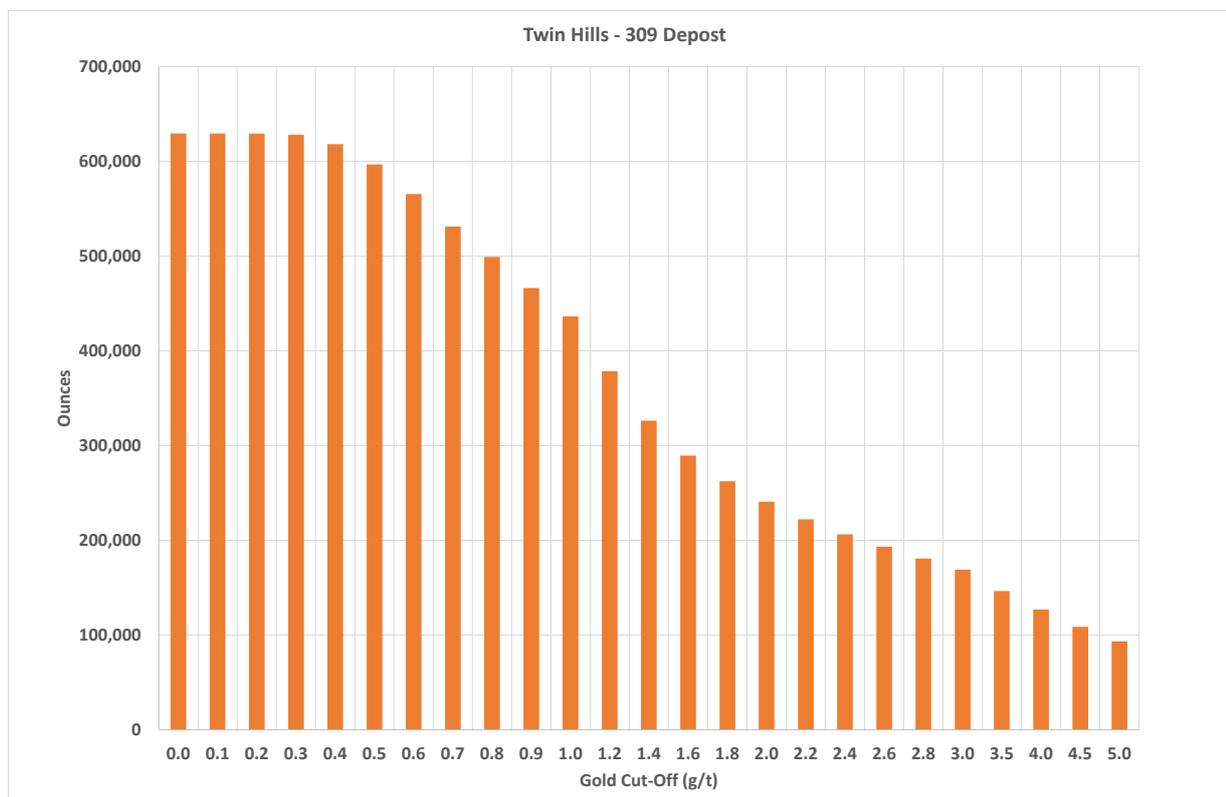


Figure 6: Contained gold for the Twin Hills 309 deposit versus cut-off grade



The Lone Sister deposit is located approximately six kilometres south of 309 and is considered part of the same auriferous Twin Hills mineralising system. Lone Sister appears as a more tightly constrained deposit, more closely related to a vein system, and was described in part as this by previous workers. This deposit has been modelled based on data from a total of 50 drillholes and on a 2.0 g/t Au cut-off grade assuming that underground mining is the likely mining method. This model also extends to approximately 400 metres below surface and, as for the 309 deposit, there is no drill testing below this depth.

Figure 7: Grade tonnage curve for the Lone Sister Gold Deposit.

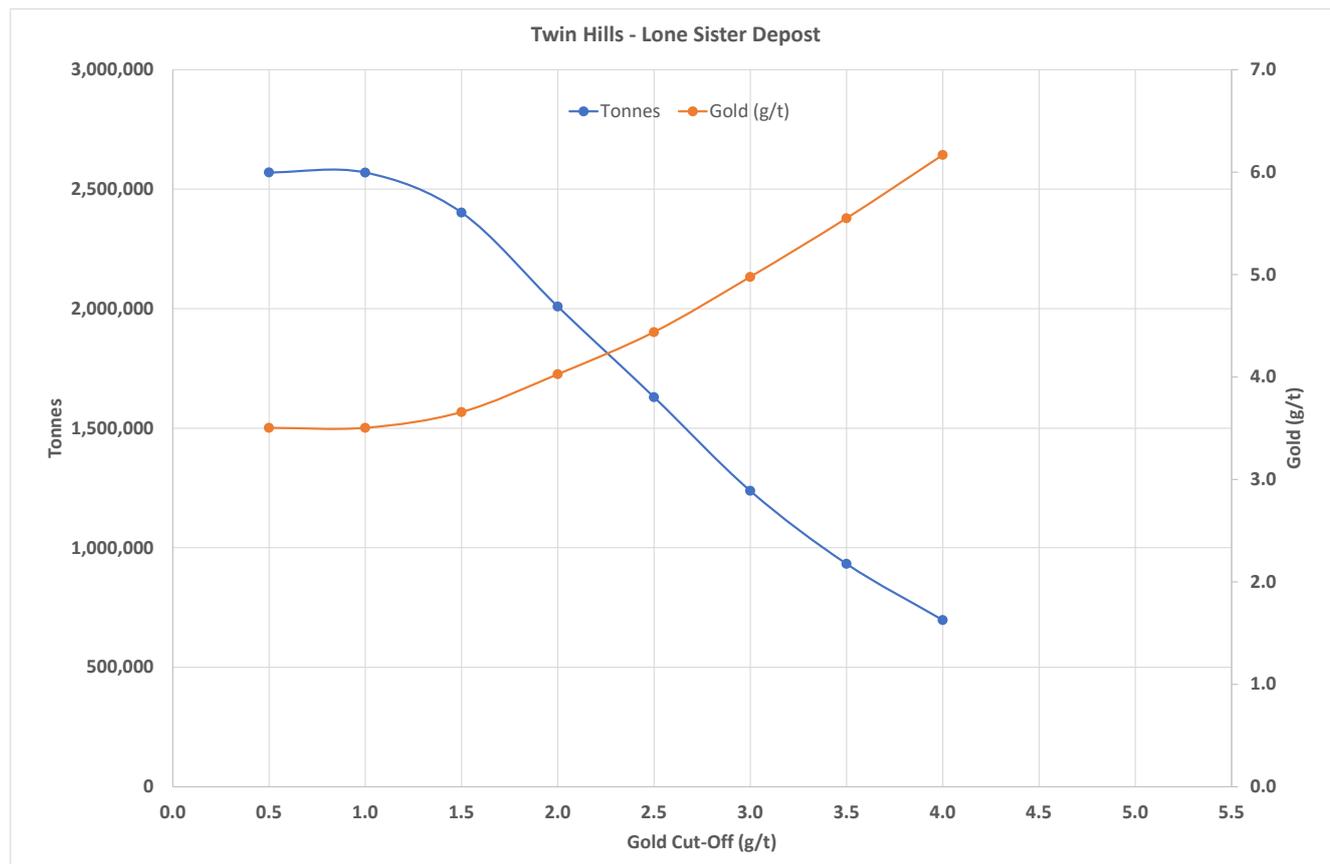
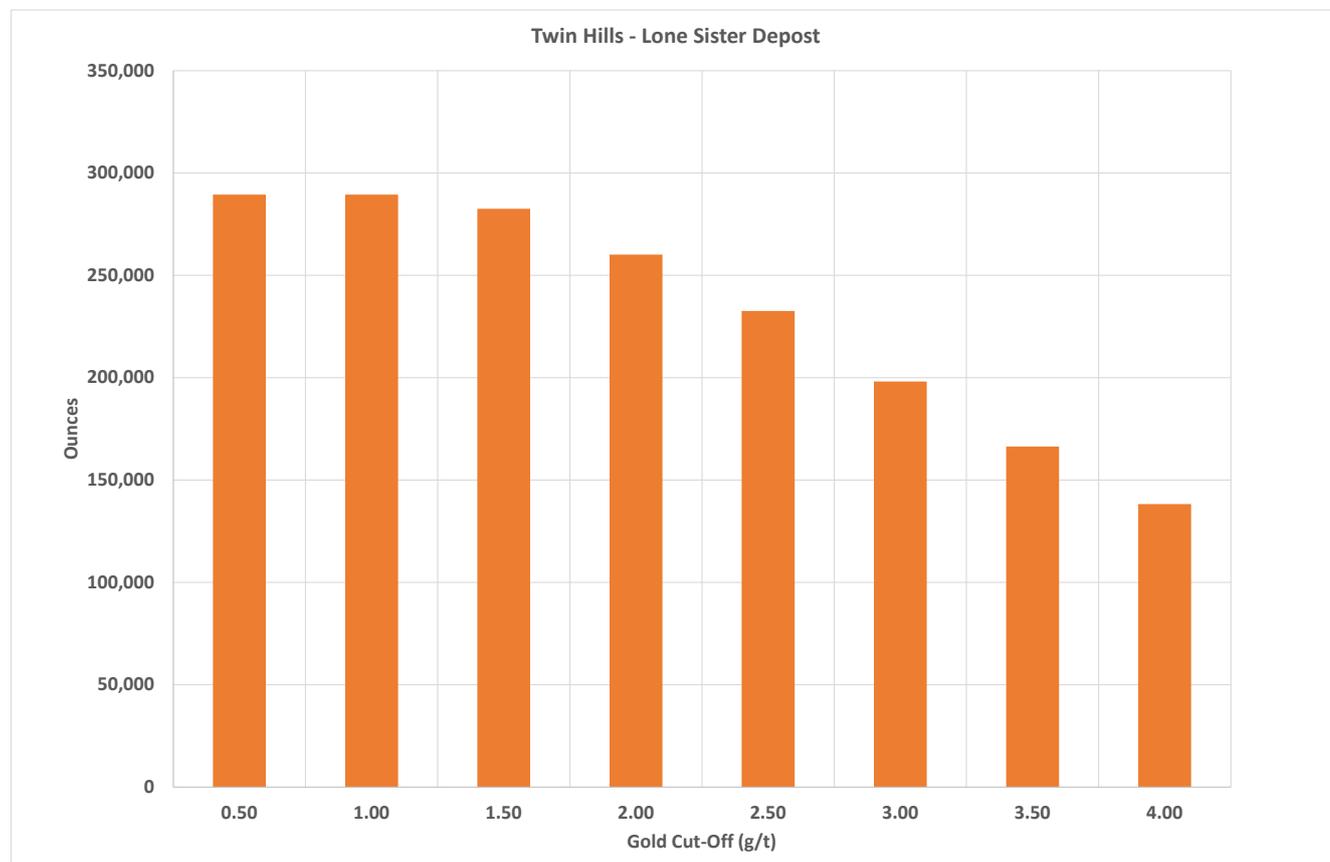


Figure 8: Contained gold versus cut-off grade for the Lone Sister Gold Deposit.



2018 Twin Hills Gold Project – 309 and Lone Sister Resource Estimates Commentary

Geology, Mineralisation and Exploration Potential

The Twin Hills 309 and Lone Sister deposits are contained within Mining Licence 70316 which is located within a broader tenement package being purchased by GBM within the western arm of the Drummond Basin (see Figure 1). Mineralisation in the Drummond Basin is characterised as high-grade epithermal style precious metal deposits. Examples include Pajingo (3.2 Moz), Wirralie (1.1 Moz), Yandan (0.6 Moz) and Koala. Mineralisation is typified by fine grained electrum in quartz veins and or breccias. These deposits are interpreted to have formed locally in extensional jogs or bends of transform fault systems.

309 Deposit

The northern and primary resource of the two advanced-stage prospects at Twin Hills and the only one with any mine development. 309 is an eruption breccia, intrusion-related low sulphidation epithermal gold-silver system with input from polyphase intrusions at depth. Sinters at surface indicate full system preservation. Local bonanza grades are present (e.g. 17 m @ 317.4 g/t Au from 222 m in TRCD728 including, 5 m @ 1036.6 g/t Au from 222 m, 4 m @ 49.03 g/t Au from 230 m) (Refer ASX: GBZ release 18 January 2019) and the deposit is zoned from silver-dominated near surface to gold-dominated with depth.

Three high grade mineralised zones were previously defined with Area 1 subjected to mining between 2005 and 2007 by BMA; Area 1 consists of chalcedonic silica-flooded breccia with fine disseminated electrum, Area 2 is a broad EW zone of narrow steeply south-dipping quartz-calcite veins carrying free gold, and Area 3 is a well-defined steeply north-dipping zone of more intense stockwork quartz-calcite veins. In general, 309 does not conform to a Pajingo-style “fat-vein” dilational epithermal model, more closely resembling an anastomosing or stockwork vein type. Current resource modelling has identified additional zones, some of which require further testing to lift the resource classification level.

Exploration potential at the 309 Deposit includes:

- Down-plunge depth extensions to identified ore shoots, including Area 2 (beyond 350 m below surface).
- Persistence of low-sulphidation Ginguero-style veins to 450 m concentrated in dilational shoots on the WNW-trending bounding Great Southern Fault – epithermal pressure-temp conditions may persist in fault dilational zones to significant depth.

Lone Sister Deposit

The deposit is located 7 km south of 309 and has seen no significant exploration activity since 2008. Mineralisation is interpreted as low-sulphide epithermal Au restricted to a porphyritic rhyolite dome host and an adjacent related hydrothermal breccia. Mineralisation extends to at least 350m below surface and is open at depth. The deposit location on flexures in through-going structures suggests mineralisation may continue to depth. Gold is present in two main styles; as sooty sulphide veinlets and as open-space banded mineralisation in the breccia.

Further exploration potential at Lone Sister:

- Intrusive source at depth for the mineralizing fluids in the breccia body
- Moderate to steep north-plunging shoots within the rhyolite unit remain open.

Drilling Techniques

This Resource estimate is based on diamond (DD) and reverse circulation (RC) drilling data compiled from previous exploration and mining activity.

Whilst the drilling was carried out under the supervision of several companies, the drilling, sampling and assaying methods varied little throughout the history of the project.

RC drilling used a face sample hammer to reduce downhole contamination. RC holes were rarely downhole surveyed. Surface diamond core drilling (HQ and NQ) was mostly tails off RC pre-collars. The diamond holes were surveyed at an average of 35m downhole.

Underground diamond drilling (BQ) was used for very closely spaced infill drilling prior to mining.

The 309 mineral resource estimate is based on 429 drill holes for 55,947.63 m. Of these 16 were diamond core holes drilled from surface totalling 2,459.1 m; 196 diamond core holes drilled from underground totalling 12,608.3 m; 111 diamond core tails of RC pre-collars totalling 29,528.31 m; 106 RC holes drilled from surface totalling 11,351.82 m.

The Lone Sister mineral resource estimate is based on 50 drill holes for 14,067.72 m. Of these 42 were diamond core tails of RC pre-collars totalling 13,260.72 m; 8 were RC holes drilled from surface totalling 807.0 m.

Diamond core recovery was very high, averaging 99.7% at 309 and 100% at Lone Sister.

Sampling Methods

RC drilling drill cuttings were sampled from the cyclone at 1.0 m intervals and sub-sampled using Jones riffle splitters to a 2 kg – 3 kg sample.

Diamond drill core was sub-sampled by cutting the core in half longitudinally using a diamond saw. The core was cut at the highest angle possible to geological features to ensure that half of each geological feature was sampled. Diamond core samples were generally to 1.0 m

Sample Analysis Method

All samples were assayed for Au by fire assay with aqua regia / AAS finish. Approximately 65% of samples were also assayed for Ag using the same method.

Assay quality control procedures varied through time with different operators. In general blanks and standards were inserted at a rate of 1 in 10 – 20 and pulp duplicates at 1 in 15 samples. 50% of samples > 0.5 g/t Au were sent to an umpire laboratory. The results of these data indicate that the quality of the data is suitable for use in resource estimation.

Estimation Methodology

The raw gold assay results were composited to 2.0 m prior to statistical analysis and variography.

For the 309 deposit gold grades were interpolated in a block model with parent blocks 5 m by 5 m by 5 m sub-blocked to 1.25 m x 1.25 m x 1.25 m Interpolation was by ordinary kriging within 2 variably oriented gold grade domains interpreted at a nominal 0.2 g/t Au. The gold grade domains were used as hard boundaries.

At the Lone Sister deposit gold grades were interpolated in a block model with parent blocks 4 m x 16 m x 10 m and sub-blocks of 1 m x 4 m x 12.5 m. Interpolation was by ordinary kriging within a gold grade domain interpreted at a nominal 1.0 g/t Au. The gold grade domain was used as a hard boundary. Grades above 60 g/t Au were restricted to within 16 m.

Oxidation domains were interpreted from logged oxidation and used to code the block model for determination of mineralisation types and assignment of bulk density.

Limited density is available so assumed densities (based on typical values for lithology and oxidation level) were assigned to blocks. All oxide material was assigned a density of 2.4 t/m³ and 2.6 t/m³ in fresh material.

Resource Classification Criteria

The 309 block model was classified in accordance with the JORC 2012 code.

Resource classification took into account:

- geological continuity
- the plausibility of alternative geological interpretations,
- data (drilling) density and configuration (distance to nearest samples, number holes used)
- kriging slope of regression

Resource classification was based on largely on measures of grade estimation quality. The low grade and simple shape of the gold grade domain meant that geological continuity was not considered for resource classification because the level of confidence in geological continuity is uniform throughout the model.

Measured resources have been classified from continuous zones where the average distance of composite samples used to interpolate a block was less than 15 m, the kriging slope of regression was greater than 0.7 and data from at least three drillholes was used.

Indicated resources were classified from continuous zones where the average distance of composite samples used to interpolate a block was 15 m – 25 m, the kriging slope of regression was 0.1 - 0.7 and data from at least two drillholes was used.

All remaining material inside the gold grade domains was classified as inferred.

All of the Lone Sister block model was classified as inferred as the geological interpretation is not unequivocal.

Validation of the block models was by:

- comparison to reported production of the 309 deposit to closure in early 2007 which totalled 75,848 t @ 10.0 g/t Au. This resource estimate reports 76,000 t at 6.8 g/t Au from the mined voids.
- visually, by comparison of block model grades to de-clustered composite grades,
- by comparison of histograms of block and composite grades and in swath plots.

Cut-off Grades

The resource has been estimated at a range of cut-off grades which are presented as grade and tonnage curves in Figures 5, 6, 7 & 8. Potentially open pit resources at 309 were reported from above 1050 RL at a cut-off of 0.4 g/t Au. The cut-off grade was revised to 0.4 g/t Au for the potential open pit resources (from 1.0 g/t Au, used in 2019) due to the significantly higher current gold price of AUD\$2,500/oz. Further metallurgical testwork on low grade samples is required to improve confidence in this area as the project advances, along with understanding the impact on costs of a larger plant fed from GBM's three resource areas. The previous pit optimisation at a gold price of AUD\$1,500/oz extended to 1050 RL. Work will soon start on new pit optimisations to explore how deep a possible pit might push at current gold prices. This is likely to result in less Underground resources and a larger potential open pit resource.

Underground resources at 309 and Lone Sister are reported from below 1050 RL at a 2.0 g/t cut-off. The cut-off grade reflects previous underground mining studies with an allowance for reasonably foreseeable gold price increases. Following planned drilling in 2022 the underground resources will be re-evaluated at lower cut-off grades.

Mining and Metallurgical Methods.

This Resource estimate is based on the following assumptions, that:

- Open pit mining is technically feasible at the 309 deposit. This is supported by previous mining studies and preliminary pit optimisations. A previous pit optimisation at a gold price of AUD\$1,500/oz extended to 1050 RL. Open pit mining was not considered an option at the Lone Sister Deposit, however future studies may demonstrate that this is possible as mineralisation does extend to near surface.
- Underground mining is supported by previous mining at the 309 orebody and later feasibility studies demonstrating that this is possible and feasible.
- An economic processing route will be found. Previous mining was conducted using conventional CIL treatment methodologies, however sulphide flotation and leaching of the concentrate might provide an alternative along with integration into a possible centralised processing plant. Significant amount of metallurgical testwork has been completed by previous operators and reviewed by GBM and supports this view.
- In a reasonable timeframe to development, the gold price increases in line with long term trends.

This ASX announcement was approved and authorised for release by:

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About GBM Resources

GBM Resources Limited (ASX: GBZ) is a well-funded Queensland based mineral exploration and development company focused on the discovery of world-class gold and copper deposits in Eastern Australia. The company has a high calibre project portfolio, hosting district scale mineral systems, located in a number of premier metallogenic terrains.

Its 100% owned flagship project in the Drummond Basin (QLD) holds ~1.6 Moz of gold in JORC resources (Mt Coolon, Yandan and Twin Hills). 2022 will see an expanded drilling program which is aiming to define 2-3 Moz and support GBM's transition into a mid-tier Australian gold company.

Separately it also holds tenements in the Mt Morgan district (subject to a vend into a TSX company) and in the Mt Isa Inlier in Queensland (JV with Nippon Mining Australia - 54%), and the Malmsbury Project (JV with Novo Resources Corp. - 50%, earning additional 10%) in the prolific Victorian Goldfields. This is complemented by the cash generating White Dam Gold-Copper Project in South Australia in which GBM now holds a 100% interest. Divestment of non-core assets will continue.

COMPETENT PERSON STATEMENT

The information in this report that relates to The Twin Hills Mineral Resources is based on information compiled by Kerrin Allwood, who is a Member of The Australasian Institute of Mining and Metallurgy and The Australasian Institute of Geoscientists. Mr Allwood is a full time employee of Geomodelling Limited. Mr Allwood has sufficient experience which is relevant to the style of mineralisation and type of deposit under consideration and to the activity which he is undertaking to qualify as a Competent Person as defined in the 2012 Edition of the 'Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves'. Mr Allwood consents to the inclusion in the report of the matters based on his information in the form and context in which it appears.

The Company confirms that it is not aware of any new information or data that materially affects the information included in the respective announcements and all material assumptions and technical parameters underpinning the resource estimates within those announcements continue to apply and have not materially changed.

The Company confirms that the form and context in which the Competent Persons findings are presented have not been materially modified from the original market announcements.

References

1. GBM Resources. (2019, January 18). *Mt Coolon and Twin Hills Combined Resource Base Approaches 1 Moz*. ASX:GBZ ASX Release.
2. GBM Resources (2020, 23 December). *Mt Coolon and Yandan Combined Resources Total 852,000 oz, Following Completion of Yandan Acquisition*. ASX:GBZ ASX Release.
3. Corbett, G. (2006). *Comments on Geology and Exploration of the Twin Hills Gold Project, QLD, Australia*. Internal report for BMA Gold
4. Sennitt, C.M. (1991). *Aspects of Epithermal Gold Mineralisation, Twin Hills, QLD*. MSc thesis, James Cook University (unpub.).
5. Alston, A.J., Hartley, J.S., Sennitt, C.M., 1991. *The Geology of the Twin Hills Epithermal Gold Deposit, Queensland*. World Gold '91, Cairns, pp. 331-339.
6. King, S. (1999). *Structural Controls on Gold Mineralisation at the Twin Hills Project, Drummond Basin, QLD*. Internal Report for Homestake Gold.

Appendices

- 1 *GBM Mineral Resource Estimates For the Drummond Basin Projects (Mt Coolon, Yandan and Twin Hills) along with other project interests*
- 2 *309 Deposit and Lone Sister Deposit JORC Table 1*

Refer ASX: GBZ release 18 January 2019 - Mt Coolon and Twin Hills Resource Base Approaches 1Moz, for the following:

- 309 Deposit Surface Drillhole Collar Table*
- Lone Sister Deposit Surface Drillhole Collar Table*
- 309 Deposit Surface Drilling Downhole Intersection Summary Table*
- Lone Sister Surface Drilling Downhole Intersection Summary Table*
- 309 Surface Drillhole Location Plan*
- Lone Sister Surface Drillhole Location Plan.*

APPENDIX 1: GBM Mineral Resource Estimate For the Drummond Basin Projects (Mt Coolon, Yandan and Twin Hills) along with other project interests

Deposit	Resource Category									Total			Cut-off
	Measured			Indicated			Inferred			000' t	Au g/t	Au oz	
	000' t	Au g/t	Au oz	000' t	Au g/t	Au oz	000' t	Au g/t	Au oz	000' t	Au g/t	Au oz	
Koala													
Open Pit				670	2.6	55,100	440	1.9	26,700	1,120	2.3	81,800	0.4
UG Extension				50	3.2	5,300	260	4	34,400	320	3.9	39,700	2.0
Tailings	114	1.7	6,200	9	1.6	400				124	1.6	6,600	1.0
Sub Total	114	1.7	6,200	729	2.6	60,800	700	2.7	61,100	1,563	2.5	128,100	
Eugenia													
Oxide - Open Pit				885	1.1	32,400	597	1.0	19,300	1,482	1.1	51,700	0.4
Sulphide - Open Pit				905	1.2	33,500	1,042	1.2	38,900	1,947	1.2	72,400	0.4
Sub Total				1,790	1.1	65,900	1,639	1.1	58,200	3,430	1.1	124,100	
Glen Eva													
Sub Total - Open Pit				1,070	1.6	55,200	580	1.2	23,100	1,660	1.5	78,300	0.4
Yandan													
East Hill - Open Pit							20,600	0.8	505,000	20,060	0.8	505,000	0.3
South Hill - Open Pit							900	0.6	16,000	900	0.6	16,000	0.3
Sub Total							21,500	0.8	521,000	21,500	0.8	521,000	
Twin Hills													
309 - Open Pit	586	2.7	50,300	5,470	1.4	253,200	4,165	0.9	120,200	10,220	1.3	423,700	0.4
309 - UG				110	4.8	16,800	510	3.7	60,100	620	3.9	76,900	2.0
Lone Sister - UG							2,010	4.0	260,100	2,010	4.0	260,100	2.0
Sub Total	586	2.7	50,300	5,580	1.5	270,000	6,685	2.0	440,400	12,850	1.8	760,700	
Drummond Basin Total	700	2.5	56,500	9,169	1.5	451,900	31,104	1.1	1,103,800	41,003	1.2	1,612,200	
White Dam													
Hannaford - Open Pit				700	0.7	16,400	1,000	0.8	26,900	1,700	0.8	43,300	0.2
Vertigo - Open Pit				300	1.0	9,400	1,400	0.6	29,000	1,700	0.7	38,400	0.2
White Dam North - Open Pit				200	0.5	2,800	1,000	0.6	17,600	1,200	0.5	20,400	0.2
Sub Total				1,200	0.7	28,600	3,400	0.7	73,500	4,600	0.7	101,900	
cut-off grade is 0.20 g/t Au for all, Vertigo is restricted to above 150RL (~70m below surface)													
Malmsbury													
Sub Total - UG							820	4.0	104,000	820	4.0	104,000	2.5
Sub Total - UG - GBM Share							410	4.0	52,000	410	4.0	52,000	2.5
GBM Total													1,766,100

The announcements containing the Table 1 Checklists of Assessment and Reporting Criteria relating to the 2012 JORC compliant Resources are:

- Koala/Glen Eva and Eugenia – GBM ASX Announcements, 4 December 2017, Mt Coolon Gold Project Scoping Study
- Yandan – GBM ASX Announcement, 23 December 2020, Mt Coolon and Yandan Combined Resources Total 852,000 oz, following completion of Yandan acquisition
- Twin Hills – GBM ASX Announcement, 18 January 2019, Mount Coolon and Twin Hills Combined Resource Base Approaches 1 Million Ounces
- White Dam - GBM ASX Announcement, 18 August 2020, White Dam Maiden JORC 2012 Resource of 102 koz
- Malmsbury – GBM ASX Announcement, 4 July 2019, Malmsbury Resource Upgraded to JORC 2012
- Including this announcement

- a) The preceding statements of Mineral Resources conforms to the “Australasian Code for Reporting Exploration Results, Mineral Resources and Ore Reserves (JORC Code) 2012 Edition”
- b) All tonnages are dry metric tonnes
- c) Data is rounded to ('000 tonnes, 0.0 g/t and '000 ounces). Discrepancies in totals may occur due to rounding
- d) Resources have been reported as both open pit and underground with varying cut-off based off several factors as discussed in the corresponding Table 1 which can be found with the original ASX announcement for each Resources.

APPENDIX 2 – Table 1

JORC CODE, 2012 EDITION – TABLE 1 REPORT FOR TWIN HILLS PROJECT, 309 AND LONE SISTER DEPOSITS

a. Section 1 Sampling Techniques and Data

(Criteria in this section apply to all succeeding sections.)

Criteria	JORC Code explanation	Commentary
Sampling techniques	<ul style="list-style-type: none"> <i>Nature and quality of sampling (e.g. cut channels, random chips, or specific specialised industry standard measurement tools appropriate to the minerals under investigation, such as down hole gamma sondes, or handheld XRF instruments, etc.). These examples should not be taken as limiting the broad meaning of sampling.</i> <i>Include reference to measures taken to ensure sample representivity and the appropriate calibration of any measurement tools or systems used.</i> <i>Aspects of the determination of mineralisation that are Material to the Public Report.</i> <i>In cases where 'industry standard' work has been done this would be relatively simple (e.g. 'reverse circulation drilling was used to obtain 1 m samples from which 3 kg was pulverised to produce a 30 g charge for fire assay'). In other cases more explanation may be required, such as where there is coarse gold that has inherent sampling problems. Unusual commodities or mineralisation types (e.g. submarine nodules) may warrant disclosure of detailed information.</i> 	<ul style="list-style-type: none"> These mineral resource estimates are based on samples from reverse circulation (RC) and diamond core (DD) drilling. RC drill cuttings were sampled from the cyclone at 1.0 m intervals and sub-sampled using Jones riffle splitters which are designed to allow the collection of unbiased sub-samples. DD core was sub-sampled by cutting the core in half longitudinally using a diamond saw. The core was cut at the highest angle possible to geological features to ensure that half of each geological feature was sampled. Diamond core samples were generally to 1.0 m intervals but honoured geological contacts where appropriate. All sub-samples were then bagged and dispatched to external commercial laboratories for assay. All samples were analysed for gold by fire assay followed by aqua regia digestion and AAS analysis. Silver was similarly assayed by fire assay / AAS for 68% of the samples and arsenic by aqua regia digest and AAS analysis for 50% of the samples. Selected samples were analysed for a multi-element suite.
Drilling techniques	<ul style="list-style-type: none"> <i>Drill type (e.g. core, reverse circulation, open-hole hammer, rotary air blast, auger, Bangka, sonic, etc.) and details (e.g. core diameter, triple or standard tube, depth of diamond tails, face-sampling bit or other type, whether core is oriented and if so, by what method, etc.).</i> 	<p>309 mineral resource estimate:</p> <ul style="list-style-type: none"> The 309 mineral resource estimate is based on 429 drill holes for 55,947.63 m. Of these 16 were diamond core holes drilled from surface totaling 2,459.1 m; 196 diamond core holes drilled from underground totaling 12,608.3 m; 111 diamond core tails of RC pre-collars totaling 29,528.31 m; 106 RC holes drilled from surface totaling 11,351.82 m. All RC drilling utilized a face sample hammer. DD core was HQ and NQ in size for surface drilling and BQ for underground drilling. <p>Lone Sister mineral resource estimate:</p> <ul style="list-style-type: none"> The Lone Sister mineral resource estimate is based on 50 drill holes for 14,067.72 m. Of these 42 were diamond core tails of RC pre-collars totaling 13,260.72 m; 8 were RC holes drilled from surface

Criteria	JORC Code explanation	Commentary
		<p>totaling 807.0 m.</p> <ul style="list-style-type: none"> All RC drilling utilized a 5.25 inch face sample hammer. DD core was HQ and NQ in size.
<i>Drill sample recovery</i>	<ul style="list-style-type: none"> <i>Method of recording and assessing core and chip sample recoveries and results assessed.</i> <i>Measures taken to maximise sample recovery and ensure representative nature of the samples.</i> <i>Whether a relationship exists between sample recovery and grade and whether sample bias may have occurred due to preferential loss/gain of fine/coarse material.</i> 	<ul style="list-style-type: none"> RC drilling recovery was not recorded, previous explorers did not note that there was any significant issue at the time of drilling. Raw assays and intercept calculations presented have not be adjusted or omitted for poor recoveries. Diamond drilling recovery was measured run by run as recovered length compared to drilled length. For the 309 mineral resource estimate diamond drilling recovery is available for 44 holes and averages 99.7% For the Lone Sister mineral resource estimate diamond drilling recovery is available for 9 holes and averages 100.0% Any potential relationship between drilling recovery and gold grade was not investigated because the diamond drilling run recovery is so high in the available data.
<i>Logging</i>	<ul style="list-style-type: none"> <i>Whether core and chip samples have been geologically and geotechnically logged to a level of detail to support appropriate Mineral Resource estimation, mining studies and metallurgical studies.</i> <i>Whether logging is qualitative or quantitative in nature. Core (or costean, channel, etc.) photography.</i> <i>The total length and percentage of the relevant intersections logged.</i> 	<ul style="list-style-type: none"> All drill core and chips were logged for lithology, colour, weathering and alteration using standardized codes. Selected diamond core was also logged for geotechnical data (RQD, strength, fracture frequency, joint type and roughness) All intersections were logged.
<i>Sub-sampling techniques and sample preparation</i>	<ul style="list-style-type: none"> <i>If core, whether cut or sawn and whether quarter, half or all core taken.</i> <i>If non-core, whether riffled, tube sampled, rotary split, etc. and whether sampled wet or dry.</i> <i>For all sample types, the nature, quality and appropriateness of the sample preparation technique.</i> <i>Quality control procedures adopted for all sub-sampling stages to maximise representivity of samples.</i> <i>Measures taken to ensure that the sampling is representative of the in situ material collected, including for instance results for field duplicate/second-half sampling.</i> <i>Whether sample sizes are appropriate to the grain size of the material</i> 	<ul style="list-style-type: none"> RC drill cuttings were sub-sampled using 3 tier (12.5%) Jones riffle splitters to yield a 2 kg – 3 kg sub-sample. RC sample moisture was not recorded and any measures taken sample wet or moist RC samples have not been recorded. DD core was sub-sampled by cutting the core in half longitudinally using a diamond saw. The core was cut at the highest angle possible to geological features to ensure that half of each geological feature was sampled. Diamond core samples were generally to 1.0 m intervals but honoured geological contacts where appropriate. Very rare visible gold (<1.0 mm) is present at both deposits. Whilst no analysis of the optimal sample size for such material was undertaken it is likely that the sample size is insufficient for a reliable result. The

Criteria	JORC Code explanation	Commentary
	<p><i>being sampled.</i></p>	<p>impact of these samples on the resource estimates was managed by the restriction of outliers during interpolation (see below).</p>
<p><i>Quality of assay data and laboratory tests</i></p>	<ul style="list-style-type: none"> <i>The nature, quality and appropriateness of the assaying and laboratory procedures used and whether the technique is considered partial or total.</i> <i>For geophysical tools, spectrometers, handheld XRF instruments, etc., the parameters used in determining the analysis including instrument make and model, reading times, calibrations factors applied and their derivation, etc..</i> <i>Nature of quality control procedures adopted (e.g. standards, blanks, duplicates, external laboratory checks) and whether acceptable levels of accuracy (i.e. lack of bias) and precision have been established.</i> 	<ul style="list-style-type: none"> The assay methods used are total and appropriate to the style of mineralization. No geophysical methods were used Assay quality control procedures varied through time with different operators. In general blanks and standards were inserted at a rate of 1 in 10 – 20 and pulp duplicates at 1 in 15 samples. 50% of samples > 0.5 g/t Au were sent to an umpire laboratory. The results of these data indicate that the quality of the data is suitable for use in resource estimation.
<p><i>Verification of sampling and assaying</i></p>	<ul style="list-style-type: none"> <i>The verification of significant intersections by either independent or alternative company personnel.</i> <i>The use of twinned holes.</i> <i>Documentation of primary data, data entry procedures, data verification, data storage (physical and electronic) protocols.</i> <i>Discuss any adjustment to assay data.</i> 	<ul style="list-style-type: none"> GBM have not carried out any check assays Twinned holes were not drilled, however the underground drilling at the 309 deposit resulted in very closely spaced (< 1.0 m) drilling. These holes showed that both gold grade and veining is highly variable at less than 10 m scale but form continuous zones at 10 m – 100 m scale. The raw assay data (laboratory certificates) was available for approximately 80% of the data. Negative values in the database less than -0.1 g/t Au were treated as null values (not sampled), negative values between 0 and -0.1 g/t Au were halved and converted to positive values on the assumption that these were below detection values.
<p><i>Location of data points</i></p>	<ul style="list-style-type: none"> <i>Accuracy and quality of surveys used to locate drill holes (collar and down-hole surveys), trenches, mine workings and other locations used in Mineral Resource estimation.</i> <i>Specification of the grid system used.</i> <i>Quality and adequacy of topographic control.</i> 	<ul style="list-style-type: none"> Surface drill collar locations were determined by differential GPS (DGPS) to +/- 0.1 m. Underground drill collar locations were determined by the mine surveyor (total station instrument). Downhole surveys were taken at an average of 35 m spacing downhole in DD and RC DD holes. Most RC holes were not surveyed down hole Topographic control in the block models are from triangulated 2.0 m contours created from surface survey traverses. The topographic surfaces are suitable for resource estimation. Underground voids at the 309 deposit are from wireframes created from underground survey data (development) and laser scans

Criteria	JORC Code explanation	Commentary
		<p>(stopes). These are suitable for resource estimation.</p> <ul style="list-style-type: none"> All locational data was originally acquired local grids. GBM used MapInfo software to convert all locational data (including historical wireframes) to MGA grid.
<p><i>Data spacing and distribution</i></p>	<ul style="list-style-type: none"> <i>Data spacing for reporting of Exploration Results.</i> <i>Whether the data spacing and distribution is sufficient to establish the degree of geological and grade continuity appropriate for the Mineral Resource and Ore Reserve estimation procedure(s) and classifications applied.</i> <i>Whether sample compositing has been applied.</i> 	<ul style="list-style-type: none"> Drill data spacing varies significantly from less than 1.0 metre near underground drill sites at the 309 deposit to 40 m in areas defined by widely spaced surface drilling. Underground drilling at 309 is on varying azimuths and dips. Surface drilling at 309 is on 25 m spaced east-west and 25m spaced north – south sections Surface drilling at Lone Sister is on 40 m spaced east-west sections and largely drilled towards the east at dips of -60° to -80°. All resource estimation work has been completed in MGA zone 55 using the GDA94 datum but with 1,000 m added to elevation to prevent negative elevations.
<p><i>Orientation of data in relation to geological structure</i></p>	<ul style="list-style-type: none"> <i>Whether the orientation of sampling achieves unbiased sampling of possible structures and the extent to which this is known, considering the deposit type.</i> <i>If the relationship between the drilling orientation and the orientation of key mineralised structures is considered to have introduced a sampling bias, this should be assessed and reported if material.</i> 	<ul style="list-style-type: none"> The gross geometry of mineralisation at 309 is largely isotropic. The varying drilling orientations used at 309 has allowed definition of this geometry and does not introduce any known bias The Lone Sister mineralization is more tabular in nature, striking north – south and dipping steeply to the west. The drilling towards the east is appropriate to define the geology of the mineralization and does not introduce any known bias
<p><i>Sample security</i></p>	<ul style="list-style-type: none"> <i>The measures taken to ensure sample security.</i> 	<ul style="list-style-type: none"> Any measures taken to ensure sample security from the site to the assay laboratories have not been recorded.
<p><i>Audits or reviews</i></p>	<ul style="list-style-type: none"> <i>The results of any audits or reviews of sampling techniques and data.</i> 	<ul style="list-style-type: none"> The competent person is not aware of any audits having been carried out on the data used in this resource estimate. The data used in this resource estimate have been reviewed several times during BMA mining and for various due diligence studies carried out when the project has changed hands.

b. Section 2 Reporting of Exploration Results

(Criteria listed in the preceding section also apply to this section.)

Criteria	JORC Code explanation	Commentary
<p><i>Mineral tenement and land tenure status</i></p>	<ul style="list-style-type: none"> <i>Type, reference name/number, location and ownership including agreements or material issues with third parties such as joint ventures, partnerships, overriding royalties, native title interests, historical sites, wilderness or national park and environmental settings.</i> <i>The security of the tenure held at the time of reporting along with any known impediments to obtaining a licence to operate in the area.</i> 	<ul style="list-style-type: none"> Twin Hills 309 and Lone Sister deposits are contained within current Mining Licence ML70316, expiry 31/12/2019. The license is jointly owned by Minjar Gold Pty Ltd through subsidiary companies NQM Gold 2 Pty Ltd (60%) and CQT Gold Australia Pty Ltd (40%) and is subject to a Sale and Purchase agreement with Mount Coolon Gold Mines Pty Ltd, a wholly owned subsidiary of GBM Resources Ltd. On completion of the purchase, royalties on gold production will be to the Queensland Government (currently 5% on all ML's in the state of QLD) and a 2.5% to Franco –Nevada Australia Pty Ltd. Consent has been obtained for the transfer Conquest Mining Ltd. From its parent company, Evolution Mining Ltd. Environmental Authority EPML00772013 is current and the Financial Assurance (now ERC) held by the Queensland Department of Environment and Science is currently AUD\$1,475,156. and will be subject to the recently lodged Progressive Rehabilitation Closure Plan (PRCP) The licence is subject to an ILUA with the Jangaa People. The NW corner of the licence falls within a Strategic Cropping Zone and the licence is contained within a Forest Management Area. There are no known impediments to future mining on this Licence.
<p><i>Exploration done by other parties</i></p>	<ul style="list-style-type: none"> <i>Acknowledgment and appraisal of exploration by other parties.</i> 	<ul style="list-style-type: none"> Exploration has been carried out by several companies over a long period of time at Twin Hills. Gold mineralisation was first recognized at Twin Hills by Metana Minerals NL in 1987. Since that time the project area has been held under either an exploration of mining licence by a variety of companies and joint ventures. BMA Gold commenced underground mining at 309 in January 2006 and ceased mining in February 2007. Of the drilling data used to inform the 309 mineral resource estimate Metana drilled 1 DD hole (120m) and 1 RC hole (89m) in 1988, Plutonic drill 31 RCDD holes (8555.41m) and 53 RC holes (5,197.4m) from 1994 to 1999 and BMA Gold drilled 15 surface DD holes (2,339.1m), 80 RCDD holes (2,0973m), 52 RC holes (6,065.42m) and 196 underground DD holes (12,608.3m) from 2002 until 2007. At the Lone Sister deposit Metana drilled 1 RCDD hole (435.5m) and

Criteria	JORC Code explanation	Commentary
		<p>2 RC holes (200m) in 1988, Plutonic drilled 15 RCDD holes (5,134.99m) and 1 RC hole (93m) from 1988 to 1997 and in 2006, Homestake Gold drilled 4 RCDD holes (1,379.33m) from 1998-1999) and BMA Gold 22 RCDD holes (6,310.9m) and 5 RC holes (514m) from 2004 to 2007.</p> <ul style="list-style-type: none"> • The Twin Hills project area has also been subject to aerial magnetic and radiometric surveys, soil geochemistry, RAB geochemistry and IP surveys. • The mineral resource estimates reported on here are based on the appropriately validated results of work completed by the above companies. • GBM have not completed any significant work at Twin Hills other than site inspections to review stored drill core, site geology, site infrastructure and access.
Geology	<ul style="list-style-type: none"> • <i>Deposit type, geological setting and style of mineralisation.</i> 	<ul style="list-style-type: none"> • The Twin Hills deposits are situated within the western domain of the Upper Devonian to Lower Carboniferous Drummond Basin, host to a number of epithermal gold deposits including the Pajingo deposit (3.2 Moz production to date). • The 309 deposit comprises a stockwork of very high grade, narrow (0.2 m) low sulphidation epithermal quartz-sulphide veins hosted in variably altered and mineralized breccias. The breccias comprise dominantly shale clasts in a very fine grained matrix. The overall geometry of gold mineralization at 309 is a steeply plunging body and is open at depth. The epithermal quartz veins form sheeted vein sets that strike north and locally vary in dip from sub-vertical to gently east dipping. Minor fluorite occurrences in associated with open space comb quartz suggest a significant magmatic component to the vein forming fluids. • The Lone Sister deposit is a more typical low sulphidation epithermal gold deposit. Gold mineralization is host by low grade quartz veins and very high grade quartz-sulphide veins. The gold mineralisation occurs in altered rhyolite. The quartz veins strike north south and dip 50° to 80° to the west.
Drill hole Information	<ul style="list-style-type: none"> • <i>A summary of all information material to the understanding of the exploration results including a tabulation of the following information for all Material drill holes:</i> 	<ul style="list-style-type: none"> • Drill hole collar and intercept tables are previously reported, (Refer ASX: GBZ release 18 January 2019)

Criteria	JORC Code explanation	Commentary
	<ul style="list-style-type: none"> ○ easting and northing of the drill hole collar ○ elevation or RL (Reduced Level – elevation above sea level in metres) of the drill hole collar ○ dip and azimuth of the hole ○ down hole length and interception depth ○ hole length. ● If the exclusion of this information is justified on the basis that the information is not Material and this exclusion does not detract from the understanding of the report, the Competent Person should clearly explain why this is the case. 	
Data aggregation methods	<ul style="list-style-type: none"> ● In reporting Exploration Results, weighting averaging techniques, maximum and/or minimum grade truncations (e.g. cutting of high grades) and cut-off grades are usually Material and should be stated. ● Where aggregate intercepts incorporate short lengths of high grade results and longer lengths of low grade results, the procedure used for such aggregation should be stated and some typical examples of such aggregations should be shown in detail. ● The assumptions used for any reporting of metal equivalent values should be clearly stated. 	<ul style="list-style-type: none"> ● All drilling intercepts from past exploration above 20 gram.metres (grade x length) have been reported. (Refer ASX: GBZ release 18 January 2019) Intervals are down hole length weighted average grades above 1 g/t Au incorporating up to 2 metres of lower grade material or internal dilution. Grades have not been cut as cutting of high grades in this style of mineralisation with a significant high grade component. ● No metal equivalents are reported.
Relationship between mineralisation widths and intercept lengths	<ul style="list-style-type: none"> ● These relationships are particularly important in the reporting of Exploration Results. ● If the geometry of the mineralisation with respect to the drill hole angle is known, its nature should be reported. ● If it is not known and only the down hole lengths are reported, there should be a clear statement to this effect (e.g. 'down hole length, true width not known'). 	<ul style="list-style-type: none"> ● Drilling is generally oriented perpendicular to the strike of the mineralisation at angles varying from acute to perpendicular. However only downhole intersections have been reported due to the variety of drill orientations and volume of drilling, the mature nature of the deposit with a range of drilling orientations.
Diagrams	<ul style="list-style-type: none"> ● Appropriate maps and sections (with scales) and tabulations of intercepts should be included for any significant discovery being reported These should include, but not be limited to a plan view of drill hole collar locations and appropriate sectional views. 	<ul style="list-style-type: none"> ● Appropriate images are included within the text of the release along with those in the original release (Refer ASX: GBZ release 18 January 2019).
Balanced reporting	<ul style="list-style-type: none"> ● Where comprehensive reporting of all Exploration Results is not practicable, representative reporting of both low and high grades and/or widths should be practiced to avoid misleading reporting of Exploration Results. 	<ul style="list-style-type: none"> ● Refer ASX: GBZ release 18 January 2019 for tables of intercepts.
Other substantive	<ul style="list-style-type: none"> ● Other exploration data, if meaningful and material, should be reported including (but not limited to): geological observations; geophysical 	<ul style="list-style-type: none"> ● No exploration results are reported.

Criteria	JORC Code explanation	Commentary
<i>exploration data</i>	<i>survey results; geochemical survey results; bulk samples – size and method of treatment; metallurgical test results; bulk density, groundwater, geotechnical and rock characteristics; potential deleterious or contaminating substances.</i>	
<i>Further work</i>	<ul style="list-style-type: none"> <i>The nature and scale of planned further work (e.g. tests for lateral extensions or depth extensions or large-scale step-out drilling).</i> <i>Diagrams clearly highlighting the areas of possible extensions, including the main geological interpretations and future drilling areas, provided this information is not commercially sensitive.</i> 	<ul style="list-style-type: none"> Further work will focus on metallurgical testwork to determine possible processing options, step out drilling to extend both the 309 and Lone Sister deposits at depth and infill drilling at the Lone Sister deposit to allow higher confidence resource estimation.

c. Section 3 Estimation and Reporting of Mineral Resources

(Criteria listed in section 1, and where relevant in section 2, also apply to this section.)

Criteria	JORC Code explanation	Commentary
<i>Database integrity</i>	<ul style="list-style-type: none"> <i>Measures taken to ensure that data has not been corrupted by, for example, transcription or keying errors, between its initial collection and its use for Mineral Resource estimation purposes.</i> <i>Data validation procedures used.</i> 	<ul style="list-style-type: none"> The data was provided as three separate databases none of which contained all the data. These 3 databases were compiled into a single database and then validated. Validation checks included checks for duplicate samples, duplicate holes, interval overlaps, extreme high grades. Where errors were found these were checked against the original assay certificates and fixed. The data provided included many types of drilling. All RAB, open hole percussion, auger and air core drilling was omitted from the database prior to use in resource estimation as these drilling methods are not considered suitable for resource estimation.
<i>Site visits</i>	<ul style="list-style-type: none"> <i>Comment on any site visits undertaken by the Competent Person and the outcome of those visits.</i> <i>If no site visits have been undertaken indicate why this is the case.</i> 	<ul style="list-style-type: none"> The competent person visited the site for a period of two days in December 2018. During this visit archived drill core was reviewed and compared to assays and the site layout and infrastructure assessed.
<i>Geological interpretation</i>	<ul style="list-style-type: none"> <i>Confidence in (or conversely, the uncertainty of) the geological interpretation of the mineral deposit.</i> <i>Nature of the data used and of any assumptions made.</i> <i>The effect, if any, of alternative interpretations on Mineral Resource estimation.</i> <i>The use of geology in guiding and controlling Mineral Resource estimation.</i> <i>The factors affecting continuity both of grade and geology.</i> 	<ul style="list-style-type: none"> The confidence in geological interpretation at the 309 deposit varies with scale. The broad mineralization envelope has been well defined by drilling and has a gradation boundary and so presents little risk at the low grade interpreted. The confidence in the geological interpretation of high grade veins within this envelope is low as the controls on these veins are not understood and drilling shows that the spatial continuity of these veins is low (less than 20m and usually less than 10 m).

Criteria	JORC Code explanation	Commentary
		<ul style="list-style-type: none"> The 309 mineral resource estimate was controlled by a gold grade domain interpreted at a nominal 0.2 g/t Au. A gold grade domain was used because gold mineralization does not honour any geological or alteration features. The grade of 0.2 g/t Au is somewhat arbitrary, being approximately 50% of the lowest reasonably foreseeable mining cut-off (assuming open pit mining and heap leach processing). A cumulative probability plot of gold grades showed no natural lower cut-off to gold mineralization. The confidence in the geological interpretation of the Lone Sister deposit is moderate. The geometry of the gold mineralization is simpler and more continuous, although in places alternative, reasonable interpretations are possible. The country rock is monotonous and so has no impact on the geological variability of gold mineralization. The Lone Sister mineral resource estimate was controlled by a gold grade domain interpreted at a nominal 1.0 g/t Au. A gold grade domain was used because gold mineralization does not honour any geological or alteration features. The grade of 1.0 g/t Au is somewhat arbitrary, being approximately 50% of the lowest reasonably foreseeable mining cut-off (assuming underground mining and CIL processing). A cumulative probability plot of gold grades showed no natural lower cut-off to gold mineralization.
Dimensions	<ul style="list-style-type: none"> The extent and variability of the Mineral Resource expressed as length (along strike or otherwise), plan width, and depth below surface to the upper and lower limits of the Mineral Resource. 	<ul style="list-style-type: none"> The 309 deposit extends about 350 m in strike (north) length and 150 m wide (east-west). The mineralized breccia has been shown by drilling to extend to at least 400m below surface and is open at depth. The Lone Sister deposit has been defined by drilling over a strike length of 250 m and to a depth of 400 m. The width varies from 2 m to a maximum of 60m but is typically 15 m – 30 m wide. The main mineralized zone plunges about 45° towards 010°.
Estimation and modelling techniques	<ul style="list-style-type: none"> The nature and appropriateness of the estimation technique(s) applied and key assumptions, including treatment of extreme grade values, domaining, interpolation parameters and maximum distance of extrapolation from data points. If a computer assisted estimation method was chosen include a description of computer software and parameters used. The availability of check estimates, previous estimates and/or mine 	<p>The 309 deposit:</p> <ul style="list-style-type: none"> Gold and silver grades were interpolated into a block model with parent blocks 5 m x 5 m x 5 m and sub-blocks of 1.25 m x 1.25 m x 1.25 m by ordinary kriging. The raw assay data were composited to 2.0 m prior to geostatistical analysis and interpolation. The interpolation was carried out within a gold grade domain interpreted from 2.0 m long composites at a nominal 0.2 g/t as hard boundary

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	<p><i>production records and whether the Mineral Resource estimate takes appropriate account of such data.</i></p> <ul style="list-style-type: none"> <i>The assumptions made regarding recovery of by-products.</i> <i>Estimation of deleterious elements or other non-grade variables of economic significance (e.g. sulphur for acid mine drainage characterisation).</i> <i>In the case of block model interpolation, the block size in relation to the average sample spacing and the search employed.</i> <i>Any assumptions behind modelling of selective mining units.</i> <i>Any assumptions about correlation between variables.</i> <i>Description of how the geological interpretation was used to control the resource estimates.</i> <i>Discussion of basis for using or not using grade cutting or capping.</i> <i>The process of validation, the checking process used, the comparison of model data to drill hole data, and use of reconciliation data if available.</i> 	<p>and two orientation domains separating zones of shallowly east dipping and sub-vertical high grade veins. High grade samples were not restricted because these showed geological continuity where there is sufficiently dense data and because the block model reconciled to past production better when these data were used. The ordinary kriging used separate variogram models in each orientation domain. The variogram models are shown below. The search neighbourhood used an appropriately oriented ellipsoid with axes 50 m by 50 m 25 m in the flat domain and 80 m x 40 m x 20 m in the steep domain. A minimum of 5 and a maximum of 30 composites (maximum 10 per quadrant) were used for interpolation in both domains.</p> <table border="1" data-bbox="1234 612 2022 1358"> <thead> <tr> <th colspan="2"></th> <th colspan="2" style="text-align: center;">Mineralisation Domain</th> </tr> <tr> <th colspan="2"></th> <th style="text-align: center;">1 (shallow east dip)</th> <th style="text-align: center;">2 (steep)</th> </tr> <tr> <th colspan="2" style="text-align: center;">Exp VG Type</th> <th style="text-align: center;">correlogram</th> <th style="text-align: center;">correlogram</th> </tr> </thead> <tbody> <tr> <td style="text-align: center;">Nugget Variance</td> <td style="text-align: center;">C0</td> <td style="text-align: center;">0.44</td> <td style="text-align: center;">0.57</td> </tr> <tr> <td rowspan="2" style="text-align: center;">Sills (Spherical)</td> <td style="text-align: center;">C1</td> <td style="text-align: center;">0.27</td> <td style="text-align: center;">0.2</td> </tr> <tr> <td style="text-align: center;">C2</td> <td style="text-align: center;">0.29</td> <td style="text-align: center;">0.2</td> </tr> <tr> <td rowspan="3" style="text-align: center;">Ranges_1 (m)</td> <td style="text-align: center;">Maj</td> <td style="text-align: center;">4.5</td> <td style="text-align: center;">5</td> </tr> <tr> <td style="text-align: center;">Sem</td> <td style="text-align: center;">5</td> <td style="text-align: center;">3</td> </tr> <tr> <td style="text-align: center;">Min</td> <td style="text-align: center;">3</td> <td style="text-align: center;">3</td> </tr> <tr> <td rowspan="3" style="text-align: center;">Ranges_2 (m)</td> <td style="text-align: center;">Maj</td> <td style="text-align: center;">16</td> <td style="text-align: center;">30</td> </tr> <tr> <td style="text-align: center;">Sem</td> <td style="text-align: center;">20</td> <td style="text-align: center;">14</td> </tr> <tr> <td style="text-align: center;">Min</td> <td style="text-align: center;">6</td> <td style="text-align: center;">8</td> </tr> <tr> <td rowspan="3" style="text-align: center;">Variogram Rot (MEDS, ZXY)</td> <td style="text-align: center;">Maj</td> <td style="text-align: center;">170</td> <td style="text-align: center;">130</td> </tr> <tr> <td style="text-align: center;">Sem</td> <td style="text-align: center;">0</td> <td style="text-align: center;">-70</td> </tr> <tr> <td style="text-align: center;">Min</td> <td style="text-align: center;">10</td> <td style="text-align: center;">90</td> </tr> <tr> <td rowspan="3" style="text-align: center;">Actual Direction (plunge/trend)</td> <td style="text-align: center;">Maj</td> <td style="text-align: center;">00/170</td> <td style="text-align: center;">-70/130</td> </tr> <tr> <td style="text-align: center;">Sem</td> <td style="text-align: center;">-10/080</td> <td style="text-align: center;">-20/302</td> </tr> <tr> <td style="text-align: center;">Min</td> <td style="text-align: center;">80/080</td> <td style="text-align: center;">02/213</td> </tr> </tbody> </table>			Mineralisation Domain				1 (shallow east dip)	2 (steep)	Exp VG Type		correlogram	correlogram	Nugget Variance	C0	0.44	0.57	Sills (Spherical)	C1	0.27	0.2	C2	0.29	0.2	Ranges_1 (m)	Maj	4.5	5	Sem	5	3	Min	3	3	Ranges_2 (m)	Maj	16	30	Sem	20	14	Min	6	8	Variogram Rot (MEDS, ZXY)	Maj	170	130	Sem	0	-70	Min	10	90	Actual Direction (plunge/trend)	Maj	00/170	-70/130	Sem	-10/080	-20/302	Min	80/080	02/213
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		<p>estimate any other variable</p> <ul style="list-style-type: none"> • Variants using no high grade restriction and inverse distance squared weighting were used to assess the interpolation parameters used. • The block model was validated visually against the raw assay data, statistically against de-clustered average composite grade and by the use of swath plots. • The block model was reconciled to past production (76,000 t at 6.8 g/t Au compared to past production of 75,848 t @ 10.0 g/t Au). <p>The Lone Sister Deposit</p> <ul style="list-style-type: none"> • Gold and silver grades were interpolated into a block model with parent blocks 4 m x 16 m x 10 m and sub-blocks of 1 m x 4 m x 12.5 m by ordinary kriging. The raw assay data were composited to 2.0 m prior to geostatistical analysis and interpolation. The interpolation was carried out within a single gold grade domain interpreted at a nominal 1.0 g/t as hard boundary. High grade samples greater than 60 g/t were restricted to within 16 m. 60 g/t was selected as the outlier restriction from cumulative probability plots and a visual assessment of the continuity of grades above 60 g/t. The variogram model used is shown below. The search neighbourhood used a spherical search ellipsoid with axes of 60 m. A minimum of 4 and a maximum of 30 composites (maximum 12 per quadrant) were used for interpolation in both domains. <table border="1" data-bbox="1234 1007 1850 1396"> <thead> <tr> <th colspan="2"></th> <th>Mineralisation Domain</th> </tr> </thead> <tbody> <tr> <td colspan="2"></td> <td>11</td> </tr> <tr> <td colspan="2"></td> <td>correlogram</td> </tr> <tr> <td>Nugget Variance</td> <td>C0</td> <td>0.60</td> </tr> <tr> <td rowspan="2">Sills (Spherical)</td> <td>C1</td> <td>0.05</td> </tr> <tr> <td>C2</td> <td>0.35</td> </tr> <tr> <td rowspan="3">Ranges_1 (m)</td> <td>Maj</td> <td>6</td> </tr> <tr> <td>Sem</td> <td>6</td> </tr> <tr> <td>Min</td> <td>6</td> </tr> </tbody> </table>			Mineralisation Domain			11			correlogram	Nugget Variance	C0	0.60	Sills (Spherical)	C1	0.05	C2	0.35	Ranges_1 (m)	Maj	6	Sem	6	Min	6
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<i>Moisture</i>	<ul style="list-style-type: none"> • <i>Whether the tonnages are estimated on a dry basis or with natural moisture, and the method of determination of the moisture content.</i> 	<ul style="list-style-type: none"> • All tonnages are reported on a dry basis. No moisture content was determined. 																					
<i>Cut-off parameters</i>	<ul style="list-style-type: none"> • <i>The basis of the adopted cut-off grade(s) or quality parameters applied.</i> 	<ul style="list-style-type: none"> • For the 309 resource estimate a cut-off grade of 0.4 g/t Au was applied above 1050 RL, and 2.0 g/t Au below 1050 RL. These cut-off grades assume that open pit mining is feasible to 1050 RL and underground mining below that and that a CIL processing plant could be economically built and operated at, or close to the site. • For the Lone Sister resource estimate a cut-off grade of 2.0 g/t Au was applied, assuming that underground mining and a CIL processing plant could be economically built and operated at, or close to the site. 																					
<i>Mining factors or assumptions</i>	<ul style="list-style-type: none"> • <i>Assumptions made regarding possible mining methods, minimum mining dimensions and internal (or, if applicable, external) mining dilution. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential mining methods, but the assumptions made regarding mining methods and parameters when estimating Mineral Resources</i> 	<ul style="list-style-type: none"> • For the 309 deposit open pit mining is assumed possible to 1050 RL based on a pit optimization of a previous model at AUD1500/oz Au. Given the increase in the gold price since that time, open pit mining is likely economic to greater depths so this may be a little conservative. Mining selectivity to a minimum of 2 metres is assumed. • For the 309 and Lone Sister deposits underground mining is assumed 																					

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	<p><i>may not always be rigorous. Where this is the case, this should be reported with an explanation of the basis of the mining assumptions made.</i></p>	<p>possible to a minimum width of 3.0m</p>
<p>Metallurgical factors or assumptions</p>	<ul style="list-style-type: none"> <i>The basis for assumptions or predictions regarding metallurgical amenability. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential metallurgical methods, but the assumptions regarding metallurgical treatment processes and parameters made when reporting Mineral Resources may not always be rigorous. Where this is the case, this should be reported with an explanation of the basis of the metallurgical assumptions made.</i> 	<ul style="list-style-type: none"> CIL recoveries greater than 85% have been demonstrated for ore above 4 g/t by past production at 309. There is limited testwork suggesting that CIL recoveries above 80% are possible for low grade 309 mineralisation. Heap leach recovery may also be possible for low grade ore at 309 but there is no test work to support this. No metallurgical testwork has been carried out on Lone Sister mineralization. Economically viable CIL recovery is assumed on the basis that the mineralization is similar to 309 mineralisation.
<p>Environmental factors or assumptions</p>	<ul style="list-style-type: none"> <i>Assumptions made regarding possible waste and process residue disposal options. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider the potential environmental impacts of the mining and processing operation. While at this stage the determination of potential environmental impacts, particularly for a greenfields project, may not always be well advanced, the status of early consideration of these potential environmental impacts should be reported. Where these aspects have not been considered this should be reported with an explanation of the environmental assumptions made.</i> 	<ul style="list-style-type: none"> It is assumed that the placement of mining waste would be permitted within the current mining lease as there are existing waste dumps. The presence of sulphide minerals in the 309 waste material suggests that it may be potentially acid forming, although there has been no test work to confirm this. The Lone Sister fresh waste also contains sulphide minerals but at a lower level and so some may be acid forming. There is currently no tailings storage facility at site.
<p>Bulk density</p>	<ul style="list-style-type: none"> <i>Whether assumed or determined. If assumed, the basis for the assumptions. If determined, the method used, whether wet or dry, the frequency of the measurements, the nature, size and representativeness of the samples.</i> <i>The bulk density for bulk material must have been measured by methods that adequately account for void spaces (vugs, porosity, etc.), moisture and differences between rock and alteration zones within the deposit.</i> <i>Discuss assumptions for bulk density estimates used in the evaluation process of the different materials.</i> 	<ul style="list-style-type: none"> Bulk density at the both deposits was assigned by weathering domain. There are only 62 density samples from the two deposits combined and all of these are from fresh material. A density of 2.6 t/m³ was applied to fresh material and 2.4 t/m³ to oxide material. 2.6 t/m³ approximates the average of the 62 fresh samples (2.61 t/m³) and the oxide density of 2.4 t/m³ was assumed from comparable projects. Voids created by previous mining at the 309 deposit were assigned a density of 0.0 t/m³.
<p>Classification</p>	<ul style="list-style-type: none"> <i>The basis for the classification of the Mineral Resources into varying confidence categories.</i> <i>Whether appropriate account has been taken of all relevant factors (i.e. relative confidence in tonnage/grade estimations, reliability of input data, confidence in continuity of geology and metal values, quality, quantity and distribution of the data).</i> 	<ul style="list-style-type: none"> The 309 mineral resources were classified into continuous zones of like confidence using the distance to the nearest composite, the average distance of all composites used to estimate a block, the kriging slope of regression (a function of the grade continuity in the variogram model and the spatial configuration of the data used to estimate a block) and the number of drill holes used to estimate a

Criteria	JORC Code explanation	Commentary
	<ul style="list-style-type: none"> Whether the result appropriately reflects the Competent Person's view of the deposit. 	<p>block. The geological continuity of the gold domain was not used for resource classification because the shape is rather simple and the resource cut-off grade is well above the domain nominal cut-off grade so that the resource limits are defined by the cut-off grade rather than the domain shape. Data quality was also considered but not used because the data quality is good and not spatially variable.</p> <ul style="list-style-type: none"> The Lone Sister deposit was all classified as inferred due to the limited
Audits or reviews	<ul style="list-style-type: none"> The results of any audits or reviews of Mineral Resource estimates. 	<ul style="list-style-type: none"> This mineral resource estimate has not been audited or reviewed as the project is at a preliminary stage of development.
Discussion of relative accuracy/confidence	<ul style="list-style-type: none"> Where appropriate a statement of the relative accuracy and confidence level in the Mineral Resource estimate using an approach or procedure deemed appropriate by the Competent Person. For example, the application of statistical or geostatistical procedures to quantify the relative accuracy of the resource within stated confidence limits, or, if such an approach is not deemed appropriate, a qualitative discussion of the factors that could affect the relative accuracy and confidence of the estimate. The statement should specify whether it relates to global or local estimates, and, if local, state the relevant tonnages, which should be relevant to technical and economic evaluation. Documentation should include assumptions made and the procedures used. These statements of relative accuracy and confidence of the estimate should be compared with production data, where available. 	<ul style="list-style-type: none"> The relative accuracy and confidence level in the Mineral Resource estimates are considered to be in line with the generally accepted accuracy and confidence of the nominated Mineral Resource categories. This has been determined on a semi- quantitative basis, and is based on the Competent Person's experience with similar deposits. The resource classification relates to both global and local estimates. For the 309 deposit the block model was reconciled to past production (75,848 t @ 10.0 g/t Au. This resource estimate reports 76,000 t at 6.8 g/t Au from the mined voids).