

Catalyst Metals' flagship asset is the 40km long Plutonic Gold Belt in Western Australia. This belt currently produces ~100koz pa at an AISC of ±A\$2,300/oz from three mines at Plutonic, Plutonic East and Trident open pit.

Catalyst is currently bringing three new mines into production – Trident UG, Old Highway and Cinnamon. Each will be processed through the existing, underutilised and centrally located 2Mtpa CIL processing plant.

Exploration is targeting down dip extensions of each of these deposits.

With the development and exploration of these five deposits, Catalyst aims to increase Reserves and production from 1.5Moz to ±2Moz and ±100koz to ±200koz annually.

In so doing, Catalyst is aiming for Plutonic to have a 10 year mine life - a unique and rare proposition for an underground Western Australian gold mine.

Catalyst also controls a processing plant and +75km of strike length immediately north of the historic +22Moz Bendigo goldfield. Here, Catalyst has delineated a high-grade, greenfield resource at 26 g/t Au. Further discoveries along strike are expected.

Capital Structure

Shares o/s: 261m
Options: 0.5m
Rights: 14.1m
Cash & Bullion: A\$277m
Debt: Nil

Reserve and Resource^{1,2}

MRE: 4.5Moz at 3.3g/t Au
ORE: 1.5Moz at 2.6g/t Au

Corporate Details

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Trident's Resource grows to 1.1Moz at 5.4g/t

Both Indicated and Inferred Resources continue to grow, further underpinning +10 year mine life at ±60koz per annum

- **Trident is an underground deposit located 30km north-east of the underutilised 2Mtpa Plutonic processing plant**
- **The mine is under development with an open pit recently completed to allow for the establishment of an underground portal**
- **When Catalyst acquired Trident, it had a Resource of 524koz at 3.6 g/t Au. Today, the Resource stands at 1.1Moz at 5.4g/t Au**
- **The more tightly drilled Indicated Resources have increased to 633koz at 6.3g/t Au, a 20% increase on prior estimates**
- **A ±10-year mine plan at an average run-rate of ±60koz per annum is underpinned by this Resource, previous Indicated Resources converting at a rate of 75% to Reserves, and further Inferred Resources to be drilled to convert to Indicated**
- **Trident remains the second-largest deposit on the Plutonic Belt and is expected to form a second, higher grade base load ore source to the central Plutonic processing plant as part of Catalyst's plan to increase annual gold production from ±100koz to ±200koz**
- **The Trident underground will be the fourth mine to be developed on the Plutonic Belt, the first three being Plutonic East, Trident open pit and K2**
- **First stoping ore from the underground mine is expected in the first half of CY2027**

Catalyst Metals Limited (**Catalyst** or the **Company**) (ASX:CYL) is pleased to provide an updated Mineral Resource Estimate for the Trident underground gold deposit, located on the Plutonic Gold Belt in Western Australia.

The updated Resource estimate is as follows:

Classification	Tonnes (Mt)	Grade (g/t Au)	Ounces (Koz)
Indicated	3.1	6.3	633
Inferred	3.1	4.6	448
Total	6.2	5.4	1,081

Catalyst's Managing Director & CEO, James Champion de Crespigny, commented:

"Our exploration team continues to have success across the belt, meaningfully growing Trident, Plutonic Main, Cinnamon, K2 and Old Highway. This 145% increase in Trident's Indicated Resources, and achievement of this 1Moz milestone, is a testament to their persistence.

A key focus now is on developing the Trident orebody. With the open pit completed, the successful development of the underground, and path to ±200koz, is becoming ever more likely."

Note 1: MRE includes Indicated Resources of 29Mt at 2.9g/t for 2.7Moz and Inferred Resources of 9Mt at 2.7g/t for 0.8Moz. ORE includes probable Reserves of 10.6Mt at 3.0g/t for 1.0Moz. Note 2: Refer to ASX announcement 14 October 2025 "Annual Report to shareholders. Note 3: Refer to ASX announcement 10 September 2025 "Plutonic Belt Reserves double, supporting growth plans 1

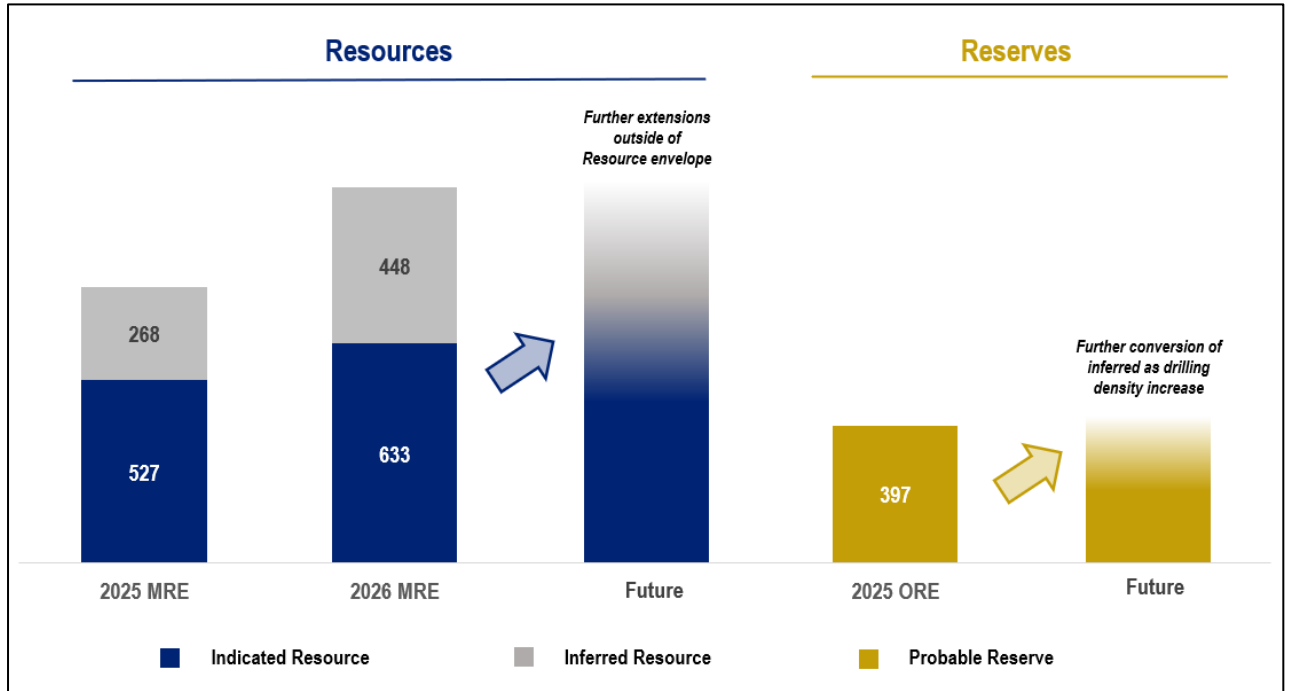


Figure 1: Past Trident MRE and ORE growth under Catalyst ownership. Continued drilling will target conversion of Resources to Reserves and growing the Resource envelope ^{3,4}

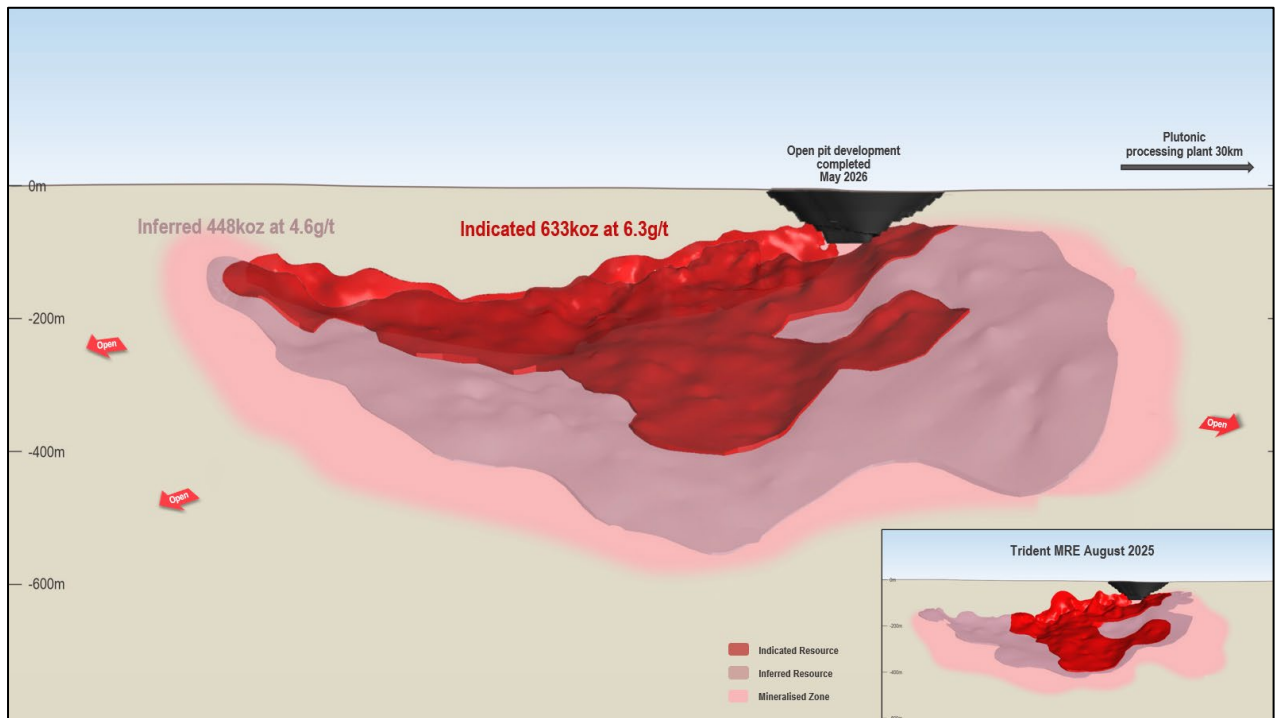


Figure 2: Trident long section showing Indicated and Inferred Resources

³ ASX announcement 4 August 2025 "Trident Indicted Resource Doubles"

⁴ ASX announcement 10 September 2025 "'Plutonic Belt Reserves double, supporting long term growth plans" and "Investor Presentation"

Trident Underground Mineral Resources

The Mineral Resource Statement for the Trident Underground Mineral Resource estimate has been prepared during June 2026 and is reported according to the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (the 'JORC Code') 2012 edition.

The Mineral Resource Estimate incorporates all drilling data completed by Catalyst since the previous MRE (July 2025) to the current data cut-off date of 4 June 2026. A total of 111,001m of drilling from 230 diamond/diamond tails and 110 RC drillholes form the basis of this Mineral Resource Update. This included 72 RC grade control holes for 11,221m to an average depth of 155m which targeted the initial 12 to 18 months of planned production.

The current vertical extent of the Trident Underground Mineral Resource is approximately 550m (to 50mRL) from the current topographic surface at 600mRL.

In the opinion of Catalyst, the resource evaluation reported herein is a reasonable representation of the global gold Mineral Resources within the Trident deposit, based on sampling data from RC, RCD, RC/DD and DD drilling available as of 4 June 2026.

The Trident Underground Mineral Resource (MRE) has been undertaken with a focus on delineating areas of the MRE with Reasonable Prospects for Eventual Economic Extraction (RPEEE) by underground mining methods. The MRE has been reported at a 2.0 g/t Au cut-off below the topographic surface within fresh rock to a maximum vertical depth of 550m below the topographic surface.

The total MRE consists of Indicated and Inferred Mineral Resources. No Measured Mineral Resources have been reported at this stage of the project.

The Mineral Resource Statement is presented in Table 1.

Table 1: Trident Underground Mineral Resource (at a 2.0 g/t Au cut-off)

Classification	Tonnes (Mt)	Grade (g/t Au)	Ounces (Koz)
Indicated	3.1	6.3	633
Inferred	3.1	4.6	448
Total	6.2	5.4	1,081

Notes:

1. Mineral Resource reported at a 2.0 g/t Au cut-off to 550m vertical depth below the ground surface
2. Numbers may not add up due to rounding

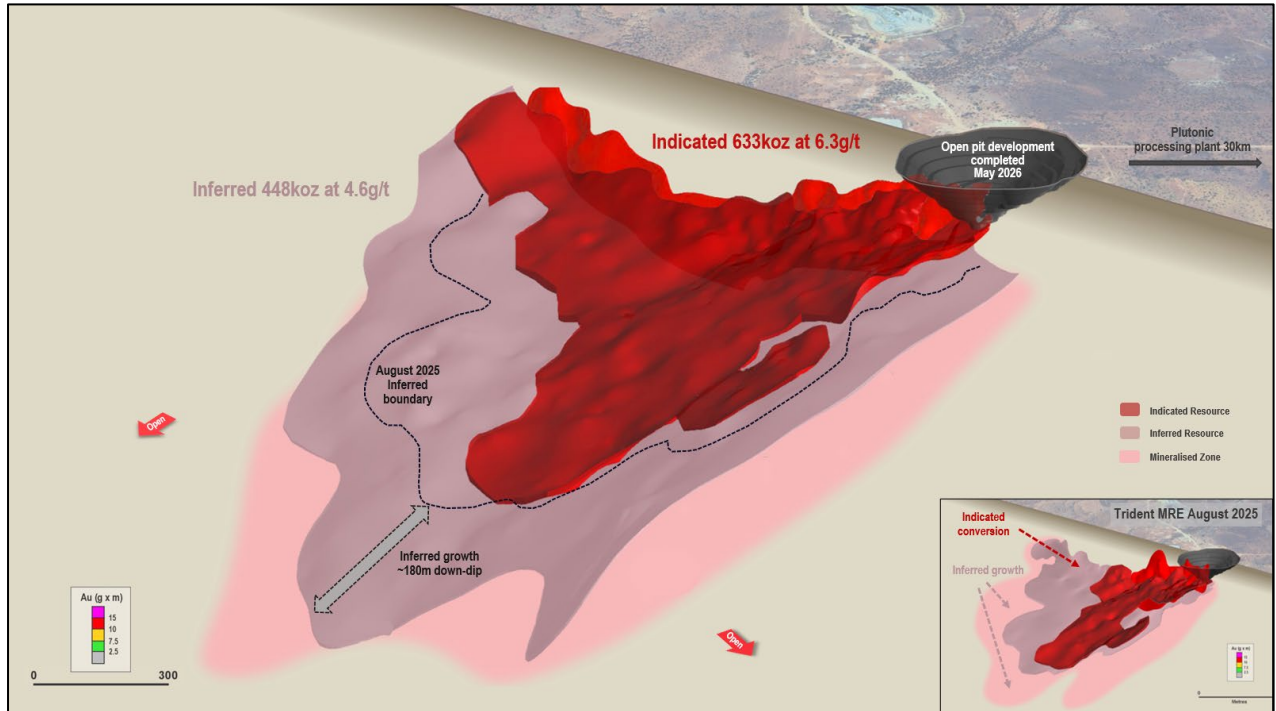


Figure 3: Trident oblique showing Indicated and Inferred Resources

Trident Gold Project

Trident is an underground gold deposit located 30km north-east of the underutilised Plutonic processing plant. Trident underground comprises a probable Reserve of 2.5Mt at 5.0g/t Au for 397koz⁵ of gold and a Resource of 6.2Mt at 5.4g/t Au for 1,081koz (including Indicated of 3.1Mt at 6.3g/t Au for 633koz).

Mining of a small open pit at Trident commenced in July 2025 and was completed in May 2026. Ore from the open pit has been stockpiled at Trident and will be progressively transported and processed at the Plutonic processing plant over the remainder of CY2026.

Catalyst’s objective with the Trident open pit was to establish access for development of the underground decline i.e. a box-cut.

Grade control drilling of the first 15 months of production has been completed. In Catalyst’s view, developing these drill stocks ahead of the mine plan provides operational flexibility and is key to de-risking early mining operations in the project’s ramp up phase.

To date, increased drilling density at Trident has resulted in improved grade. This was noted with both Resource conversion and grade control drilling. This is encouraging as drilling to convert the existing 448koz at 4.6g/t Inferred Resource continues.

Trident is currently the second largest deposit on the Plutonic Gold Belt. The underground mine is expected to operate at a run rate of around ±60koz per annum and will form a second, higher-grade base load ore source feeding the Plutonic processing plant.

⁵ ASX announcement 10 September 2025 “Plutonic Belt Reserves double, supporting growth plans.”

Catalyst's 10-year production plan

In September 2025, Catalyst released a 10-year production plan showing growth in gold production at the Plutonic Gold Belt from ±100koz pa to ±200koz pa (refer to Figure 4). This production is planned to be sourced from five underground mines - Plutonic Main, Plutonic East, Trident, K2 and Old Highway.

Trident, K2 and Old Highway underground mines are three higher-grade ore sources to be brought on-line. Higher grade ore sources will lift the overall blended grade to be processed at the Plutonic processing plant. This in turn is expected to lower unit costs (refer to Figure 4).

The nearby Cinnamon deposit was not included in this production plan, however following recent exploration success, Catalyst considers Cinnamon will form a sixth ore source and is progressing approvals to move the deposit toward development.

The current development plan assumes full utilisation of Plutonic's existing, but underutilised 2Mtpa processing plant. A second, 1Mtpa processing train located at Plutonic has been on care and maintenance since 2008. A decision to refurbish this plant has not yet been made by Catalyst. Catalyst's current plan to develop and process the six ore sources and increase gold production to ±200koz pa is expected to be achieved with the existing 2Mtpa plant.

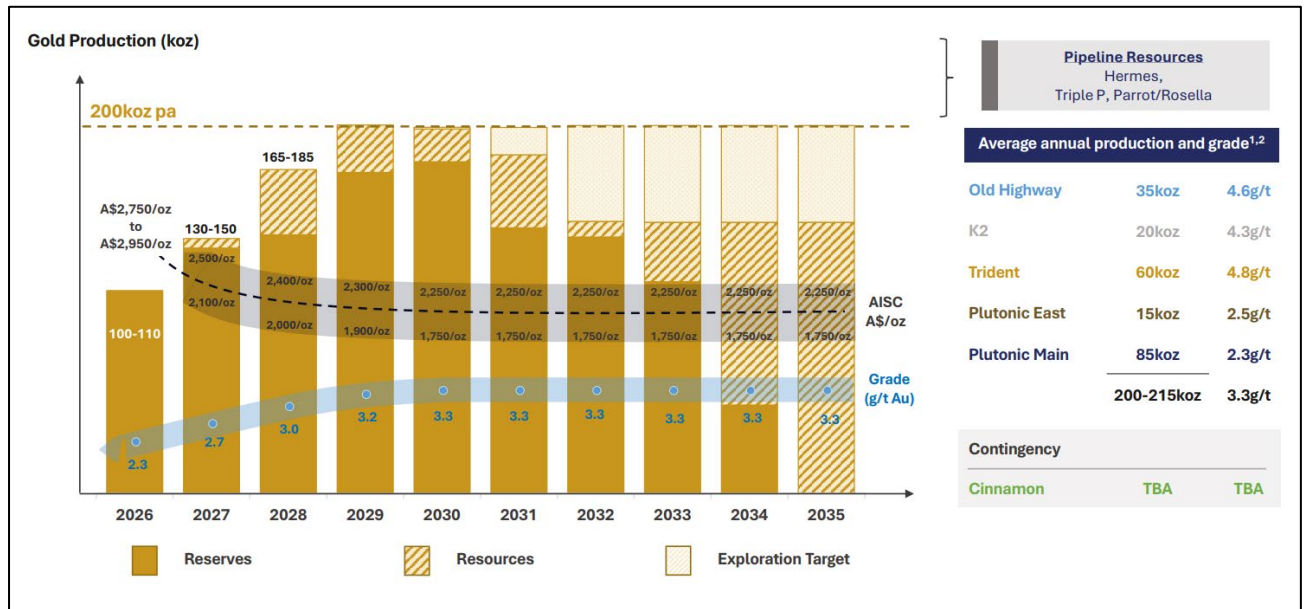


Figure 4: Catalyst's 10-year production target^{6,7}

⁶ ASX announcement 8 May 2025 "Catalyst to acquire Old Highway Gold Project"

⁷ ASX announcement 10 September 2025 "'Plutonic Belt Reserves double, supporting long term growth plans" and "Investor Presentation"

MINERAL RESOURCE ESTIMATE (Trident Underground)

Drilling Techniques

All drilling data used in this Mineral Resource Estimate has been sourced from Diamond (DD) and Reverse Circulation (RC) drilling methods.

The drillhole Quest database, managed by EarthSQL, has been compiled from information collected when the Project was under ownership of numerous companies including:

- Catalyst Metals (2023 to current),
- Vango (2014 to 2023),
- Dampier Gold (2012 to 2014),
- Barrick Gold (2001 to 2012),
- Homestake (1999 to 2001),
- Resolute (1995 to 1998).

Catalyst planned drillholes have been routinely pegged using a Differential Global Positioning System (DGPS) handheld unit and marked with wooden pegs. Once completed, the collar position has been re-surveyed using the same DGPS unit to an accuracy of +/- 20mm. The surveyed co-ordinates replace the planned co-ordinates in the database.

All drilling has been undertaken in the MGA_GDA94 Zone 50 grid.

RC drilling has been conducted utilizing a 5.75" face sampling bit. Diamond drilling utilises NQ core with a diameter of 47.6 mm.

Historical Drilling

Extensive previous work was completed by Resolute Mining, Homestake Gold, Battle Mountain Australia, Barrick Mining and Dampier Gold. Previous metallurgical and resource work was completed by Resolute Mining, Barrick Mining and Dampier Gold.

The quality of historical drilling information is varied, but all of the above companies used high quality methodology at the time.

Catalyst consolidated the belt in 2023 following the successful acquisition of Vango Mining and the merger with Superior Gold Inc.

Vango (2014 to 2023)

Planned drillhole collars were pegged with a DGPS and marked with wooden pegs hammered into the ground and flagged with high visibility flagging tape. On completion of drilling, the actual drillhole collar position was measured by survey staff using a DGPS working off a network control of survey stations, to an accuracy of 20mm. These surveyed coordinates replaced the planned coordinates in the geological database.

All reported coordinates were referenced to grid system MGA_GDA94 Zone 50. The topography is relatively flat at the location of drilling.

The survey station network met the Mine Safety and Inspection Regulations 1995, section 3.49, where the accuracy of a survey must be not less than 1:5000.

The collar locations of historic drillholes were validated from geological logging information from annual reports and the original database when Vango acquired the tenure.

Recent (2023) downhole survey data was collected by Westdrill using an Axis Mining Technology Champ North Seeking Gyro tool. Surveys were conducted at end-of-hole (EOH) using a north-seeking gyroscope with a reading collected every 5m. If early drilling found strong drillhole deviation, surveys were conducted during drilling (collar, 30m, 60m, 90m etc to EOH). Survey deviation was monitored by the geologist onsite, with major deviation discussed with the driller at the time.

Recent Vango RC drilling was conducted utilizing 5.75" face sampling bit. Diamond drilling was conducted utilising NQ2 core. Core was orientated by downhole spear methodology.

Pre-Vango (1995 – 2014)

Previous downhole survey data was collected using a REFLEX gyro tool and historically with Eastman camera survey tools, with follow-up downhole surveys carried out by Surtron using gyroscopic survey equipment. Historical downhole surveys were reviewed and verified where information was available through direct comparison within the database.

Historical Diamond drillholes utilised PQ3, HQ3 or NQ2 core diameter, while RC drilling utilised a 5.5" drill bit.

Sampling and Sub-Sampling Techniques

Catalyst has undertaken both RC and DD drilling at the Trident deposit since acquiring the project.

RC drillholes and RC pre-collars have been sampled using 3m composited samples through the overlying granite and cover rock, collected from the original 1m riffle split samples from the rig mounted cyclone.

Pre-collars located within the Resource footprint have not been sampled through the barren granite cover rocks.

All RC composited samples were dispatched to the ALS laboratory in Perth for gold fire assay analysis.

Sample preparation procedures for RC samples includes:

- 1-4 hours drying at 150°C depending on moisture content,
- Entire sample crushed to 10mm,
- Riffle splitting to obtain between 1.2 to 3kg.
- Pulverisation to 85% passing 75µm,
- Scoop 250-300g aliquot
- Ore grade assay determination for gold by lead collection fire assay with Atomic Absorption Spectroscopy (AAS [Au-AA26]), using a 50g nominal sample weight.

Half-cut diamond core has been sampled on 1m intervals or to geological contacts, with sample lengths varying between 0.2m to 1.6m. Broken/sheared core, with poor rock strength, that were unable to be cut were initially grab sampled by selecting core pieces. Whole core sampling was later instigated to resolve any possible grade bias issues with sampling of half-core core grab samples in broken/sheared zones.

Whole core sampling for Resource drilling has been instigated to resolve any possible biasing issues with 1/2 core grab samples in broken/sheared zones which could not be cut.

NQ core samples have been sent to several labs: the majority to ALS (Au-AA26), some to SGS (GO_FAP50V10) and several holes to Plutonic onsite lab (PAL_DIBK).

Sample preparation procedures for DD by Plutonic onsite lab (PAL_DIBK) includes:

- 1-4 hours drying at 150°C depending on moisture content

- Entire core sample is crushed to a particle size of 85% passing 3.2mm.
- Riffle splitting to obtain between 300 to 600g.

Sample preparation procedures for DD by Fire Assay method includes:

- 1-4 hours drying at 150°C depending on moisture content
- Entire core sample is crushed to 10mm
- 3kg riffle split for pulverisation
- Pulverise to 90% passing 75µm
- Scoop 250-300g

The sample preparation protocols and sample sizes are considered appropriate for the style of mineralisation encountered and should provide representative results.

Quality Assurance and Quality Control (QAQC) samples are included in each sample batch and are dispatched to the assaying laboratory to measure the accuracy and precision of the returned assays. QAQC samples inserted into the sample stream include:

- Certified Reference Material (CRM's) that are submitted every 20 samples. The selected CRM's are matrix matched and reflect the expected grade of the samples,
- Blanks are inserted every 100 samples,
- Field duplicates were inserted every 20 samples in the RC drilling. A mix of field duplicates and crush duplicates were inserted in the DD drilling, at a combined 1 in every 20 sample rate.

Crush sizing analysis is conducted randomly by the Laboratory as part of their internal QAQC processes. Pulp residues are expected to have 90% passing $\leq 75\mu\text{m}$, which is monitored by the Laboratory Supervisor, and grind times can be lengthened accordingly.

Current procedures dictate a process of validation and checking of laboratory results when the assay data is returned by the laboratory upon loading into the independently managed Quest database. A routine set of plots and checks are undertaken, and if results fall outside of the expected limits, re-assaying of failing samples is requested. QAQC reports are generated by the database administrator and documented from automated database routines.

Historical Sampling

Vango (2014 to 2023)

Vango RC drilling assays were collected as 1m samples split on the cyclone, using a cone splitter. Each RC sample weighed approximately 3 to 5kg. Four metre composites were collected from 4 of the 1m samples within the cover sequence and overlying granite.

Vango diamond drilling samples were mostly half core with minor quarter core, NQ2 and HQ size core. This is considered to be sufficient material for a representative sample. Core samples were taken at 1m intervals or at geological boundaries, ranging between 0.8-1.25m in length. The DD drillholes were geologically logged to geological boundaries in addition to being structurally and geotechnically logged.

Recovery in DD was based on the measured core returned for each 3m run. RC drilling was bagged on 1m intervals and an estimate of sample recovery made based on the size of each sample.

QAQC protocols included the collection and analysis of field duplicates, the insertion of appropriate commercial standards (CRMs) and the insertion of blanks. Matrix and grade matched CRMs were submitted every 20 samples. Blanks were inserted every 20 samples.

Pre-Vango (1995 – 2014)

Sampling procedures earlier than 2018 are not available.

Historical RC samples were collected as 4m composite spear samples. Mineralised zones were sampled at 1m intervals using a 1/8 riffle splitter.

Core samples were halved using a diamond saw and sampled at 1m intervals, or to geological contacts. RC field duplicate sampling was completed by passing the bulk reject sample from the plastic bag through a riffle splitter. DD field duplicates were collected as quarter core samples and routinely submitted. Duplicate sample intervals were designated by the logging geologist.

No assessment of RC chip sample recoveries was undertaken on historical data however a comprehensive historical review of sampling procedures was undertaken which indicates that standard procedures were enacted to ensure minimal sample loss. Where information on the recoveries had been recorded, they have been found to be consistent with those noted by subsequent drilling.

Recovery in diamond drilling based on measured core was returned for each 3m run.

Sample Analysis Methods

NQ core samples have been sent to several labs: the majority to ALS (Au-AA26), some to SGS (GO_FAP50V10) and several holes to Plutonic onsite lab (PAL_DIBK).

The Pulverising and Leach (PAL) method utilised at the Plutonic onsite lab is not considered to be a total gold analysis, however the larger sample size (500g) still produced a representative and comparable result.

The majority of the samples used the fire assay (50g) technique by AAS method which is considered to be a total gold analysis.

For the historical drilling, gold was analysed using fire assay with a 25g to 50g charge for gold.

Specific QAQC procedures for previous owners were unavailable.

Although sample collection, sample preparation, sample logging and analytical techniques have varied over the Project's history, all can be considered as industry standard at the time of drilling. The amount of QAQC data that has been collected has also varied over the Project's history however overall is considered as being acceptable to support the MRE.

Regional and Local Geology

The Trident Deposit is part of the Plutonic Project located approximately 200km north-east of Meekatharra, WA, on the Three Rivers and Marymia Pastoral Leases, within the Peak Hill Mineral Field of WA (Figure 1).

The Project is covered by the Peak Hill (SG 50-8) 1:250,000 Sheet and the Marymia (2847) and Three Rivers (2747) 1:100,000 sheets. Access to the Plutonic Gold Mine Camp is via the Great Northern Highway and the Plutonic Mine access road.

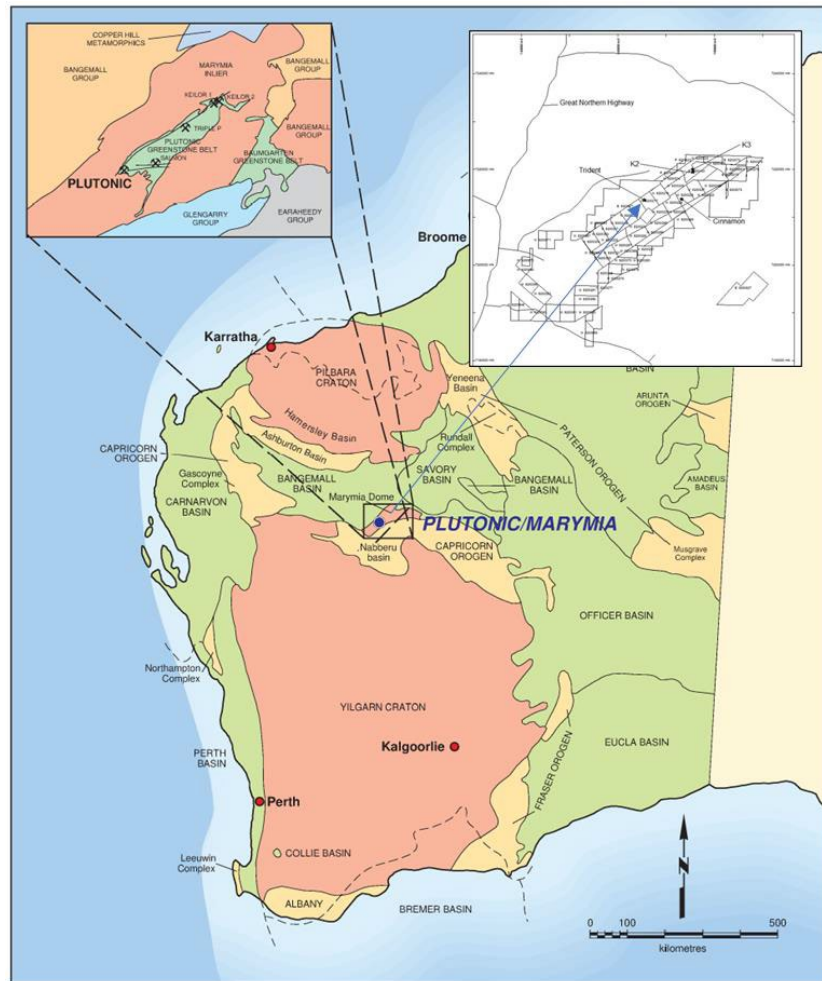


Figure 1 – Plutonic Regional Map

Regionally, the Plutonic Gold Belt lies in the Archaean Plutonic Well Greenstone Belt, an elongate NE trending belt within the Marymia Inlier. The Marymia Inlier is an Archaean basement remnant within the Proterozoic Capricorn Orogen and comprises two mineralised greenstone belts (Plutonic Well and Baumgarten Greenstone Belts), with surrounding granite and gneissic complexes. The Capricorn Orogen is situated between the Pilbara and Yilgarn cratons and is interpreted to be the result of the oblique collision of these two Archaean cratons in the early Proterozoic.

The Marymia Greenstone Belt comprises two corridors of northeast–southwest trending mafic/ultramafic and sedimentary sequences separated by a conglomerate-dominated sedimentary sequence.

Three major structural events are interpreted to have shaped the belt, including D1 low-angle thrusting and isoclinal folding that has emplaced mafic and ultramafic units structurally above the sedimentary units in the northwest side of the belt (the overthrust terrane), followed by southeast directed upright D2 folding and faulting, granite/porphyry sheet intrusion then D3 high-angle thrusting, open folding of earlier structures plus reactivation of the D1 and D2 thrusts.

The Trident gold deposit is a structurally controlled, orogenic, mesothermal (amphibolite metamorphic facies) gold deposit hosted by ultramafic rocks that are part of strike extensions to the Plutonic Gold Mine stratigraphy. It is specifically hosted by shallow to moderate dipping, ultramafic tremolite–phlogopite

(mica) schist, immediately overlying serpentinised ultramafic units, derived from higher MgO ultramafic volcanics.

An overthrust granite package forms the barren hanging wall to mineralisation hosted within the sheared ultramafic host rock package.

High-grade gold zones are best developed within the shallow dipping ultramafic tremolite–phlogopite schist where it is bent into a concave flexure, in the hanging wall of steep, north-westerly dipping fault structures. Vertical “dragging” movement against these steeply dipping faults appears to have played a role in dilating the cleavage of the ultramafic schist, resulting in mineralisation and alteration between the dilated cleavage planes. The steeply dipping faults also host gold mineralisation.

Mineralisation is associated with potassic, phlogopite mica alteration and has a low proportion of quartz and sulphides, including minor pyrrhotite, pentlandite, chalcopyrite and, directly associated with gold, bismuthinite and rare bismuth tellurides. Rarely observed gold grains, in microscopy, are predominantly fine (<50 micron) but free and/or attached to, and rarely occluded within, sulphide grains.

Geological Interpretation

A total of 111,001m of drilling from 230 diamond/diamond tails and 110 RC drillholes were available for interpretation and estimation of the MRE. This included 72 RC grade control holes for 11,221m to an average depth of 155m which targeted the initial 12 to 18 months of planned production.

Drillholes are at a nominal spacing of 20m x 20m in the upper portions of the project, out to greater than 40m x 40m at depth (Figure 2).

Mineralisation domains have been interpreted primarily using geological logging and the location downhole of geological contacts, based on lithology, grade distribution, major faults and geometry. Weathering surfaces have been created by interpreting the existing logging for oxidation state and have been extended laterally beyond the limits of the Mineral Resource block model.

Interpretations of mineralisation continuity have been generated in Leapfrog software using all available drillholes. Intercepts correlating to gold mineralisation and underpinned by strike continuity have been independently identified and manually selected within Leapfrog prior to creation of implicit volume models.

A nominal cut-off grade of 0.5 g/t Au was used to guide the geological continuity of the interpreted mineralisation. Selection of the cut-off grade was based on statistical and spatial analysis of composite data indicating a natural mineralisation population exists above 0.5 g/t Au. Within the mineralised wireframe, if an intercept fell below the nominal cut-off but continuity was supported by host lithologies and mineralisation style, the intercept was retained for continuity purposes due to the commodity and the style of deposit.

CYL considers confidence in mineralisation continuity and distribution, as implied within the Mineral Resource estimate classification of Indicated and Inferred, as moderate to high. The regularised drill pattern, drill centre spacing (varying from 10m to 60m) has allowed confident interpretation of the geometry and continuity of the higher grade mineralised zones informing these Mineral Resources (Figure 2).

The mineralisation consists of multiple stacked zones, with the main mineralised domain extending along a northeast/southwest strike for 1,200m. The system dips at 30° degrees towards 330° and extends down-dip for up to 1,000m, but has now been essentially closed out at depth. Mineralised zones can vary in width from 0.6m up to 15m with an average thickness of around 4.5m.

The higher grade zones are focussed along north-south structures and multiple north-east trending shoots and at the interaction points of these two dominant trends. The orientation of mineralisation can be variable particularly where cross-cutting structures are intersected. These inflexion points are likely dilation zones which can host thicker and higher grade mineralisation intersection (Figure 3).

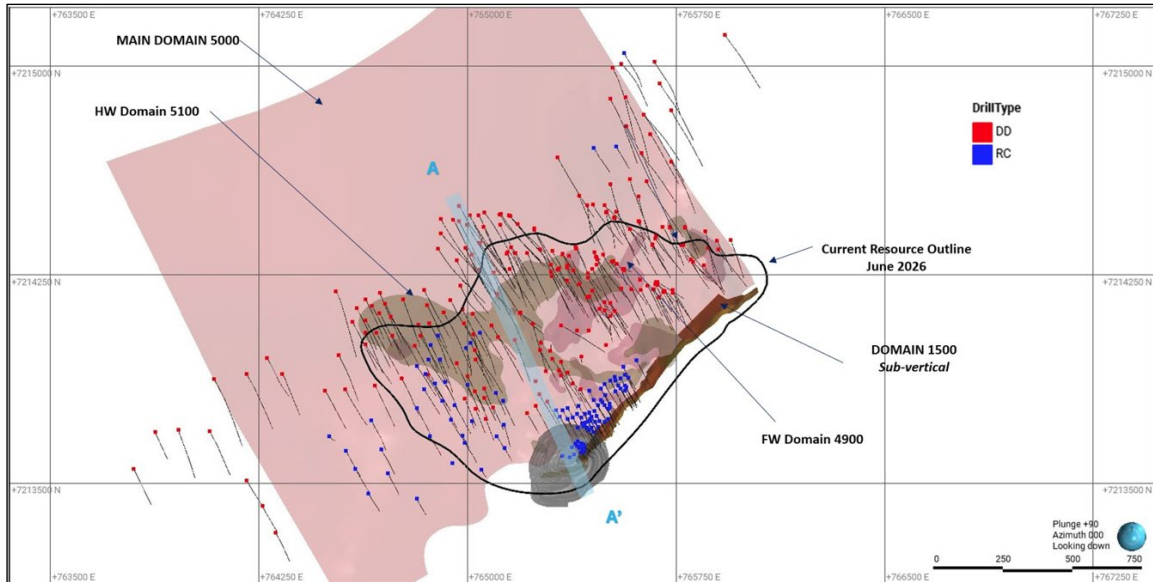


Figure 2 - Trident Plan View – Drilling Type and Mineralisation Domains

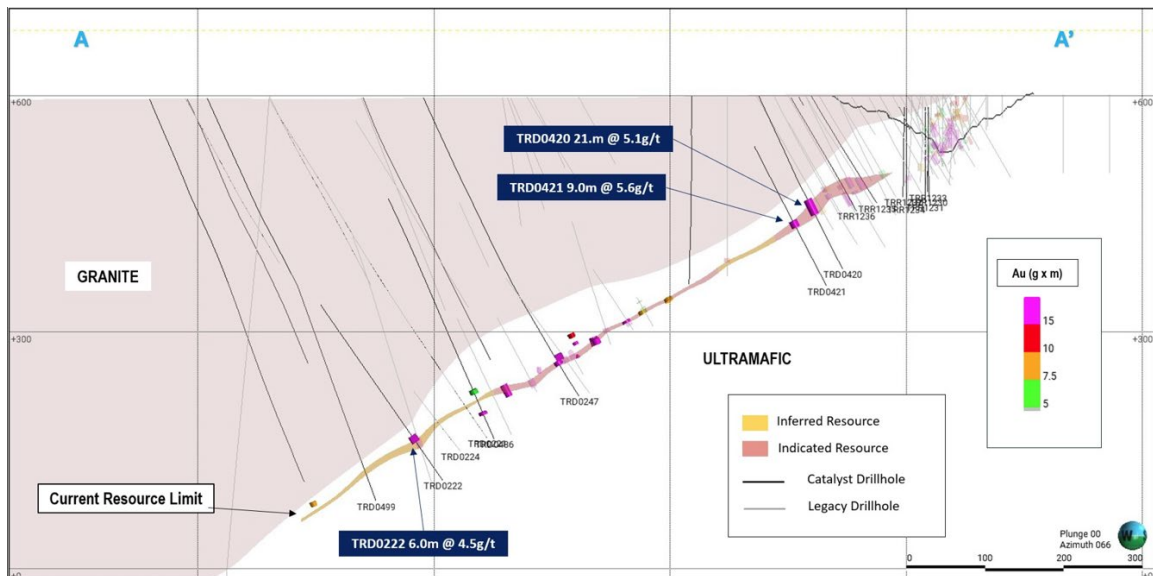


Figure 3 - Trident Cross Section A-A'

Estimation Methodology

All geological domains used in the Trident Underground MRE have been constructed in Leapfrog software. Block modelling and grade interpolation have been carried out using Surpac software. Statistical analysis has been completed using Supervisor software.

Block model constraints have been created by applying the interpreted mineralised domain wireframes. Sub-celling in all domains has been set at 1.25m x 1.25m x 1.25m to accurately reflect the volumes of the interpreted wireframes.

All drillhole assay samples have been uniquely flagged according to the mineralisation domains. All drillholes are composited to 1m downhole using a best-fit methodology and 0.5m minimum threshold on inclusions. A small number of residual composites have been retained for use in the estimation.

The geological model is comprised of one main domain (5000), and a smaller sub-vertical domain (1500). Additionally, smaller discontinuous hanging-wall and foot-wall domains are located either side of the Main Lode as outlined below (Table 2) and shown in Figure 4.

Domain Type	Domain Code	Description
Trident West Open Pit	1000	Open Pit Domain
Steep Shear Zones	1700	Hanging Wall Splay 1
	1600	Hanging Wall Splay 2
	1500	Sub-vertical shear zone
Shallow Dipping Zones	5200	Hanging Wall Zone 1
	5100	Hanging Wall Zone 2
	5000	Main Lode
	4900	Footwall Zone 1
	4800	Footwall Zone 2

Table 2 – Trident Underground Estimation Domains

The distribution of gold grades within the mineralised lenses is highly variable and is characterised by distinct cohesive regions of higher tenor gold grades. Whilst these higher-grade zones appear reasonably cohesive, they are manifested by a high-degree of short-scale variability, making it difficult to manually interpret constraining domains. These internal; high-grade regions are often surrounded by peripheral regions of lower grade mineralisation that are also highly variable.

Raw Coefficients of Variation (CoV) are typically in the order of 2.2 to 4.2, indicating moderate to high grade variability.

This moderate to high grade variability and complex spatial continuity of high grades at the Trident deposit requires a pseudo non-linear approach to deal with these high grades during estimation. Categorical Indicator Kriging (CIK) has been utilised for the estimation of all 9 domains:

Prior to estimation, a reference surface for each domain has been exported from Leapfrog. This is calculated as the best fit surface using the hangingwall and footwall surfaces. The reference surface is then imported into Surpac and the dip and dip-direction of each triangle facet is imported into the Surpac block model to provide information for dynamic search and variogram model orientation during interpolation. Dynamic anisotropy is applied for the estimation of the CIK indicators, and the gold grades for all estimation domains.

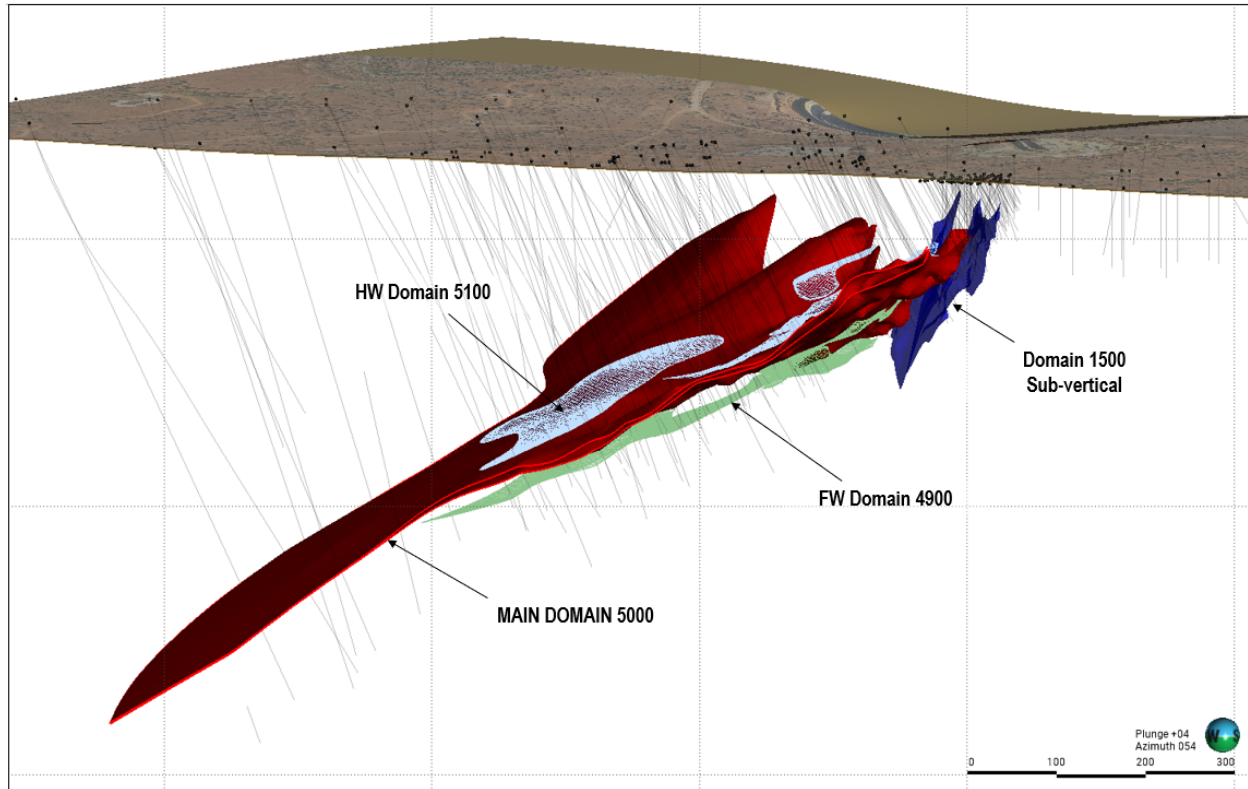


Figure 4 - Trident Estimation Domains – Isometric view looking northeast

Categorical Indicator Kriging Workflow

The estimation method applied to the CIK domains utilises a probabilistic method to define internal estimation sub-domains, together with applying distance limiting at chosen grade thresholds, to restrict the influence of the high grade and extreme grade values during the grade estimation.

Two Categorical Indicator values are determined for the CIK domains:

- A low-grade (LG) indicator assigned to differentiate between background ‘waste’ and low-tenor mineralisation.
- A high-grade (HG) indicator assigned to define broad areas of consistent higher-tenor mineralisation.

The indicators applied for each domain are summarised below:

Domain Code Groups	Indicator (Au g/t)	Description
1500, 1600, 1700, 1000	0.4	Low grade
	1.8	High grade
5100, 5200	0.4	Low grade
	1.2	High grade
5000	0.4	Low grade
	1.4	High grade
4800, 4900	0.4	Low grade
	1.2	High grade

Table 3 – Trident Estimation Indicators

Variograms were generated for the LG and HG indicators for Domain 1500, 5000, and 5100 and then applied to the other domains in the respective group (i.e. shallow dipping and steep dipping).

The CIK indicators have been estimated using OK into a finely gridded block model with block dimensions of 1.25m x 1.25m x 1.25m. The small block size for the indicator process is beneficial for creating categorical sub-domains at a resolution which can be used to accurately back-flag composite data.

Three categorical sub-domains have been generated: low-grade (LG), medium-grade (MG) and high-grade (HG) areas. The HG sub-domain has been based on an indicator probability threshold of 0.4 and the LG sub-domain has been based on an indicator probability threshold of 0.6. The MG sub-domain is assigned to blocks that do not satisfy either the HG or LG sub-domain criteria.

The three categorical block model sub-domains (HG, MG and LG) have been used to 'back-flag' the 1m composites, creating a separate composite file for each sub-domain.

Assay top-cuts have been applied to the sub-domain composite files with the following values:

Domain Code Groups	LG cut (Au g/t)	MG cut (Au g/t)	HG cut (Au g/t)
1500, 1600, 1700, 1000	0.4	6	100
5100, 5200	0.45	1.4	30
5000	2	4	140
4800, 4900	0.4	1.2	40

Table 4 – Trident Assay Top Cuts

The assay top-cuts are generally above the 99th percentile of the distribution and have been aimed at globally limiting extreme values only. Top-cuts are not used as the primary tool to control metal risk. The use of grade thresholds and distance limiting is considered a more objective and influential method in controlling metal risk, while better reflecting the actual localised occurrence of discontinuous high-grade gold mineralisation.

Variograms were generated for the LG and HG indicators for Domain 1500, 5000, and 5100 and then applied to the other domains in the respective group (i.e. shallow dipping and steep dipping).

Grade thresholds for distance limiting have been determined for each domain using log-probability plots and visual inspection. The final applied grade distance limits applied are:

Domain Code	Indicator	Grade Threshold (Au g/t)	Distance Threshold (metres)
1500, 1600, 1700, 1000	LG	0.3	25
	MG	1.8	25
	HG	25	25
5100, 5200	LG	0.4	25
	MG	1	25
	HG	16	25
5000	LG	0.4	25
	MG	1.8	25
	HG	14	25
4800, 4900	LG	0.3	25
	MG	0.9	25
	HG	18	25

Table 5 – Trident Distance Limiting Thresholds

Prior to grade estimation, sub-domain codes from the 1.25m resolution block model have been imported into a 2.5m x 2.5m x 2.5m resolution block model and the proportion of LG, MG and HG is calculated for each 2.5m block.

Grade estimation for the LG, MG and HG domains has been undertaken in Surpac software using OK with grade threshold distance limiting applied by domain. Search ellipse orientation and variogram orientations are drawn from the pre-populated dynamic anisotropy information recorded in each block.

Final block grades at a 2.5m x 2.5m x 2.5m block resolution have been calculated by weighting the estimated grades for each sub-domain by the relevant domain proportion. The parent estimation block size is 2.5m x 2.5m x 2.5m. A minimum of 2 and maximum of 12 composites have been used for each sub-domain estimate per block. It is possible that up to 36 composites can be used to estimate a parent block where there is a proportion of all three sub-domains present.

A single-pass estimation strategy has been used across all domains, whereby a search range of 100m by 100m by 50m and minimum 2 to maximum 12 composites has been utilised to estimate a grade. Octant restrictions have not been used.

Block model validation was completed to check that the grade estimates within the block model are an appropriate reflection of the underlying composite sample data, and to confirm that the interpolation parameters have been applied as intended. Checks of the estimated block grade with the corresponding composite dataset have been completed using several approaches involving both numerical and spatial aspects as follows:

- Semi-Local: Using swath plots in X, Y and Z directions comparing the estimates to the sample data.
- Local: Visual inspection of the estimated block grades viewed in conjunction with the sample data.

The final grade model for Main Domain 5000 in relation to the informing drill pierce points are shown in plan view in Figure 5.

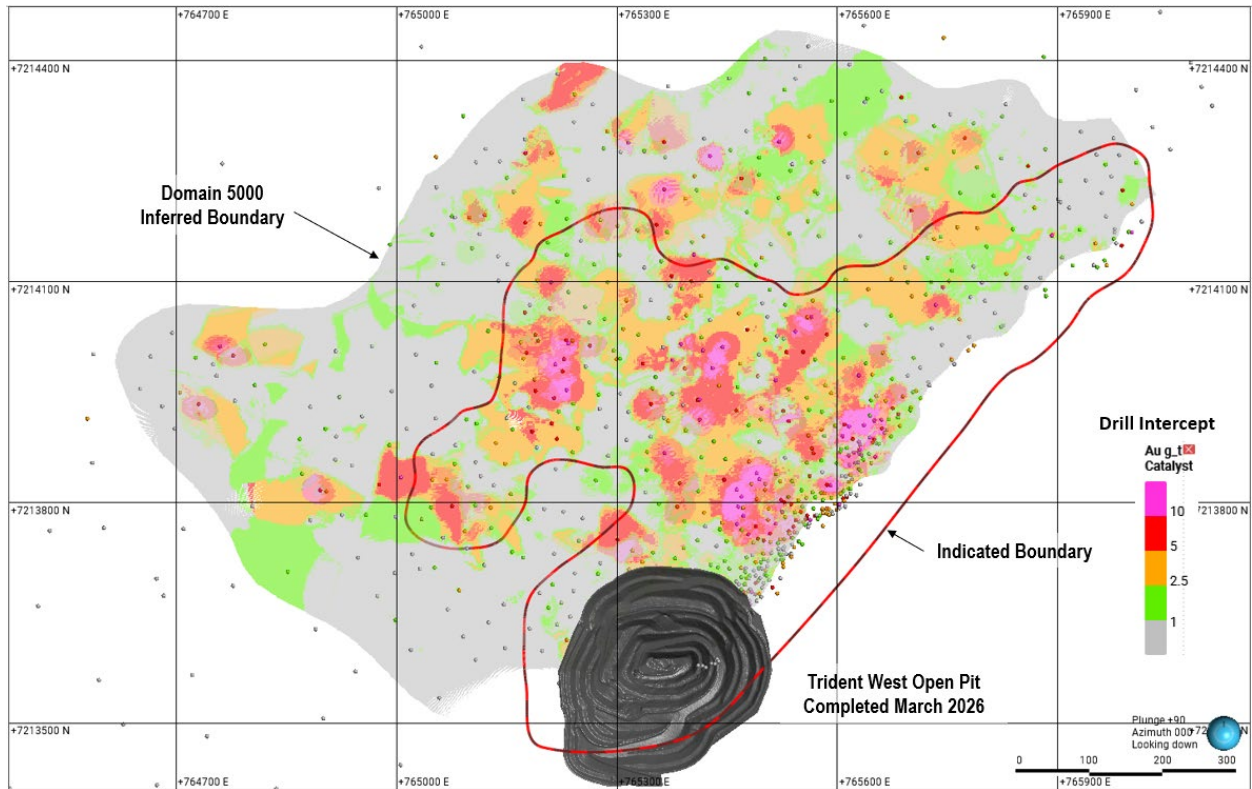


Figure 5 - Trident Grade Model (Au g/t) with Informing Compositing Drill Intercepts - Main Domain

Bulk Density

Density has been assigned to the resource using interpreted weathering surfaces determined from drillhole logging using water immersion density measurements from the diamond drill core. The material present within the Trident MRE is all located beneath the Top of Fresh Rock (TOFR) surface, and therefore all material is coded as fresh rock with a density of 2.9 t/m³.

Classification Criteria

Mineral Resources were classified as Indicated and Inferred to appropriately represent confidence and risk with respect to data quality, drillhole spacing, geological and grade continuity and mineralisation volumes. Additional considerations were the stage of project assessment, amount of DD and RC drilling undertaken, current understanding of mineralisation controls and mining selectivity within an open pit mining environment.

The drilling, surveying and sampling undertaken, and analytical methods and quality controls used, are appropriate for the style of deposit under consideration.

Indicated Mineral Resources were defined where a moderate level of geological confidence in geometry, continuity and grade was demonstrated, and were identified as areas where:

- The portions of the Trident MRE classified as Indicated have been flagged in areas of the block model where the average distance to the nearest informing compositing drill intercept was nominally 30m or closer. This drill data spacing within the Indicated portion of the resource is considered appropriate for defining the continuity and volume of the mineralised drillhole spacing on nominal 25m drill spacing on 25m sections.

- Blocks were interpolated with a neighbourhood largely informed by the maximum number of samples.

Inferred Mineral Resources were defined where a low to moderate level of geological confidence in geometry, continuity and grade was demonstrated, and were identified as areas where:

- The portions of the MRE classified as Inferred typically represent minor lodes or lower grade portions of the larger domains where geological continuity is present but not consistently confirmed by 30m spaced composited drill data.
- The Inferred Resources were interpolated where the average distance to the nearest composited drill intercept was nominally 60m or closer. The extrapolation of the resource was limited to a maximum of 60m past the last sample point.

Further considerations of resource classification include;

- data type and quality
 - drilling type
 - drilling orientations
 - down hole surveys
 - sampling, and
 - assaying methods
 - geological confidence and geostatistical considerations.

Mineralisation within the block model which did not satisfy the criteria for classification as Mineral Resources remained unclassified.

The delineation of Indicated and Inferred Mineral Resources appropriately reflects the Competent Person's view on continuity and risk at the deposit and is shown in Figure 6.

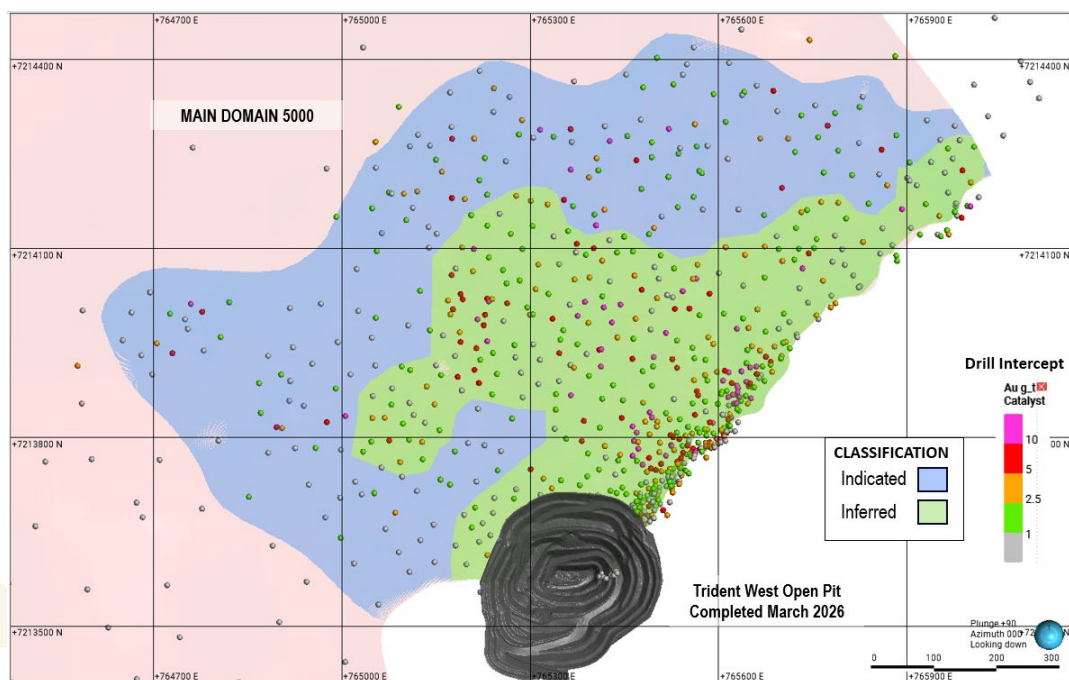


Figure 6 - Trident MRE Classification with Drill Pierce Points for the Main Domain

Cut-off Grade

The Trident Underground Mineral Resources is reported at a cut-off grade of 2.0 g/t Au.

The cut-off grade has been derived from current mining and processing costs and metallurgical parameters at the nearby Plutonic Underground Mine. Inputs into the cut-off grade calculation include:

- Average Mining Cost = AUD\$161/t
- Processing and Other Costs = AUD\$24/t ore
- Metallurgical Recovery = 83.5%
- Royalties = 2.5%
- Gold Price = AUD\$3,500/oz

In addition to applying a cut-off grade of 2.0 g/t Au, the MRE has been limited to a maximum vertical depth of 550m below the topographic surface.

The grade-tonnage curve for the Trident classified resource for all domains is shown in Figure 7.

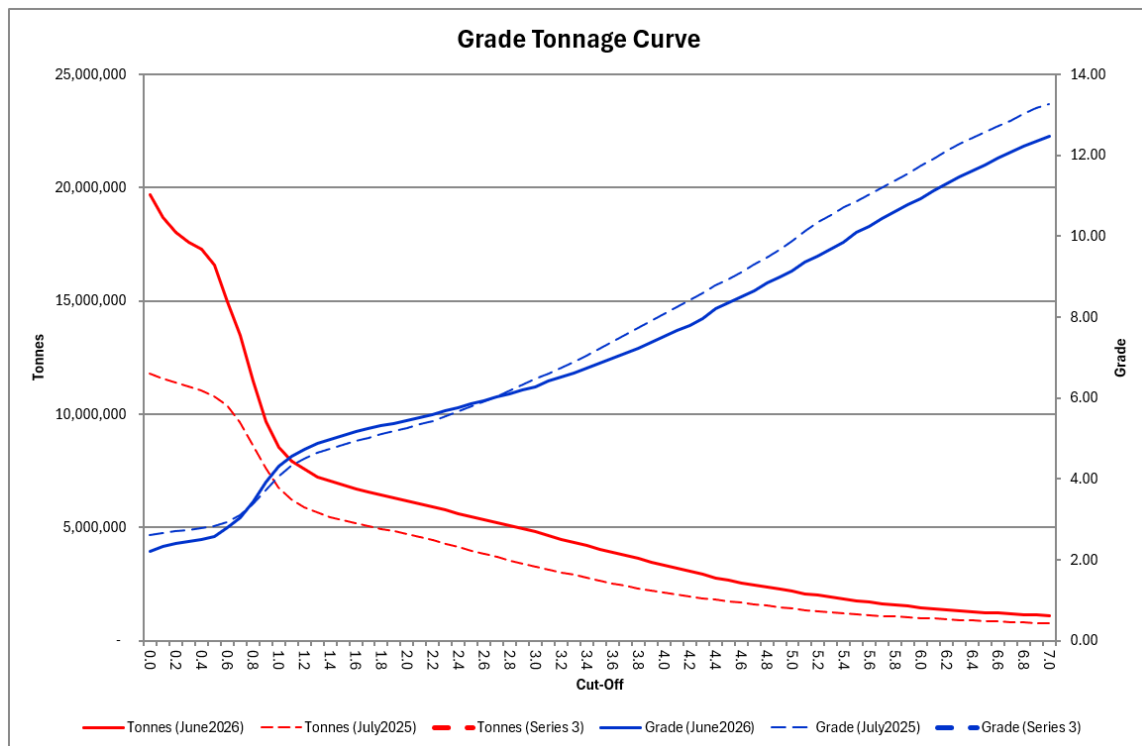


Figure 7 - Trident MRE June 2026 vs MRE July 2025 Grade and Tonnage Curve – Indicated and Inferred

Assessment of Reasonable Prospects for Eventual Economic Extraction

The Trident Underground Mineral Resource has been undertaken with a focus on delineating areas of the MRE with Reasonable Prospects for Eventual Economic Extraction (RPEEE) by underground mining methods.

The following criteria were considered in determining the RPEEE:

- Deposit scale
- Geometry, in particular mineralisation true width and minimum mining width
- Proximity to ground surface
- Established ORE/existing mine plan for material above the Underground Resource

- Access to existing critical infrastructure
- Current gold price

The MRE has been constrained to a maximum vertical depth of 550m below the topographic surface.

The Mineral Resource is considered to have reasonable prospects for eventual economic extraction) given the access to critical infrastructure, the volume and grade of mineralisation available for mining and the RPEEE criteria which have been applied prior to reporting the Mineral Resource.

Mining and Depletion

No underground mining at Trident has been undertaken in the past.

Metallurgy

The Plutonic Underground Gold Mine is an operating mine and there are no material metallurgical issues that are known to exist. No metallurgical recovery factors were applied to the Mineral Resources or resource tabulations.

This announcement has been approved for release by the Board of Directors of Catalyst Metals Limited.

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Competent person's statement

The information in the announcements to which this Mineral Resource Statement is attached that relates to the estimation and reporting of gold Mineral Resources at the Trident deposit is based on information compiled by Mr Andrew Finch, BSc, a Competent Person who is a current Member of Australian Institute of Geoscientists (MAIG 3827). Mr Finch, Geology Manager, at Catalyst Metals Ltd has sufficient experience relevant to the style of mineralisation and deposit type under consideration and to the activities being undertaken 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 Finch consents to the inclusion in the report of matters based on his information in the form and context in which it appears.

JORC 2012 Mineral Resources, Ore Reserves

The information in this announcement that relates to a Catalyst prior exploration results, production targets, estimates of ore reserves and mineral resources are extracted from ASX announcements referenced and available on the Company website www.catalystmetals.com.au and the ASX website (ASX code: CYL).

Catalyst confirms that it is not aware of any new information or data that materially affects the information included in the original market announcements and that all material assumptions and technical parameters underpinning the estimates in the relevant market 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 announcement.

Catalyst confirms that all material assumptions underpinning the production target, or the forecast financial information derived from a production target, in the initial announcement continue to apply and have not materially changed.

Section 1 Sampling Techniques and Data

Trident Deposit

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

Criteria	Commentary
Sampling techniques	<ul style="list-style-type: none"> This release relates to the Mineral Resource Estimate (MRE) update for the Trident Underground Deposit. All drilling data completed by Catalyst since the previous MRE (July 2025) to the current data cut-off date (4 June 2026) has been incorporated. The majority of holes have been drilled using a combination of RC pre-collars to an average depth of 220m, followed by DD tails to a maximum depth of 699.7m. A total of 111,001m of drilling from 230 diamond/diamond tails and 110 RC drillholes form the basis of this Mineral Resource Update. This included 72 RC grade control holes for 11,221m to an average depth of 155m. Reverse Circulation (RC pre-collars) through the overlying granite cover rocks have been sampled using 3m composited samples from the original 1m samples from the rig mounted cyclone. Pre-collars located within the Resource footprint have not been sampled through the barren granite cover rocks. DD tails have been sampled using NQ2 half core through the ultramafic host rocks at 1 m intervals or to geological boundaries For DD samples, downhole depth is recorded by the drillers on core blocks after every run. This is checked and compared to the measurements of the core by a geologist to honour geological boundaries (lithology, mineral assemblage, alteration etc). Sample lengths typically vary between 0.2m and 1.6m.
Drilling techniques	<ul style="list-style-type: none"> Reverse Circulation drilling has been conducted utilizing a 5.75 inch face sampling bit. Diamond drilling utilises NQ core with a diameter of 47.6 mm.
Drill sample recovery	<ul style="list-style-type: none"> All holes have been logged on site by an experienced geologist. The core is jig-sawed back together and metre marked carefully. Discrepancies to core blocks have been brought up with the drill contractor. Occasionally core loss blocks are inserted. Core recovery for the diamond drilling is based on the measured core returned for each 3m run. Overall drill core recovery is very good, with an average recovery of 99% through the mineralised zones. RC drilling has been bagged on 1m intervals and an estimate of sample recovery made on the size of each sample. There is no known relationship between sample recovery and grade at Trident.
Logging	<ul style="list-style-type: none"> All RC pre-collars have been logged on 1m intervals. DD samples have been logged by a qualified geologists to a level of detail to support appropriate Mineral Resource estimation, mining studies and metallurgical studies. Logging is both qualitative and quantitative. Logging records include: depth from, depth to, lithology, texture, colour, alteration style, alteration intensity, alteration mineralogy, sulphide (percentage and type), quartz (percentage), veining, and general comments. Orientated core structural measurements have been taken at relevant structures and where the foliation is relatively consistent. All DD core has been digitally photographed.
Sub-sampling techniques and sample preparation	<ul style="list-style-type: none"> Reverse Circulation (RC pre-collars) through the overlying granite cover rocks were sampled using 3m composited samples from the original 1m riffle split samples from the rig mounted cyclone. Pre-collars located within the Resource footprint have not been sampled through the barren granite cover rocks. Half cut diamond core was sampled on 1m intervals or to geological contacts, with sample lengths varying between 0.2m to 1.6m. Whole core sampling for Resource drilling has been instigated to resolve any possible biasing issues with 1/2 core grab samples in broken/sheared zones which could not be cut. All RC composited samples were dispatched to the ALS laboratory in Perth for gold fire assay analysis. NQ core samples have been sent to several labs: the majority to ALS (Au-AA26), some to SGS (GO_FAP50V10) and several holes to Plutonic onsite lab (PAL_DIBK). Sample preparation procedures for RC samples includes: <ul style="list-style-type: none"> 1-4 hours drying at 150°C depending on moisture content Riffle splitting to obtain between 1.2 to 3kg. Pulverising to a particle size of 85% passing 75µm.

Criteria	Commentary
	<ul style="list-style-type: none"> • Sample preparation procedures for DD by Plutonic onsite lab (PAL_DIBK) includes: <ul style="list-style-type: none"> ○ 1-4 hours drying at 150°C depending on moisture content ○ Entire core sample is crushed to a particle size of 85% passing 3.2mm. ○ Riffle splitting to obtain between 300 to 600g. • Sample preparation procedures for DD by Fire Assay method includes: <ul style="list-style-type: none"> ○ 1-4 hours drying at 150°C depending on moisture content ○ Entire core sample is crushed to 10mm ○ 3kg riffle split for pulverisation ○ Pulverise to 90% passing 75µm ○ Scoop 250-300g • Sample preparation protocols and sample sizes are considered appropriate for the style of mineralisation encountered and should provide representative results.
Quality of assay data and laboratory tests	<ul style="list-style-type: none"> • Samples analysed at ALS Laboratories using a 50g Fire Assay method (Au-AA26). • Samples analysed at SGS Laboratories using a 50g Fire Assay method (GO_FAP50V10). • Fire assay gold analysis is considered to be total gold analysis. • Samples analysed at Plutonic onsite lab using PAL method (PAL_DIBK). The Pulverising and Leach (PAL) method is not considered to be a total gold analysis, however the larger sample size still produces a representative result. • Samples were dried, crushed and pulverised prior to analysis. • Certified Reference Material (CRM's) were submitted every 20 samples. CRM's are of similar grade tenor to those expected grades in the sampling and were selected based on their grade range and mineralogical properties with an emphasis on sulphide ores. • Blanks are inserted every 100 samples. • Field duplicates were inserted every 20 samples in the RC drilling. A mix of Field duplicates and crush duplicates were inserted in the DD drilling, at a combined 1 in every 20 sample rate. • Crush sizing analysis is conducted randomly by the Laboratory as part of their QC process. Pulp residues are expected to have 90% passing ≤75µm. This data is monitored by the Laboratory Supervisor. Grind times can be lengthened accordingly. • Current procedures dictate a process of validation and checking of laboratory results when data is returned by the laboratory as it is loaded into the independently managed Quest database. A standard set of plots and checks are undertaken, and if results fall outside of the expected limits, then re-assaying is requested. QAQC reports are generated by the database administrator and documented from automated routines out of the database.
Verification of sampling and assaying	<ul style="list-style-type: none"> • RC and diamond drilling data has been verified by the geologist first and then the Database Administrator before importing into the main Quest database (proprietary database system). • RC and DD logging is completed electronically on laptops. Database protocols and rules are applied upon data entry. • All drill data within site databases are regularly validated using both internal database systems and external validation tools.
Location of data points	<ul style="list-style-type: none"> • All drill collars have been accurately located using DGPS. • Downhole survey data is collected using an Axis Mining Technology Champ North Seeking Gyro tool. Surveys are undertaken on 10m intervals as the tool is removed from the drillholes once the drillhole is completed. • Downhole surveys are visually inspected for anomalous changes in drill trace, (i.e does the drillhole apparently bend inordinately).
Data spacing and distribution	<ul style="list-style-type: none"> • Drilling completed to date within the Inferred portion of the MRE is at a nominal 60m x 60m drill intercept spacing. • Comprehensive RC/DD infill drilling has been completed at a nominal 30m x 30m drill intercept spacing to confirm the grade continuity and allow conversion of a large part of the Inferred MRE to the Indicated category. • The drill spacing for the broader drilling outside of the current MRE is wide, ranging from 100m to 500m and should be considered exploratory in nature. • The purpose of the step out drilling program using nominal 200m and 500m spaced drillholes is to test for extensions to the mineralised zones and to define the extents of the mineralised system. • Sample compositing has only been used in the RC pre-collars through the granite cover rocks.
Orientation of data in relation to	<ul style="list-style-type: none"> • The orientation of a majority of the drilling is approximately perpendicular to the strike and dip of the mineralisation and is unlikely to have introduced any sampling bias. • Certain drillholes may have been drilled parallel to key structures, but density of drilling and drilling

Criteria	Commentary
geological structure	on other orientations has allowed detailed geological modelling of these structures and hence any sampling bias in a single drillhole has been removed.
Sample security	<ul style="list-style-type: none"> The chain of custody is managed by Catalyst employees and contractors. Geologists are responsible for marking the sample intervals and placement of Blanks and CRM's within the sample lot. The Project Geologist and Senior Geologist complete quality control checks on the GC drilling data daily. Field Staff are primarily responsible for sampling of core, generating the sample numbers for core submission, creating a sample submission sheet, selecting and recording the CRM's to be sent to the laboratory and the transportation of the samples to the laboratory. Samples are tracked during shipping. Once a drillhole has been sampled, the sample intervals and checked geology documents are uploaded into the Quest database system managed by EarthSQL. The independent Database Administrator (DBA) merges the validated drilling data with the certified laboratory assay files where validation routines for QA/QC are completed before database exports and reports are issued. Catalyst samples have been stored on site and delivered to the ALS assay laboratory in Perth by a Contracted Transport Company. Consignment notes have been used to track the samples. Operator sample security is assumed to be consistent and adequate.
Audits or reviews	<ul style="list-style-type: none"> No audit or reviews of sampling techniques have been undertaken however the data is managed by company geologist who has internal checks/protocols in place for all QA/QC. Historical reviews of the database for the Trident area have been examined previously and a proportion of drillholes compared to original data sources and found to be consistent wherever checked.

Section 2 Reporting of Exploration Results

Trident Deposit

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

Criteria	Commentary
Mineral tenement and Land tenure status	<ul style="list-style-type: none"> Located in the Marymia - Plutonic Greenstone Belt ~218 km northeast of Meekatharra in the Midwest mining district in WA Trident is located in M52/217 – a granted Mining tenement in good standing. The tenement predates Native title interests but is covered by the Gingirana Native Title claim. The tenement is 100% owned by Vango Mining Limited and subsidiary Dampier (Plutonic) Pty Ltd. Gold production will be subject to a 2.5% government royalty.
Exploration done by other parties	<ul style="list-style-type: none"> Comprehensive drilling of the deposit was first undertaken by Resolute Limited from 1995 to 1998 completing approximately 263 RC and 37 DD holes. From 1999 Homestake and then later Barrick Gold (2002) completed numerous drilling campaigns at Trident. Dampier Gold completed RC and DD programs at Trident from 2012 until 2014 when Vango Mining took over the project completing 6 Diamond drillholes for 946 metres plus three RC drillholes for 747 metres. Catalyst consolidated the belt in 2023 following the successful acquisition of Vango Mining and the merger with Superior Gold Inc. this was followed by more diamond drilling, completed in 2023. In 2024 an RC program was completed which has been included in this MRE.
Geology	<ul style="list-style-type: none"> Gold mineralisation at Trident is orogenic, hosted within a sheared contact zone in ultramafic rocks. Shallow plunging high grade 'shoots' of mineralisation are associated with flexures in the mineralised host shear zones combined with steeply dipping intersecting structures. The mineralisation consists of multiple stacked zones, with the main mineralised domain extending along a northeast/southwest strike for 1,100m. The system dips at 30° degrees towards 330° and extends down dip for at least 1,000m and remains open at depth. Mineralised zones can vary in width from 0.6m up to 15m with an average thickness of around 4.5m. The higher grade zones are focussed along north-south structures and multiple north-east trending shoots and at the interaction points of these two dominant trends. The orientation of mineralisation can be variable particularly where cross-cutting structures are intersected. These inflexion points are likely dilation zones which can host thicker and higher grade mineralisation intersections.

Criteria	Commentary
	<ul style="list-style-type: none"> An overthrust granite package forms the barren hangingwall to the mineralisation hosted within the sheared ultramafic host rock package. The mineralised zones are characterised by biotite-phlogopite alteration with a sulphide assemblage of pyrite-pyrrhotite-chalcopyrite-arsenopyrite.
Drill hole Information	<ul style="list-style-type: none"> As all available drill hole information has been incorporated into the Mineral Resource, the drill intercepts have not been reported separately.
Data aggregation methods	<ul style="list-style-type: none"> As all available drill hole information has been incorporated into the Mineral Resource, the drill intercepts have not been reported separately.
Relationship between mineralisation widths and intercept lengths	As all available drill hole information has been incorporated into the Mineral Resource, the drill intercepts have not been reported separately.
Diagrams	<ul style="list-style-type: none"> As all available drill hole information has been incorporated into the Mineral Resource, the drill intercepts have not been reported separately.
Balanced reporting	<ul style="list-style-type: none"> As all available drill hole information has been incorporated into the Mineral Resource, the drill intercepts have not been reported separately.
Other substantive exploration data	<ul style="list-style-type: none"> No additional exploration data is included in this release.
Further work	<ul style="list-style-type: none"> Resource definition, infill and extensional drilling programs are ongoing, and will continue in line with mine development requirements.

Section 3 Estimation and Reporting of Mineral Resources

Trident Deposit

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

Criteria	Commentary
Database integrity	<ul style="list-style-type: none"> The Catalyst drillhole database is managed by a third party consultant EarthSQL in Acquire software. Trident data has been exported to MS Access and audited before estimation. Various validation checks in GEOVIA Surpac™ and Seequent Leapfrog Geo™ 3D software and data queries in MS Access were undertaken such as overlapping samples, duplicate entries, missing data, sample length exceeding hole length, unusual assay values and a review of below detection limit samples. A visual examination of the data was also completed to check for erroneous downhole surveys. The data validation process identified no major drill hole data issues that would materially affect the MRE outcomes. Database checks included the following: <ul style="list-style-type: none"> Checking for duplicate drillhole names and duplicate coordinates in the collar table. Checking for missing drillholes in the collar, survey, assay and geology tables based on drillhole names. Checking for survey inconsistencies including dips and azimuths <0°, dips >90°, azimuths >360° and negative depth values.
Site visits	<ul style="list-style-type: none"> The Competent Person undertakes frequent site visits to the Plutonic Gold Operation and associated Marymia tenements.
Geological interpretation	<ul style="list-style-type: none"> Gold mineralisation at Trident is orogenic, hosted within a sheared contact zone in ultramafic rocks. Shallow plunging high grade 'shoots' of mineralisation are associated with flexures in the mineralised host shear zones combined with steeply dipping intersecting structures. The mineralisation consists of multiple stacked zones, with the main mineralised domain extending along a northeast/southwest strike for 1,200m. The system dips at 30° degrees towards 330° and extends down-dip for up to 1,000m and has been mostly closed out at depth. Mineralised zones can vary in width from 0.6m up to 15m with an average thickness of around 4.5m. A total of 111,001m of drilling from 230 diamond/diamond tails and 110 RC drillholes were available for interpretation and estimation of the MRE. This included 72 RC grade control holes for 11,221m to an average depth of 155m. Drilling density at Trident is variable, ranging from <20m x 20m close to surface, to >40m x 40m at depth. Mineralisation domains have been interpreted primarily using geological logging and downhole geological contacts, based on lithology, gold grade distribution, major fault location and geometry. Interpretation of the mineralisation has been undertaken in Seequent Leapfrog Geo™ software using

Criteria	Commentary																							
	<p>all available drillholes. Intercepts correlating to gold mineralisation and underpinned by strike continuity have been independently identified and manually selected within Seequent Leapfrog Geo™ prior to creation of implicit volume models.</p> <ul style="list-style-type: none"> The geological model is comprised of one main domain (5000) and a smaller sub-vertical domain (1500) located close to the Top of Fresh Rock. Additionally, smaller discontinuous hanging-wall domains (5100 and 5200) and smaller foot-wall domains (4800 and 4900) are located either side of the Main Lode (5000). The mineralisation shallowly dips to the northwest with a thickness of 1m to 15 m. Mineralisation has been delineated at a nominal 0.5g/t Au cut-off. Catalyst considers confidence is moderate to high in the geological interpretation and continuity of the mineralisation domains. 																							
Dimensions	<ul style="list-style-type: none"> Mineralisation extends over a strike length of approximately 1,200 m and down-dip for up to 1,000 m. 																							
Estimation and modelling techniques	<ul style="list-style-type: none"> Interpretations used all available drillhole data. The domain intercepts have been imported into the MS Access database and prepared for compositing in Surpac. No deleterious elements have been estimated or assumed. Only gold grade has been estimated. The model has been validated by comparing statistics of the estimated blocks against the composited sample data, and visual examination of the of the block grades versus the assay data in section and swath plots. <p>CIK Methodology</p> <ul style="list-style-type: none"> All drillhole assay data has been composited to 1m downhole using the best fit algorithm in Surpac. The moderate to high grade variability and lack of spatial continuity of high grades requires a non-linear approach to deal with the high grades during estimation. A traditional approach of physical domaining, assay cutting, and linear estimation (IDW or OK) is considered inadequate in dealing with this complexity. The estimation method combines Categorical Indicator Kriging (CIK) to define broad estimation domains, together with applying distance limiting at chosen grade thresholds to restrict the influence of the high-grade values during grade interpolation. Prior to estimation, a reference surface for the domain has been exported from Leapfrog. This is calculated as the best fit surface using the hanging wall and footwall surfaces. The reference surface has been imported into Surpac and a dip and dip-direction value of each triangle facets is imported into the Surpac block model to provide information for dynamic search and variogram model orientation during interpolation. Dynamic estimation has been applied for estimating the CIK indicators and gold grades. Two Categorical Indicator values have been determined: <ul style="list-style-type: none"> A low-grade (LG) indicator was assigned to differentiate between background ‘waste’ and low-tenor mineralisation. A high-grade (HG) indicator was assigned to define broad areas of consistent higher-tenor mineralisation. The indicators applied for each domain are summarised below: <table border="1"> <thead> <tr> <th>Domain Code Groups</th> <th>Indicator (Au g/t)</th> <th>Description</th> </tr> </thead> <tbody> <tr> <td rowspan="2">1500, 1600, 1700, 1000</td> <td>0.4</td> <td>Low grade</td> </tr> <tr> <td>1.8</td> <td>High grade</td> </tr> <tr> <td rowspan="2">5100, 5200</td> <td>0.4</td> <td>Low grade</td> </tr> <tr> <td>1.2</td> <td>High grade</td> </tr> <tr> <td rowspan="2">5000</td> <td>0.4</td> <td>Low grade</td> </tr> <tr> <td>1.4</td> <td>High grade</td> </tr> <tr> <td rowspan="2">4800, 4900</td> <td>0.4</td> <td>Low grade</td> </tr> <tr> <td>1.2</td> <td>High grade</td> </tr> </tbody> </table> <ul style="list-style-type: none"> Variograms were generated for the LG and HG indicators for Domain 1500, 5000, and 5100 and then applied to the other domains in the respective group (i.e. shallow dipping and steep dipping). The CIK indicators have been estimated using Ordinary Kriging into a finely gridded block model with block dimensions of 1.25m x 1.25m x 1.25m. The small block size for the indicator process is beneficial for creating categorical sub-domains at a resolution which can be used to accurately back-flag 	Domain Code Groups	Indicator (Au g/t)	Description	1500, 1600, 1700, 1000	0.4	Low grade	1.8	High grade	5100, 5200	0.4	Low grade	1.2	High grade	5000	0.4	Low grade	1.4	High grade	4800, 4900	0.4	Low grade	1.2	High grade
Domain Code Groups	Indicator (Au g/t)	Description																						
1500, 1600, 1700, 1000	0.4	Low grade																						
	1.8	High grade																						
5100, 5200	0.4	Low grade																						
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Criteria	Commentary																																																																
	<p>composite data.</p> <ul style="list-style-type: none"> Three categorical sub-domains have been generated: low-grade (LG), medium-grade (MG) and high-grade (HG) areas. The HG sub-domain has been based on an indicator probability threshold of 0.4 and the LG sub-domain has been based on an indicator probability threshold of 0.6. The MG sub-domain is assigned to blocks that do not satisfy either the HG or LG sub-domain criteria. The three categorical block model sub-domains (HG, MG and LG) have been used to 'back-flag' the 1m composites, thus creating a separate composite file for each sub-domain. Assay top-cuts have been applied to the sub-domain composite files as follows: <table border="1"> <thead> <tr> <th>Domain Code Groups</th> <th>LG cut (Au g/t)</th> <th>MG cut (Au g/t)</th> <th>HG cut (Au g/t)</th> </tr> </thead> <tbody> <tr> <td>1500, 1600, 1700, 1000</td> <td>0.4</td> <td>6</td> <td>100</td> </tr> <tr> <td>5100, 5200</td> <td>0.45</td> <td>1.4</td> <td>30</td> </tr> <tr> <td>5000</td> <td>2</td> <td>4</td> <td>140</td> </tr> <tr> <td>4800, 4900</td> <td>0.4</td> <td>1.2</td> <td>40</td> </tr> </tbody> </table> <ul style="list-style-type: none"> The assay top-cuts are generally above the 99th percentile of the distribution and have been aimed at globally limiting extreme values only. In this estimation methodology, top-cuts are not used as the primary tool to control metal risk. The use of grade thresholds and distance limiting is considered a more objective and influential method in controlling metal risk, while better reflecting the actual localised occurrence of discontinuous high-grade gold mineralisation. Grade thresholds for distance limiting have been determined from log-probability plots and detailed review of the spatial locations of the gold grades. The final applied grade distance limits applied were: <table border="1"> <thead> <tr> <th>Domain Code</th> <th>Indicator</th> <th>Grade Threshold (Au g/t)</th> <th>Distance Threshold (metres)</th> </tr> </thead> <tbody> <tr> <td rowspan="3">1500, 1600, 1700, 1000</td> <td>LG</td> <td>0.3</td> <td>25</td> </tr> <tr> <td>MG</td> <td>1.8</td> <td>25</td> </tr> <tr> <td>HG</td> <td>25</td> <td>25</td> </tr> <tr> <td rowspan="3">5100, 5200</td> <td>LG</td> <td>0.4</td> <td>25</td> </tr> <tr> <td>MG</td> <td>1</td> <td>25</td> </tr> <tr> <td>HG</td> <td>16</td> <td>25</td> </tr> <tr> <td rowspan="3">5000</td> <td>LG</td> <td>0.4</td> <td>25</td> </tr> <tr> <td>MG</td> <td>1.8</td> <td>25</td> </tr> <tr> <td>HG</td> <td>14</td> <td>25</td> </tr> <tr> <td rowspan="3">4800, 4900</td> <td>LG</td> <td>0.3</td> <td>25</td> </tr> <tr> <td>MG</td> <td>0.9</td> <td>25</td> </tr> <tr> <td>HG</td> <td>18</td> <td>25</td> </tr> </tbody> </table> <ul style="list-style-type: none"> Prior to grade estimation, sub-domain codes from the 1.25m resolution block model have been imported into a Y=2.5m x X=2.5m x 2.5m resolution block model and the proportion of LG, MG and HG is calculated for each 2.5m x 2.5m x 2.5m block. The choice of a 2.5m x 2.5m x 2.5m block size reflects the likely mining selectivity achievable for an underground mining scenario. Grade estimation for the LG, MG and HG sub-domains has been undertaken in Surpac software using Ordinary Kriging (OK) with grade threshold distance limiting. Search routines and variogram orientations are drawn from the pre-populated dynamic search information recorded in each block (dynamic anisotropy). Final block grades at a 2.5m x 2.5m x 2.5m block resolution have been calculated by weighting the estimated grades for each sub-domain by the relevant domain proportion. The parent estimation block size is 2.5m x 2.5m x 2.5m. A minimum of 2 and maximum of 12 composites have been used for each sub-domain estimate per block. It is possible that up to 36 composites can be used to estimate a parent block where there is a proportion of all three sub-domains present. A single-pass estimation strategy has been used across all domains, whereby a search range of 100m by 100m by 50m and minimum 2 to maximum 12 composites has been utilised to estimate a grade. Octant restrictions have not been used. 	Domain Code Groups	LG cut (Au g/t)	MG cut (Au g/t)	HG cut (Au g/t)	1500, 1600, 1700, 1000	0.4	6	100	5100, 5200	0.45	1.4	30	5000	2	4	140	4800, 4900	0.4	1.2	40	Domain Code	Indicator	Grade Threshold (Au g/t)	Distance Threshold (metres)	1500, 1600, 1700, 1000	LG	0.3	25	MG	1.8	25	HG	25	25	5100, 5200	LG	0.4	25	MG	1	25	HG	16	25	5000	LG	0.4	25	MG	1.8	25	HG	14	25	4800, 4900	LG	0.3	25	MG	0.9	25	HG	18	25
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Moisture	<ul style="list-style-type: none"> All estimations have been carried out on a 'dry' basis. 																																																																
Cut-off parameters	<ul style="list-style-type: none"> The Trident Underground Mineral Resources is reported at a cut-off grade of 2.0 g/t Au. The cut-off grade has been derived from current mining and processing costs and metallurgical parameters at the nearby Plutonic Underground Mine. Inputs into the cut-off grade calculation include: <ul style="list-style-type: none"> Average Mining Cost = AUD\$161/t Processing and Other Costs = AUD\$24/t ore 																																																																

Criteria	Commentary
	<ul style="list-style-type: none"> ○ Metallurgical Recovery = 83.5% ○ Royalties = 2.5% ○ Gold Price = AUD\$3,500/oz
Mining factors or assumptions	<ul style="list-style-type: none"> • The Trident Underground Mineral Resources is reported to a maximum vertical depth of 550m below the topographic surface using a gold price of AUD\$3,500/oz. • Mining selectivity is assumed to include a minimum mining width of 1.5m, minimum stope length of 5m, and a minimum stope height of 5m.
Metallurgical factors or assumptions	<ul style="list-style-type: none"> • It is assumed the material will be trucked and processed at the Plutonic Gold Plant. Recovery factors are assigned based on lab test work. • No metallurgical assumptions have been built or applied to the resource model.
Environmental factors or assumptions	<ul style="list-style-type: none"> • A conventional storage facility is used for the process plant tailings. • The small amount of waste rock is stored in a traditional waste rock landform 'waste dump'. Due to low sulphide content and the presence of carbonate alteration the potential for acid content is considered low.
Bulk density	<ul style="list-style-type: none"> • Density has been assigned to the resource models using interpreted weathering surfaces determined from drill hole logging. • The entire Trident UG Mineral Resource is located below the Top of Fresh Rock (TOFR) and all blocks have been assigned a density of 2.9 kg/m³.
Classification	<ul style="list-style-type: none"> • Factors considered when classifying the model include: <ul style="list-style-type: none"> ○ The portions of the Trident MRE classified as Indicated have been flagged in areas of the block model where the average distance to the nearest informing composited drill intercept was nominally 30m or closer. ○ The portions of the MRE classified as Inferred typically represent minor lodes or lower grade portions of the larger domains where geological continuity is present but not consistently confirmed by 30m spaced composited drill data. The extrapolation of the resource was limited to a maximum of 60m past the last sample point. ○ Further considerations of resource classification include; data type and quality (drilling type, drilling orientations, down hole surveys, sampling and assaying methods); geological mapping and understanding. • The Mineral Resource estimate appropriately reflects the view of the Competent Person.
Audits or reviews	<ul style="list-style-type: none"> • The geological interpretation, estimation parameters and validation of the resource model was peer reviewed by Catalyst staff. • No external reviews of the resource estimate had been carried out at the time of writing.
Discussion of relative accuracy/ confidence	<ul style="list-style-type: none"> • The relative accuracy of the Mineral Resource estimate is reflected in the reporting of the Mineral Resource as per the guidelines of the 2012 JORC Code. • The statement relates to the global estimates of tonnes and grade.

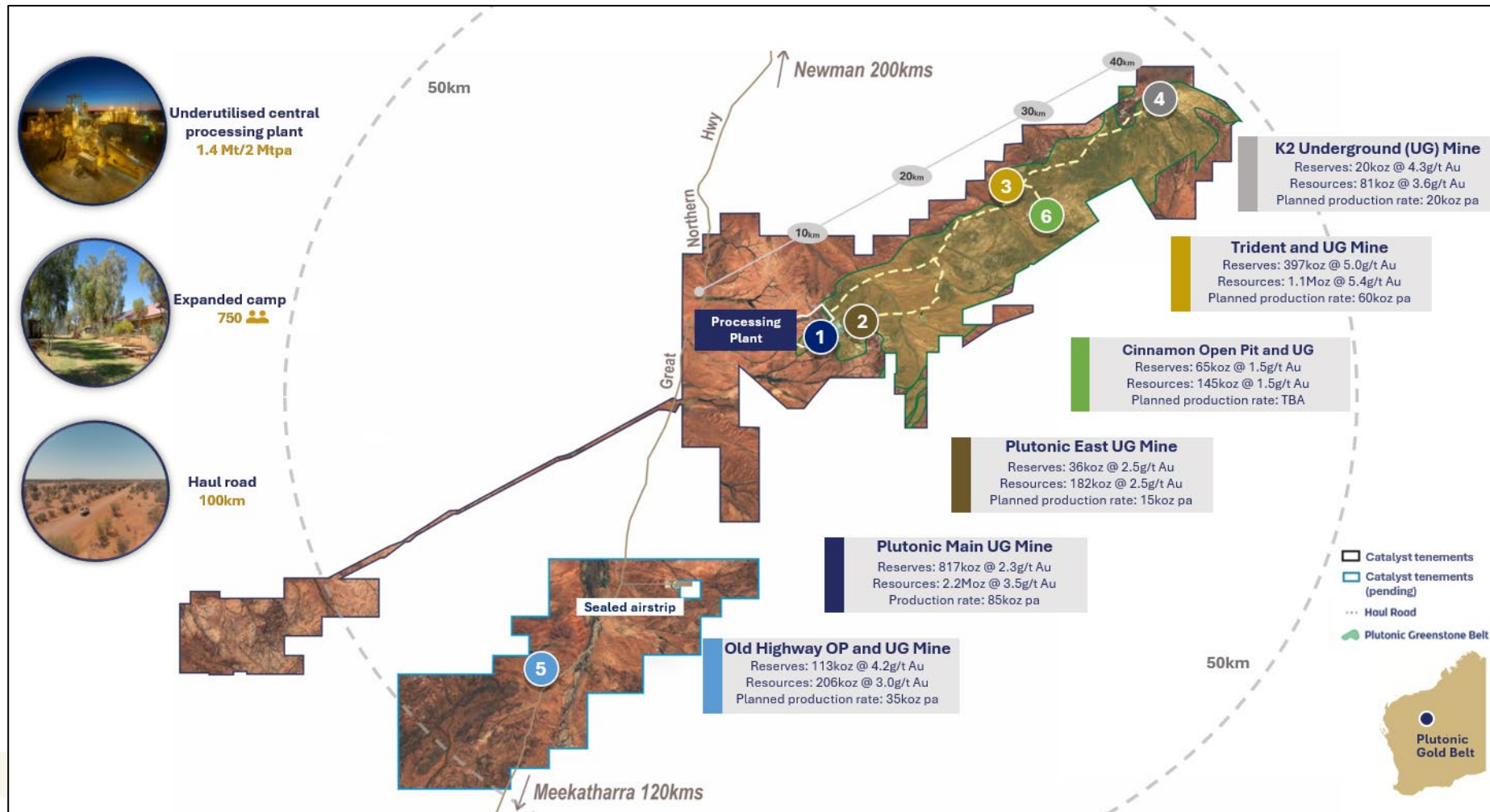


Figure 5: Plutonic Gold Belt showing location of Trident relative to the Plutonic processing facility