## MOOLARBEN COAL PROJECT

# APPENDIX 

Geochemical Assessment

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On Behalf of Moolarben Coal Pty Limited

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## Geochemical Assessment of the Moolarben Coal Project

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## Executive Summary

Environmental Geochemistry International Pty Ltd (EGi) were commissioned by Wells Environmental Services on behalf of Moolarben Coal Mines Pty Limited to carry out a geochemical assessment of the Moolarben Coal Project, located in the Hunter Valley, NSW, approximately 3 km east of the village of Ulan. The objective of this work was to assess the acid rock drainage (ARD), salinity and sodicity hazards associated with development of the coal resource.

Testing was carried out on representative samples of overburden from the proposed underground and open cut developments, and samples of Ulan Seam coal and washability trial rejects.

Results of ARD investigations indicate that over $90 \%$ of overburden material for open cut and underground operations is likely to be non acid forming (NAF). The remainder is expected to be potentially acid forming low capacity (PAF-LC), with a low ARD potential. No potentially acid forming (PAF) materials were identified for floor samples from the open cut, which suggests that final pit floors will not be a source of ARD. Preliminary results indicate roof and floor materials for the underground project may be PAF-LC.

Most of the coal seam samples tested were PAF-LC, indicating potential acid release from coal stockpiles and underground workings. The coal reject samples were acid forming, with export coal rejects showing the highest ARD potential.

Testing also indicates that overburden, floor and coal reject materials are likely to be nonsaline. Coal samples were moderately saline to saline.

Exchangeable sodium percentages and Emerson aggregate test (EAT) results indicate a possible sodicity hazard for topsoil, Quaternary/Tertiary alluvials and weathered Permian. Materials with sodic/dispersion potential may require treatment (with gypsum or lime) if exposed on dump surfaces or used in engineered structures.

No significant enrichment of metals/metaloids was detected in overburden, coal or reject solids.

The findings of these initial investigations have the following implications for materials management:

- Results suggest that normal run-of-mine operational blending of overburden should be sufficient to control ARD, pending confirmation with leach column testing.
- Containment of run off and leachate from coal stockpiles and underground operations may be required to monitor water quality and determine whether treatment is required. Results indicate that these waters may be saline and acidic. The sensitivity of groundwater and surface water to saline and acidic water should be investigated to determine the degree of management required. Provision for acid treatment may be
needed, which could include use of a mobile lime dosing plant to treat acid waters and broadcast application of agricultural lime.
- Rejects appear to have a higher ARD risk than other mine materials, and are likely to require specific management to control ARD. Possible approaches include underwater disposal, lime treatment, isolation from infiltration, or a combination of these.
- Materials with sodic/dispersion potential may require treatment (with gypsum or lime) if exposed on dump surfaces or used in engineered structures.
- A routine system of ARD testing should be established during operations to check the ARD potential of mine materials and allow for modification of materials management strategies if required.


### 1.0 Introduction

Environmental Geochemistry International Pty Ltd (EGi) were commissioned by Wells Environmental Services on behalf of Moolarben Coal Mines Pty Limited to carry out a geochemical assessment of the Moolarben Coal Project, located in the Hunter Valley, NSW, approximately 3 km east of the village of Ulan. The objective of this work was to assess the acid rock drainage (ARD), salinity and sodicity hazards associated with development of the coal resource. It is understood that findings from this report will be used as part of an Environmental Assessment Report (EAR).

The work carried out included the following:

- A site visit by EGi to view the project area, examine drill hole samples, gather background data, discuss the project with relevant site personnel, and scope the investigations required;
- Selection of samples in conjunction with site personnel, and assistance in organising sample crushing and splitting prior to delivery to the EGi laboratory;
- Characterisation of materials in terms of ARD potential, sodicity, salinity, and enrichment/availability of elements of environmental concern;
- Evaluation of results and completion of a technical report, detailing results and implications for mine operations and overburden management.


### 2.0 Background Data

The Moolarben project stratigraphy comprises a sequence of Permian sandstone, siltstone, mudstone, tuff and coal, of which the Ulan Seam is the only seam of economic significance. The Permian sediments appear to have been deposited in a mainly fluvial dominated environment, although worm burrows are observed in some of the sandstone horizons, suggesting occasional marine influence. Pyrite is not generally observed in the drill core. Quaternary/Tertiary alluvial erosion channels cut through portions of the Permian sequence, and could make up around $10 \%$ of the total overburden. Siderite is the main carbonate observed, and application of $10 \% \mathrm{HCl}$ to the core during the site visit showed only minor and occasional fizzing, suggesting a lack of acid neutralising minerals.

Mining will comprise both open cut and underground development operating concurrently, with all coal washed on site. The full Ulan seam will be recovered in the open cut operations, and a partial section will be recovered in the underground operations.

Open pit mining will utilise conventional truck and shovel methods. Open Cut 1 (northern pit) will be developed first, followed by Open Cut 2 (middle pit) and Open Cut 3 (Southern Pit). It is understood that the open cut operations will produce two coal products, a high ash domestic thermal product from the combined mining and washing of seams A to C, and an export thermal product from the combined mining and washing of
seams D to E . The full Ulan seam will be mined in two passes and processed separately to produce these two products.

Out of pit placement of overburden will be required during initial pit development, but it is understood that most overburden will be backfilled into the pit.

Underground mining will be carried out north of the open pit development and use longwall mining methods. The underground coal will be restricted to the D and E top section of the Ulan seam, and will also be separated into a low ash export product and a high ash domestic product. Overburden extracted as part of access drives, ventilation shafts and other underground workings will be dumped with overburden.

### 3.0 Sample Collection and Preparation

One hole from each of the three proposed open pits, and one hole from the underground development were selected for sampling. In each case holes were selected that best represented the full mine stratigraphic section.

Diamond drillholes WMLB24, WMLB15 and WMLB5 were selected to represent stratigraphy in pits Open Cut 1 (OC1), Open Cut 2 (OC2) and Open Cut 3 (OC3) respectively. The weathered upper portion $(6-15 \mathrm{~m})$ of these holes was pre-collared by open hole and chips were not available for sampling. WMLB75 was collared close to WMLB5, and chips from the upper 7 m of this hole were used to provide a full section in pit OC3, that included weathered Permian. Open hole TB103 was used to represent overburden materials expected to be extracted during development of vent shaft and access for the proposed underground operations.

Open hole chips from WMLB54 in OC1 were used to represent Quaternary/Tertiary alluvials for the open cut, and additional Quaternary/Tertiary alluvials were sampled in the top 24 m of drillhole TB103 testing underground development. The Quaternary/Tertiary alluvials are not expected to vary significantly across the site.

Sample intervals were selected based on geological descriptions provided by Moolarben project personnel and consultants. Continuous intervals of overburden were sampled in each hole to ensure the full stratigraphic section was represented. Moolarben personnel organised diamond core samples to be crushed and split, and $1-5 \mathrm{~kg}$ sub-samples were dispatched to EGi. Splits of open hole chip samples were sent to EGi without further crushing. At EGi a 300 g split of all crushed core and open chip samples was collected, and dispatched to Sydney Environmental and Soil Laboratory (SESL) for pulverising to $-75 \mu \mathrm{~m}$.

In addition to overburden samples, Ulan Seam samples from hole TB103, and two reject samples from washability trials of the Ulan Seam from hole WMLB24 were tested. The two reject samples were produced by Carbon Consulting and are expected to approximately represent the combined fine and coarse rejects from the proposed wash
plant after processing the two coal products from open cut mining.

### 4.0 Methodology

The following tests were carried out on all samples:

- $\mathrm{pH}_{1: 2}$ and electrical conductivity $(\mathrm{EC})_{1: 2}$ on deionised water extracts;
- Leco total S;
- acid neutralising capacity (ANC);
- the net acid producing potential (NAPP) calculated from total S and ANC results; and
- single addition net acid generation (NAG) test.

The following tests were carried out on selected samples:

- organic carbon single addition NAG (NAGorg) for samples with high organic carbon contents;
- kinetic NAG;
- acid buffering characteristic curve ( ABCC ) tests;
- multi-element scans of solids;
- multi-element scans of water extracts at a ratio of $1: 2(\mathrm{w} / \mathrm{w})$ solid to deionised water; and
- soluble and exchangeable cations and Emerson dispersion test

A general description of the $\mathrm{pH} / \mathrm{EC}$, total $\mathrm{S}, \mathrm{ANC}$ and NAG test methods is included in Appendix A.

High organic carbon contents ( $>7 \% \mathrm{C}$ ) can cause generation of organic acids in the NAG test, which can cause misleading low NAGpH values. The NAGorg test involves a combination of extended heating and NAG solution assay to identify and quantify any effects of organic acid generation on NAG test results.

Total sulphur assays, soluble and exchangeable cation testing, and Emerson dispersion tests were carried out by Sydney Environmental and Soil Laboratory (SESL). Multielements analyses were carried out by Genalysis Pty Ltd (Perth). All other analyses were carried out by EGi.

### 5.0 Overburden and Floor Results

## 5.1 pH and EC

The $\mathrm{pH}_{1: 2}$ and $\mathrm{EC}_{1: 2}$ tests were carried out by equilibrating crushed solid sample in deionised water for approximately 16 hours at a solid to water ratio of $1: 2(\mathrm{w} / \mathrm{w})$. This gives an indication of the inherent acidity and salinity of the waste material when initially exposed in a waste emplacement area. Results are shown in Table 1 (open cut) and 2 (underground).

Circum-neutral $\mathrm{pH}_{1: 2}$ values were recorded for all overburden samples from open cut and underground drill holes, indicating no existing acidity in these samples. $\mathrm{EC}_{1: 2}$ values were all non saline at less than $0.4 \mathrm{dS} / \mathrm{m}$, indicating a negligible salinity potential from these materials.

Ulan Seam floor samples from open pit and underground drillholes also have circumneutral $\mathrm{pH}_{1: 2}$ values and non-saline to slightly saline $\mathrm{EC}_{1: 2}$ values of up to $0.41 \mathrm{dS} / \mathrm{m}$.

### 5.2 Acid Base (NAPP) Results

Total S, ANC and net acid production potential (NAPP) results are presented in Table 1 and 2.

Total S is low for overburden and floor samples, ranging from below detection to $0.29 \% \mathrm{~S}$, and with a median of $0.03 \% \mathrm{~S}$. ANC ranges from 0 to $94 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, but is mainly ( $85 \%$ ) less than $20 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, with a median of $5 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$.

The NAPP value is an acid-base account calculation using measured total S and ANC values. It represents the balance between the maximum potential acidity (MPA) calculated from the total S and ANC. A negative NAPP value indicates that the sample may have sufficient ANC to prevent acid generation. Conversely, a positive NAPP value indicates that the material may be acid generating.

Figure 1 is an acid base accounting plot showing total $S$ versus ANC, with NAPP positive and NAPP negative domains indicated. The plot highlights the low $\mathrm{S}(<0.1 \%)$ for most samples and high ANC/MPA ratios for many samples. There is a small population of NAPP positive samples, accounting for approximately $20 \%$ of all samples tested.

### 5.3 Single Addition NAG Results and Sample Classification

Single addition NAG test results are presented in Tables 1 and 2. A NAGpH $<4.5$ indicates the sample is acid producing. NAG test results are used in conjunction with NAPP values to classify samples according to acid forming potential.

NAGpH values for overburden and floor range from 2.2 to 8.8 , with most samples ( $>80 \%$ ) having a NAGpH greater than 4.5 .

Figure 2 is an ARD classification plot showing NAGpH versus the NAPP value. Potentially acid forming (PAF), non acid forming (NAF) and uncertain (UC) classification domains are indicated. A sample is classified PAF when it has a positive NAPP and NAGpH $<4.5$, and NAF when it has a negative NAPP and NAGpH $\geq 4.5$. Samples are classified uncertain when there is an apparent conflict between the NAPP and NAG results, i.e. when the NAPP is positive and NAGpH $\geq 4.5$, or when the NAPP is negative and $\mathrm{NAGpH}<4.5$.

Figure 2 shows that most samples plot clearly within either the NAF or PAF domains, but there is a population of NAPP negative samples that plot in the bottom left hand uncertain domain. These samples have low total S of less than $0.3 \% \mathrm{~S}$ and are generally described as being carbonaceous or including carbonaceous materials. Carbonaceous materials can interfere with the standard NAG test by releasing organic acids causing anomalously low NAGpH values. These samples are therefore expected to be NAF as indicated by the NAPP results, and have been classified UC(NAF).

For a number of the PAF samples, the NAG to pH 4.5 values exceed the maximum potential acidity (MPA), again indicating overestimation of acid capacity in the NAG test due to carbonaceous materials. NAG test results which are influenced by organic acid effects are highlighted yellow in Tables 1 and 2.

The NAGorg method was carried out on 8 selected overburden samples in which organic acid effects were indicated. Results are shown in Table 3, which show that in all cases the NAGpH value after the extended boiling step is greater than 4.5 compared to acidic NAGpH values in the standard test, which confirms the effects of organic acids. The calculated NAG value is based on the concentration of dissolved S assumed to be derived from pyrite, and the concentration of cations (particularly Ca and Mg ) indicative of acid neutralising reactions. The calculated NAG values of samples 30392 and 30412 are negative, indicating that all acid generated in the NAG test is organic, and that these samples are unlikely to generate acid. The remaining samples have low calculated NAG values of $3-6 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, indicating these samples are PAF-low capacity (PAF-LC), consistent with the NAPP values.

Based on results and discussions above, the geochemical classifications for samples are provided in Tables 1 and 2. Classification into NAF, UC(NAF), PAF-LC and PAF was carried out on the following basis:

- Samples with a NAPP $\leq 0$ and NAGpH $\geq 4.5$ were classified NAF.
- Samples with total S $\leq 0.05 \%$ were classified NAF regardless of NAPP or NAG results due to the very low risk of acid formation from these samples.
- Samples with a positive NAPP and NAGpH $<4.5$ were classified PAF-LC, since all NAPP positive values were less than $10 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$.
- Uncertain samples with a NAPP $>0$ and $\mathrm{NAGpH} \geq 4.5$ were classified UC(NAF), since the NAGpH values indicate the samples are unlikely to generate significant acid.
- Uncertain samples with a NAPP $\leq 0$ and NAGpH $<4.5$ were classified UC(NAF), since the NAGpH values were affected by organic acids and in this case the NAPP values are expected to be a better indication of acid potential.

Overall results suggest that the majority of overburden from the open cut and underground development will be NAF, with a small proportion of PAF-LC. Drillholes sampled for the open cut development were selected to maximise the mine stratigraphy tested, and results are not necessarily representative of the overall distribution of ARD types in the overburden. However, as a guide, the weighted distribution (results weighted according to sample interval length) of NAF in the overburden tested for the open cut is greater than $90 \%$. Only one sample in the underground drillhole was PAF-LC, which again equates to over $90 \%$ NAF overburden.

Figure 3 is a box plot showing the distribution of MPA and ANC for overburden samples from open cut and underground operations. The plot shows that although the median ANC values are low (less than $10 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ), they significantly exceed the median MPA values. Given the small proportion and low ARD potential of the PAF-LC materials and the higher background ANC, normal run-of-mine operational blending of overburden materials during mining should be sufficient to prevent ARD, without the need for any further materials management. Leach column testing of blended materials and ongoing testing and monitoring of overburden materials during operations would be required to confirm the validity of this approach.

None of the floor materials tested from the open cut were classified PAF-LC, which suggests that final pit floors will not be a source of ARD. Note that no specific testing of roof and floor materials for the underground development was carried out, but it is expected that these materials may be PAF-LC, consistent with Ulan Seam coal samples tested in hole TB103 (see Section 6).

### 5.4 Acid Buffering Characteristic Curve (ABCC) Testing

An acid buffering characteristic curve ( ABCC ) is produced by slow titration of a sample with acid, and provides an indication of the relative reactivity of the ANC measured. The acid buffering of a sample to pH 4 can be used as an estimate of the proportion of readily available ANC. ABCC tests were carried out on 8 selected samples from overburden and floor to evaluate the availability of the ANC measured. Results are presented in Figures 4 to 8 , with calcite, dolomite, ferroan dolomite and siderite standard curves as reference. Calcite and dolomite readily dissolve in acid and exhibit strongly buffered pH curves in the ABCC test, rapidly dropping once the ANC value is reached. The siderite standard provides very poor acid buffering, exhibiting a very steep pH curve in the ABCC test. Ferroan dolomite is between siderite and dolomite in acid buffering availability.

Figure 4 shows the ABCC profile for sandstone sample 28523, which has a high ANC value of $94 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. The plot shows strong buffering similar to dolomite, and indicates most ( $>90 \%$ ) of the total ANC measured is readily available for acid buffering.

Figure 5 shows the ABCC profile for sandstone sample 30418, which has a high ANC value of $82 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. The sample profile plots between the dolomite and ferroan dolomite standard curves, indicating the ANC is due to ferroan dolomite, and will be less reactive than dolomite. The curve indicates that approximately $75 \%$ of the total ANC will be readily available for acid buffering.

Figures 6 to 8 show ABCC profiles for a variety of lithologies with varying ANC. All curves plot between the ferroan dolomite and siderite trends, indicating poor reactivity. Results suggest that a large proportion of the total ANC will be ineffective, with readily available proportions ranging from $15 \%$ to $60 \%$ of the total ANC.

Results indicate that the measured ANC in the overburden at Moolarben may not be fully effective, except for the higher ANC ( $>80 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ) sandstone. Reliance on acid buffering from these materials is likely to require long residence times and low flushing rates.

### 5.5 Elemental Enrichment and Solubilities

Results of multi-element scans for 17 selected overburden and floor sample solids were compared to the median soil abundance (from Bowen, 1979 ${ }^{1}$ ) to highlight enriched elements.

The extent of enrichment is reported as the Geochemical Abundance Index (GAI), which relates the actual concentration with an average abundance on a $\log 2$ scale. The GAI is expressed in integer increments where a GAI of 0 indicates the element is present at a concentration similar to, or less than, average abundance; and a GAI of 6 indicates approximately a 100 -fold enrichment above average abundance. As a general rule, a GAI of 3 or greater signifies enrichment that warrants further examination.

Results of multi-element analysis are presented in Table 4, and the corresponding GAI values are presented in Table 5. Results show enrichment of Be in most samples but with a maximum of $6.2 \mathrm{mg} / \mathrm{kg}$, which is within normal ranges for soils. No other elements were significantly enriched compared to normal soils.

The same 17 overburden sample solids were subjected to water extraction at a solids:liquor ratio of $1: 2$. The results are shown in Table 6 , and provide a guide to those elements readily liberated when flushed. The pH of the extracts was greater than 6.5 , and little mobilisation of metals and metaloids was expected. The extracts show elevated concentrations of $\mathrm{Al}(>1 \mathrm{mg} / \mathrm{L})$, but given the circum neutral pH , the elevated Al is likely to be due to the presence of colloidal or very fine silicate minerals in the solution after

[^0]filtering the extracts. Slightly elevated $\mathrm{Co}(0.3 \mathrm{mg} / \mathrm{L})$ and $\mathrm{Ni}(0.4 \mathrm{mg} / \mathrm{L})$ were measured in sample 28495. No other significant mobilisation of elements of environmental concern was evident.

### 5.6 Sodicity and Dispersion

Soluble and exchangeable cations and Emerson aggregate tests (EAT) were carried out on selected overburden samples to provide a preliminary indication of any sodicity and dispersion issues. Results are presented in Table 7.

Sodic materials tend to form low permeability soil horizons, accelerating erosion and inhibiting plant growth. Sodic soils are also dispersive and should not be used as construction materials since they are prone to tunnelling and collapse. The exchangeable sodium percentage (ESP) is a measure of exchangeable Na as a percentage of the total effective cation exchange capacity (ECEC). The ESP can be used to classify samples according to sodicity as follows:

ESP $<6 \%$ - Non-Sodic
ESP 6-15\% - Sodic
ESP $15-30 \%$ - Strongly Sodic
ESP >30\% - Very Strongly Sodic
The ESP for the single topsoil sample was sodic (ESP 6\%-15\%). ESP's for Quaternary/Tertiary Alluvial samples ranged from sodic (ESP 6\%-15\%) to strongly sodic (15-30\%). ESP's for weathered Permian samples were all sodic (ESP 6\%-15\%). ESP's for unweathered Permian samples were all non-sodic (ESP $<6 \%$ ). Sodic materials may be subject to surface crusting and high erosion rates if placed in the surface of dumps and exposed directly to rainfall.

The dispersive properties of materials can also be measured more directly using the Emerson aggregate test (EAT). This test assigns classes to samples according to dispersive behaviour of sample aggregates in water. The samples are divided into 8 main classes, and up to 4 sub classes. In general, samples classified as Class 1 or 2 indicate dispersion and associated risk of tunnelling, surface crusting and water erosion.

Emerson aggregate test (EAT) classes generally indicated low dispersion risk (Class 5), apart from topsoil sample 30386, Quaternary/Tertiary Alluvial sample 30380 and weathered Permian sample 30377. A moderate dispersion risk (Class 2) was indicated for the weathered Permian sample 30377. Topsoil sample 30386 and Quaternary/Tertiary Alluvial sample 30380 had ESP's of class 3, and are not likely to be dispersive unless worked extensively when wet.

Results indicate a possible sodicity hazard for topsoil, Quaternary/Tertiary alluvials and weathered Permian. Materials with sodic/dispersion potential may require treatment (with gypsum or lime) if exposed on dump surfaces or used in engineered structures.

### 6.0 Coal and Reject Results

### 6.1 Geochemical Characterisation

Six Ulan Coal Seam samples were tested from underground drillhole TB103, representing the full seam interval from 114 m to 128 m depth. Two reject samples from coal washing trials of coal samples from WMLB24 were also tested, representing a high ash domestic thermal product (Ulan seams A to C) and an export thermal product (Ulan seams D to E). Results of geochemical characterisation for coal samples and reject samples are shown in Tables 8 and 9 , respectively.

The coal samples generally had slightly acidic $\mathrm{pH}_{1: 2}$ values of down to pH 4.5 , and moderately saline to saline $\mathrm{EC}_{1: 2}$ values of up to $2 \mathrm{dS} / \mathrm{m}$. The samples also have elevated total sulphur of up to $0.66 \% \mathrm{~S}$, and it is likely that the lower pH and elevated EC are due to accumulation of pyrite oxidation products between sampling and testing.

The high ash domestic coal reject sample (EGi sample No 30372) had a pH of 5.9 and a non saline $E C$ of $0.31 \mathrm{dS} / \mathrm{m}$, indicating a lack of existing acidity or salinity in this sample. The export coal reject sample has a lower pH of 4.2, indicating some minor existing acidity is this sample, but a non saline EC of $0.31 \mathrm{dS} / \mathrm{m}$.

Total S ranged from $0.14 \%$ to $0.66 \%$ S in the coal, and $0.27 \%$ to $1.06 \% \mathrm{~S}$ in the rejects. The export thermal coal reject sample had the highest S. ANC was low for coal and rejects, ranging from 2 to $11 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ for the coal, and 0 to $1 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ for the rejects. Sulphur ranges for the coal were consistent with Moolarben total S test results of washed coal samples from the open cut deposit (reproduced in Table 10), and raw coal from underground deposit (reproduced in Table 11).

Figure 9 is an acid base account plot for coal and reject samples. The figure shows that all samples plot in the NAPP positive domain, apart from one coal sample (30425). Figure 10 is an ARD classification plot for coal and reject samples. The plot shows that most samples plot in the PAF domain, one sample in the NAF domain, and one in the upper right hand uncertain domain. The uncertain coal sample has a very low NAPP value of $2 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, and a low S value of $0.30 \% \mathrm{~S}$, which if present as pyrite would be expected to completely react in a single addition NAG test. The NAG results suggest that this borderline NAPP positive sample is NAF.

NAG test results to pH 4.5 for 4 of the coal samples and the high ash coal reject sample 30372 are greater than the MPA, indicating organic acid effects. The NAGorg method was carried out on 2 coal samples and both reject samples to determine the proportion of acid measured in the standard NAG test that can be attributed to pyrite oxidation. Results are shown in Tables 8 and 9.

The calculated NAG values for the coal samples are significantly lower than the NAPP values, suggesting that most of the S measured in the coal samples is in non-acid
generating forms (possibly organic S). The calculated NAG value for one of the coal samples (30428) was $0 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, and is classified UC(NAF). The calculated NAG value for the other coal sample (30429) was $4 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, indicating the sample is PAF, but with a low capacity (PAF-LC).

The high ash coal reject sample (30372) has a calculated NAG value of $6 \mathrm{~kg} \mathrm{H} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, consistent with the NAPP value and the sample is classified PAF-LC. The other reject sample has $1 \%$ S, and the NAPP, standard NAG to pH 4.5 , and calculated NAG are all similar, confirming a PAF classification for this sample.

Kinetic NAG tests provide an indication of the kinetics of sulphide oxidation and acid generation for a sample. Kinetic NAG testing was carried out on the export thermal coal reject sample (30373) with a S value of $1 \%$ S. The kinetic NAG pH and temperature profiles are presented in Figure 11. The sample shows very rapid acid production within the first few minutes of the test, and indicates that pyritic reject materials are likely to produce acid within days to weeks after exposure. In addition, the temperature profile is typical of pyritic samples, and confirms that most of the $1 \% \mathrm{~S}$ in the sample is in pyritic form.

Figure 12 shows the ABCC profile for a coal sample with an ANC of $11 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. The sample profile plots between the ferroan dolomite and siderite trends, indicating poor reactivity as with many of the overburden samples (see Section 5.4). Results suggest that about $40 \%$ of the total ANC is readily available.

Results of testing indicate that coal may leach minor amounts of ARD. Run off and leachate from coal stockpiles and underground workings (in which coal seams will be exposed in pillars, roof and floor) may need to be contained, monitored and possibly treated.

Based on results from the two reject samples tested, rejects appear to be acid forming and fast reacting, with the export coal rejects showing the highest ARD potential. Pyritic rejects will require management to control ARD, which may involve underwater disposal, lime treatment, isolation from infiltration, or a combination of these.

### 6.2 Multi-Elements

Two coal and both reject sample solids were analysed for multi-elements. Results of multi-element analysis and the corresponding GAI values are presented in Table 12. Results show that in addition to S (which is discussed in relation to acid forming potential), there is enrichment of Be in the coal and export thermal coal reject sample, and enrichment of Ag in the high ash coal reject sample. However, overall the concentration of metals and metaloids in the solids is similar to background in soils.

The same sample solids were subjected to water extraction at a solids:liquor ratio of 1:2. The results are shown in Table 13. There was no significant mobilisation of metals/metaloids from the two coal samples, but some minor mobilisation of $\mathrm{Co}, \mathrm{Cu}$ and

Zn was indicated for the high ash coal reject extract, and minor mobilisation of $\mathrm{Al}, \mathrm{Ni}$ and Zn for the export coal reject extract. Mobilisation of these elements and other metals/metaloids will be largely controlled by pH , so that management of ARD will effectively manage metal/metaloid release.

### 7.0 Conclusions and Recommendations

Results of ARD investigations indicate that over $90 \%$ of overburden material for open cut and underground operations is likely to be non acid forming (NAF). PAF-LC materials appear to have low ARD potential and represent a small proportion of overburden to be mined. Although the ANC is poorly reactive, it significantly exceeds the background MPA, and normal run-of-mine operational blending of overburden materials during mining should be sufficient to control ARD without the need for any further materials management. Leach column testing of blended materials and ongoing testing and monitoring of overburden materials during operations would be required to confirm the validity of this approach.

No PAF materials were identified for floor samples from the open cut, which suggests that final pit floors will not be a source of ARD. Preliminary results indicate roof and floor materials for the underground project may be PAF-LC.

Most of the coal seam samples tested were PAF-LC, indicating potential acid release from coal stockpiles and underground workings. Coal reject samples were acid forming, with export coal rejects showing the highest ARD potential.

Testing indicates that overburden, floor and coal reject materials are likely to be nonsaline. Coal samples were moderately saline to saline.

Exchangeable sodium percentages and Emerson aggregate test (EAT) results indicate a possible sodicity hazard for topsoil, Quaternary/Tertiary alluvials and weathered Permian. Materials with sodic/dispersion potential may require treatment (with gypsum or lime) if exposed on dump surfaces or used in engineered structures.

No significant enrichment of metals/metaloids was detected in overburden, coal or reject solids.

The findings of these initial investigations have the following implications for materials management:

- Results suggest that normal run-of-mine operational blending of overburden should be sufficient to control ARD.
- Containment of run off and leachate from coal stockpiles and underground operations may be required to monitor water quality and determine whether treatment is required. Results indicate that these waters may be saline and acidic. The sensitivity of groundwater and surface water to saline and acidic water should be investigated to
determine the degree of management required. Provision for acid treatment may be needed, which could include use of a mobile lime dosing plant to treat acid waters and broadcast application of agricultural lime.
- Rejects appear to have a higher ARD risk than other mine materials, and are likely to require specific management to control ARD. Possible approaches include underwater disposal, lime treatment, isolation from infiltration, or a combination of these.
- Materials with sodic/dispersion potential may require treatment (with gypsum or lime) if exposed on dump surfaces or used in engineered structures.
- A routine system of ARD testing should be established during operations to check the ARD potential of mine materials and allow for modification of materials management strategies if required.

The following work is recommended to finalise materials management requirements:

- More widespread sampling and testing of overburden materials should be carried out to ensure there are no significant PAF units within overburden and floor, and confirm the distribution and ARD potential of mine materials indicated by testing to date.
- Leach column testing of PAF-LC overburden blended with NAF overburden should be carried out to confirm the validity of the operational blending approach.
- Additional samples of representative reject and coal materials should be tested to determine the variability in ARD potential.
- Leach column testing of coal and reject materials should be carried out to determine lag times and acid release rates, which in turn can be used to determine appropriate treatment rates and other management requirements.
Table 1：Geochemical characterisation results for samples from the proposed open pit．

|  |  | $\frac{4}{2} \frac{4}{2}$ | $\frac{u}{2}$ |  |  |  |  |  | 㞱 | $\frac{4}{\text { L }}$ | $\frac{u}{z}$ | $\frac{u}{2} \frac{u}{2}$ |  |  | 2 | $\left\|\begin{array}{c} 0 \\ \frac{1}{4} \\ \dot{L} \\ \vdots \end{array}\right\|$ | $\frac{4}{2}$ | $\stackrel{4}{4}$ | $\frac{1}{4}$ |  | $\frac{4}{5}$ |  | $2$ | $\geq$ | $2$ | $\frac{1}{2}$ | $\begin{aligned} & 0 \\ & \hline 1 \\ & \frac{1}{2} \\ & \frac{1}{2} \\ & \end{aligned}$ | $\frac{4}{4}$ | 殅 | 害 |  | 炭 | 殅 |  | $\frac{1}{\frac{1}{2}}$ |
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| $\begin{array}{l\|l} \frac{2}{4} & \frac{0}{4} \\ \frac{2}{4} & \frac{1}{2} \end{array}$ |  |  | $\because$ |  |  |  |  |  | $\checkmark$ | － | ָ |  |  | － |  | $\checkmark$ | $\checkmark$ |  |  |  | $\stackrel{\square}{9}$ |  |  |  |  |  |  | T | F |  |  | － | $\cdots$ |  | $\div \cdots$ |
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|  |  | $\begin{array}{ll} \overline{5} \\ \dot{y} \\ \dot{v} \\ \dot{v} \\ \hline \end{array}$ |  |  |  |  |  |  | $\begin{aligned} & \bar{\circ} \\ & \stackrel{\rightharpoonup}{v} \end{aligned}$ | $\begin{aligned} & \bar{\circ} \\ & \stackrel{\rightharpoonup}{v} \end{aligned}$ | No | $$ |  | No | $0$ | No | $\bar{o}_{0}$ | $\stackrel{O}{\circ}$ | en |  | No |  |  | $\stackrel{m}{0}$ | $\mathfrak{c}$ | $0$ | $\underset{\substack{\mathrm{N} \\ \hline \\ \underset{\sim}{2} \\ \hline}}{ }$ | $\stackrel{\rightharpoonup}{v}$ | No. |  |  | \％ | No |  | OM |
| ¢ِّ |  |  |  |  |  |  |  | Bo: | O | $\stackrel{\circ}{\circ}$ | O8 | $0$ |  | $8$ | \|o| | $\underset{0}{F}$ | O웅 | $\stackrel{\square}{\circ}$ | $\%$ |  | $\mathrm{O}_{\mathrm{O}}^{\circ}$ |  |  |  |  |  |  | $\bigcirc$ | $\stackrel{m}{\circ}$ |  |  | \％ | $\div$ |  | O80 |
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Table 1: Geochemical characterisation results for samples from the proposed open pit.

| Hole No | Open Cut Pit No | $\begin{gathered} \text { EGi } \\ \text { Code } \end{gathered}$ | Moolarben Sample No | Depth (m) |  |  | Unit | Material Type | Geological Description | $\mathrm{pH}_{1: 2}$ | $\mathrm{EC}_{1: 2}$ | ACID-BASE ANALYSIS |  |  |  |  | NAG TEST |  |  | ARD Classification |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  |  | From | To | Interval |  |  |  |  |  | $\begin{gathered} \text { Total } \\ \% \mathbf{S} \end{gathered}$ | MPA | ANC | NAPP | ANC/MPA | NAGpH | NAG ${ }_{(\text {(pr4.5) }}$ | NAG ${ }_{\text {(ph7.0) }}$ |  |
| WMLB15 | OC2 | 28526 | WMLB15-E21 | 34.25 | 34.88 | 0.63 | Permian | Overburden | Shale, tending to mudstone, carbonaceous | 6.73 | 0.10 | 0.05 | 2 | 3 | -1 | 1.96 | 5.7 | 0 | 2 | NAF |
| WMLB15 | OC2 | 28527 | WMLB15-E22 | 34.88 | 36.41 | 1.53 | Permian | Overburden | Sandstone, fine to medium, few silty phases and carbonaceous traces, tending to carbonaceous siltstone at base | 6.59 | 0.13 | 0.02 | 1 | 5 | -4 | 8.17 | 7.1 | 0 | 0 | NAF |
| WMLB15 | OC2 |  | Coal | 36.41 | 48.21 | 11.80 | Permian | Coal | Coal - Ulan Seam A-E |  |  |  |  |  |  |  |  |  |  |  |
| WMLB15 | OC2 | 28528 | WMLB15-E23 | 48.21 | 49.20 | 0.99 | Permian | Floor | Sandstone, fine, Carbonaceous silty laminae throughout, minor carbonaceous material | 7.21 | 0.09 | 0.04 | 1 | 1 | 0.2 | 0.82 | 3.0 | 7 | 25 | NAF |
| WMLB15 | OC2 | 28529 | WMLB15-E24 | 49.20 | 51.35 | 2.15 | Permian | Floor | Sandstone, medium fine, few silty laminae, tending to massive | 7.11 | 0.19 | 0.09 | 3 | 3 | 0 | 1.09 | 5.6 | 0 | 1 | NAF |
| WMLB75 | OC3 | 30374 |  | 0.00 | 2.00 | 2.00 | Quaternary/Tertiary | Overburden | Alluvium/weathered Permian colluvium (close to WMLB5) | 7.21 | 0.09 | <0.01 | 0 | 1 | -1.0 |  | 6.1 | 0 | 6 | NAF |
| WMLB75 | OC3 | 30375 |  | 2.00 | 3.00 | 1.00 | Quaternary/Tertiary | Overburden | Alluvium/weathered Permian colluvium (close to WMLB5) | 7.34 | 0.09 | <0.01 | 0 | 2 | -2.0 |  | 6.2 | 0 | 4 | NAF |
| WMLB75 | OC3 | 30376 |  | 3.00 | 4.00 | 1.00 | Permian | Overburden | Weathered Permian (close to WMLB5) | 8.02 | 0.08 | <0.01 | 0 | 1 | -1.0 |  | 6.2 | 0 |  | NAF |
| WMLB75 | OC3 | 30377 |  | 4.00 | 5.00 | 1.00 | Permian | Overburden | Weathered Permian (close to WMLB5) | 7.61 | 0.06 | $<0.01$ | 0 | 2 | -2.0 |  | 6.0 | 0 | 5 | NAF |
| WMLB75 | OC3 | 30378 |  | 5.00 | 6.00 | 1.00 | Permian | Overburden | Weathered Permian (close to WMLB5) | 8.06 | 0.08 | <0.01 | 0 | 1 | -1.0 |  | 6.1 | 0 | 4 | NAF |
| WMLB75 | OC3 | 30379 |  | 6.00 | 7.00 | 1.00 | Permian | Overburden | Weathered Permian (close to WMLB5) | 7.37 | 0.08 | 0.02 | 0 | 0 | 0.0 |  | 6.1 | 0 | 2 | NAF |
| WMLB5 | OC3 |  | Open Hole | 0.00 | 7.85 | 7.85 | Permian |  |  |  |  |  |  |  |  |  |  |  |  |  |
| WMLB5 | OC3 | 28530 | WMLB5-E1 | 7.85 | 8.02 | 0.17 | Permian | Overburden | Clay, red brown, moderately soft | 7.16 | 0.11 | 0.02 | 1 | 24 | -23 | 39.22 | 7.4 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28531 | WMLB5-E2 | 8.02 | 9.10 | 1.08 | Permian | Overburden | Siltstone, dark grey, hard | 7.23 | 0.11 | 0.14 | 4 | 10 | -6 | 2.33 | 6.5 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28532 | WMLB5-E3 | 9.10 | 9.30 | 0.20 | Permian | Overburden | Tuff, white, moderately hard | 6.67 | 0.12 | 0.03 | 1 | 9 | -8 | 9.80 | 5.4 | 0 | 9 | NAF |
| WMLB5 | OC3 | 28533 | WMLB5-E4 | 9.30 | 11.00 | 1.70 | Permian | Overburden | Siltstone, grey, hard, minor tuff | 6.55 | 0.11 | 0.10 | 3 | 37 | -34 | 12.09 | 8.0 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28534 | WMLB5-E5 | 11.00 | 12.04 | 1.04 | Permian | Overburden | Siltstone, grey, hard | 6.49 | 0.14 | 0.03 | 1 | 7 | -6 | 7.63 | 5.2 | 0 | 11 | NAF |
| WMLB5 | OC3 | 28535 | WMLB5-E6 | 12.04 | 13.37 | 1.33 | Permian | Overburden | Tuff (white sandy) at top and base with friable siltstone in middle | 7.32 | 0.15 | 0.04 | 1 | 21 | -20 | 17.16 | 7.6 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28536 | WMLB5-E7 | 13.37 | 14.03 | 0.66 | Permian | Overburden | Siltstone, grey, massive, sandy at base, hard | 6.46 | 0.11 | 0.02 | 1 | 2 | -1 | 3.27 | 3.3 | 4 | 22 | NAF |
| WMLB5 | OC3 | 28537 | WMLB5-E8 | 14.03 | 14.27 | 0.24 | Permian | Overburden | Sandstone, light brown, medium grained, carbonaceous wisps throughout, hard | 6.55 | 0.13 | 0.03 | 1 | 10 | -9 | 10.89 | 7.3 | 0 | 0 | NAF |
| WMLB5 | OC3 |  | Coal | 14.27 | 15.47 | 1.20 | Permian | Coal | Coal - Ulan Seam B |  |  |  |  |  |  |  |  |  |  |  |
| WMLB5 | OC3 | 28538 | WMLB5-E9 | 15.47 | 16.03 | 0.56 | Permian | Overburden | Siltstone, carbonaceous, tending to very fine sandstone | 7.14 | 0.11 | 0.07 | 2 | 4 | -2 | 1.87 | 3.9 | 2 | 17 | UC(NAF) |
| WMLB5 | OC3 | 28539 | WMLB5-E10 | 16.03 | 16.62 | 0.59 | Permian | Overburden | Sandstone, fine to coarse, silty laminae, carbonaceous traces and granular conglomerate | 7.12 | 0.17 | 0.03 | 1 | 3 | -2 | 3.27 | 6.9 | 0 | 0 | NAF |
| WMLB5 | OC3 |  | Coal | 16.62 | 16.92 | 0.30 | Permian | Coal | Coal - Ulan Seam C |  |  |  |  |  |  |  |  |  |  |  |
| WMLB5 | OC3 | 28540 | WMLB5-E11 | 16.92 | 17.10 | 0.18 | Permian | Overburden | Sandstone, fine, carbonaceous | 7.16 | 0.16 | 0.07 | 2 | 1 | 1 | 0.47 | 3.0 |  | 24 | PAF-LC |
| WMLB5 | OC3 | 28541 | WMLB5-E12 | 17.10 | 18.01 | 0.91 | Permian | Overburden | Sandstone, fine to very coarse, siltstone, laminated | 7.21 | 0.14 | 0.04 | 1 | 2 | -1 | 1.63 | 6.6 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28542 | WMLB5-E13 | 18.01 | 19.80 | 1.79 | Permian | Overburden | Intrusion, altered soft greasy chloritic in part, greyish green | 7.16 | 0.11 | 0.12 | 4 | 34 | -30 | 9.26 | 8.2 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28543 | WMLB5-E14 | 19.80 | 21.37 | 1.57 | Permian | Overburden | Intrusion, altered soft greasy chloritic in part, greyish green | 7.29 | 0.10 | 0.13 | 4 | 33 | -29 | 8.30 | 8.4 | 0 | 0 | NAF |
| WMLB5 | OC3 | 28544 | WMLB5-E15 | 21.37 | 22.00 | 0.63 | Permian | Overburden | Sandstone, very coarse, tending to conglomerate, few coaly traces | 7.21 | 0.08 | 0.04 | 1 | 1 | 0.2 | 0.82 | 4.6 | 0 | 7 | NAF |
| WMLB5 | OC3 | 28545 | WMLB5-E16 | 22.00 | 22.61 | 0.61 | Permian | Overburden | Shale and sandstone, partly carbonaceous, banded with claystone and stoney coal | 7.35 | 0.09 | 0.11 | 3 | 2 | 1 | 0.59 | 2.8 | 21 | 44 | PAF-LC |
| WMLB5 | OC3 | 28546 | WMLB5-E17 | 22.61 | 24.62 | 2.01 | Permian | Overburden | Sandstone, fine to coarse tending to granular conglomerate, some silty laminae | 7.39 | 0.11 | 0.03 | 1 | 1 | 0 | 1.09 | 5.3 | 0 | 5 | NAF |
| WMLB5 | OC3 | 28547 | WMLB5-E18 | 24.62 | 25.91 | 1.29 | Permian | Overburden | Sandstone (medium-coarse) and carbonaceous shale | 7.46 | 0.12 | 0.08 | 2 | 2 | 0.4 | 0.82 | 3.6 | 2 | 13 | PAF-LC |
| WMLB5 | OC3 |  | Coal | 25.91 | 29.19 | 3.28 | Permian | Coal | Coal - Ulan Seam D - EBT |  |  |  |  |  |  |  |  |  |  |  |
| WMLB5 | OC3 | 28548 | WMLB5-E19 | 29.19 | 30.79 | 1.60 | Permian | Floor | Sandstone (fine) and carbonaceous shale and siderite (inlcudes Ulan Seam ELW) | 7.41 | 0.11 | 0.14 | 4 | 4 | 0.3 | 0.93 | 4.8 | 0 | 11 | UC(NAF) |
| WMLB5 | OC3 | 28549 | WMLB5-E20 | 30.79 | 32.35 | 1.56 | Permian | Floor | Sandstone, fine to medium, few carbonaceous traces | 7.11 | 0.10 | 0.06 | 2 | 2 | 0 | 1.09 | 6.9 | 0 | 0 | NAF |
| KEY |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{pH}_{1: 2}=\mathrm{pH}$ of 1:2 extract |  |  |  |  |  |  |  |  | NAGpH $=\mathrm{pH}$ of NAG liquor |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{EC}_{1: 2}=$ Electrical Conductivity of 1:2 extract (dS/m) |  |  |  |  |  |  |  |  | $\mathrm{NAG}_{(\text {(p44,5) }}=$ Net Acid Generation capacity to $\mathrm{pH} 4.5\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |
| MPA $=$ Maximum Potential Acidity $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  | $\mathrm{NAG}_{(\text {prl7.0) }}=$ Net Acid Generation capacity to $\mathrm{pH} 7.0\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |
| ANC $=$ Acid Neutralising Capacity ( $\left.\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  | $\square=$ NAG Result Affected by High Organic Carbon in Sample |  |  |  |  |  |  |  |  |  |  |  |
| NAPP $=$ Net Acid Producing Potential $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |

Table 2: Geochemical characterisation results for samples from hole TB103, drilled close to proposed ventilation shaft and access portals for the underground mine.

Table 3: Modified NAG test results for selected overburden samples.

| EGi Code | Hole No | Material Type | Geological Description | Acid-Base Analysis |  |  |  | Standard NAG Test |  |  | NAGorg TEST |  | ARD Classification |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  |  | $\begin{array}{\|c\|} \hline \text { Total } \\ \text { \%S } \\ \hline \end{array}$ | MPA | ANC | NAPP | NAGpH | NAG ${ }_{(\text {(PH44 }}{ }^{\text {a }}$ | NAG ${ }_{(\text {(pH7.0) }}$ | Extended Boil NAGpH | $\begin{gathered} \text { Calculated } \\ \text { NAG } \\ \hline \end{gathered}$ |  |
| 28490 | WMLB24 | Overburden | Coal, shaley, banded with mudstone, dark brown | 0.19 | 6 | 3 | 3 | 2.3 | 76 | 122 | 6.8 | 6 | PAF-LC |
| 28493 | WMLB24 | Overburden | Mudstone, coaly and carbonaceous | 0.14 | 4 | 2 | 2 | 2.2 | 45 | 79 | 6.7 | 4 | PAF-LC |
| 28503 | WMLB24 | Overburden | Mudstone, carbonaceous, few penny bands of $b$ coal | 0.20 | 6 | 2 | 4 | 2.4 | 43 | 73 | 6.8 | 5 | PAF-LC |
| 28508 | WMLB15 | Overburden | Coal (Moolarben), Dmb, stoney, minor tuff and sideritic in part | 0.21 | 6 | 5 | 1 | 2.4 | 68 | 109 | 6.8 | 4 | PAF-LC |
| 28514 | WMLB15 | Overburden | Mudstone, carbonaceous | 0.20 | 6 | 2 | 4 | 2.4 | 37 | 68 | 5.4 | 3 | PAF-LC |
| 28545 | WMLB5 | Overburden | Shale and sandstone, partly carbonaceous, banded with claystone and stoney coal | 0.11 | 3 | 2 | 1 | 2.8 | 21 | 44 | 6.5 | 3 | PAF-LC |
| 30392 | TB103 | Overburden | Silty clay, brown with black shale | 0.13 | 4 | 1 | 3 | 2.6 | 18 | 45 | 7.5 | -1 | NAF |
| 30412 | TB103 | Overburden | Carbonaceous shale, black | 0.23 | 7 | 9 | -2 | 3.3 | 7 | 22 | 7.7 | -4 | NAF |

[^1]ANC $=$ Acid Neutralising Capacity $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$
NAPP $=$ Net Acid Producing Potential $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$
NAGpH $=\mathrm{pH}$ of NAG liquor
$\mathrm{NAG}_{(\mathrm{pH} 4.5)}=$ Net Acid Generation capacity to $\mathrm{pH} 4.5\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$
$\mathrm{NAG}_{(\mathrm{pH7} .0)}=$ Net Acid Generation capacity to pH $7.0\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$
Extended Boil NAGpH $=\mathrm{pH}$ of NAG liquor after extended heating
Calculated NAG = The net acid potential based on assay of anions and cations released to the NAG solution $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$
Table 4: Multi-element composition of selected overburden and floor solid samples ( $\mathrm{mg} / \mathrm{kg}$ except where shown).

| Element | Detection Limit | EGi Sample Number and Sample Description |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | 28491 | 28494 | 28495 | 28497 | 28503 | 28508 | 28513 | 28514 | 28542 | 30377 | 30386 | 30390 | 30403 | 30412 | 30418 | 30422 | 30431 |
|  |  | Mudstone | Siltstone | Sandstone | Sandstone | Mudstone | Coal (Moolarben Seam) | Tuff | Mudstone | Intrusion | Weathered Permian | Top Soil | Sandy Clay | Shale | Shale | Sandstone | Shale | Shale |
| Ag | 0.1 | $<0.1$ | 0.3 | 0.1 | 0.2 | 0.2 | 0.1 | 0.2 | 0.1 | 0.1 | <0.1 | $<0.1$ | <0.1 | $<0.1$ | $<0.1$ | <0.1 | <0.1 | <0.1 |
| Al | 0.002\% | 5.59\% | 7.92\% | 3.45\% | 10.04\% | 7.23\% | 3.47\% | 12.77\% | 4.01\% | 4.59\% | 3.70\% | 1.99\% | 1.80\% | 7.64\% | 5.39\% | 5.15\% | 7.91\% | 7.87\% |
| As | 1 | 1 | 2 | 3 | 20 | 8 | 3 | 6 | 1 | 3 | 5 | 1 | 2 | 6 | 3 | 6 | 8 | 3 |
| Ba | 0.1 | 217.9 | 295 | 197.8 | 236.6 | 159 | 137.5 | 75.4 | 121.6 | 354 | 175.5 | 91.9 | 91.4 | 353.5 | 192.8 | 217.2 | 332 | 265.8 |
| Be | 0.1 | 3.8 | 3 | 0.6 | 2.7 | 4.2 | 4.3 | 2.6 | 5.2 | 3.2 | 2.5 | 0.5 | 0.4 | 3.7 | 3.5 | 1.4 | 3.2 | 3 |
| Ca | 0.001\% | 0.05\% | 0.05\% | 0.02\% | 0.06\% | 0.05\% | 0.09\% | 0.11\% | 0.03\% | 0.16\% | 0.06\% | 0.03\% | 0.03\% | 0.33\% | 0.15\% | 1.69\% | 0.21\% | 0.11\% |
| Cd | 0.1 | 0.1 | 0.2 | $<0.1$ | 0.2 | 0.1 | $<0.1$ | 0.4 | 0.1 | 0.2 | $<0.1$ | <0.1 | <0.1 | $<0.1$ | 0.1 | $<0.1$ | 0.2 | 0.2 |
| Co | 0.1 | 1.9 | 5.5 | 1.6 | 37.5 | 5.1 | 4.3 | 5.4 | 8.5 | 6.4 | 11.2 | 3.9 | 1.6 | 6.8 | 7 | 9.9 | 11.3 | 8.4 |
| Cr | 2.0 | 24 | 62 | 17 | 88 | 110 | 6 | 5 | 17 | 22 | 80 | 68 | 30 | 28 | 12 | 69 | 96 | 44 |
| Cu | 1.0 | 17 | 37 | 3 | 32 | 39 | 6 | 4 | 16 | 13 | 12 | 7 | 3 | 26 | 7 | 7 | 34 | 16 |
| F | 50 | 465 | 494 | 158 | 392 | 491 | 394 | 1073 | 356 | 962 | 229 | 152 | 118 | 447 | 651 | 260 | 295 | 416 |
| Fe | 0.01\% | 0.40\% | 0.38\% | 0.41\% | 1.29\% | 0.38\% | 0.62\% | 0.16\% | 0.10\% | 1.11\% | 1.43\% | 0.57\% | 0.38\% | 2.20\% | 1.22\% | 1.99\% | 3.42\% | 1.50\% |
| Hg | 0.01 | <0.05 | 0.07 | $<0.05$ | 0.06 | 0.08 | 0.06 | 0.16 | $<0.05$ | $<0.05$ | $<0.05$ | $<0.05$ | $<0.05$ | 0.19 | <0.05 | <0.05 | <0.05 | <0.05 |
| K | 0.002\% | 0.84\% | 1.58\% | 1.03\% | 1.27\% | 1.13\% | 0.36\% | 0.89\% | 0.29\% | 1.20\% | 0.26\% | 0.24\% | 0.18\% | 1.10\% | 0.95\% | 1.34\% | 2.52\% | 1.99\% |
| Mg | 0.002\% | 0.15\% | 0.21\% | 0.06\% | 0.20\% | 0.19\% | 0.15\% | 0.28\% | 0.06\% | 0.33\% | 0.13\% | 0.06\% | 0.04\% | 0.34\% | 0.25\% | 0.82\% | 0.45\% | 0.24\% |
| Mn | 1.0 | 43 | 31 | 33 | 223 | 37 | 134 | 4 | 8 | 198 | 47 | 81 | 25 | 510 | 150 | 436 | 462 | 221 |
| Mo | 0.1 | 2.7 | 0.4 | 0.2 | 2.6 | 0.5 | 2.6 | 1.7 | 1.6 | 2.6 | 0.6 | 0.4 | 0.3 | 2.4 | 4.4 | 0.5 | 0.9 | 1.2 |
| Na | 0.002\% | 0.03\% | 0.04\% | 0.03\% | 0.03\% | 0.03\% | 0.02\% | 0.02\% | 0.02\% | 0.05\% | 0.05\% | 0.03\% | 0.03\% | 0.04\% | 0.03\% | 0.03\% | 0.04\% | 0.04\% |
| Ni | 1.0 | 9 | 20 | 5 | 99 | 55 | 24 | 26 | 31 | 26 | 35 | 15 | 11 | 48 | 19 | 53 | 68 | 20 |
| P | 20 | 89 | 70 | 88 | 284 | 111 | 63 | 236 | 80 | 85 | 229 | 95 | 74 | 101 | 47 | 124 | 232 | 142 |
| Pb | 2.0 | 17 | 24 | 11 | 27 | 20 | 15 | 56 | 16 | 21 | 12 | 8 | 5 | 31 | 22 | 13 | 18 | 23 |
| S | 0.001\% | 0.12\% | 0.06\% | 0.01\% | 0.03\% | 0.21\% | 0.27\% | 0.06\% | 0.23\% | 0.14\% | 0.01\% | 0.01\% | 0.01\% | 0.06\% | 0.16\% | 0.01\% | 0.07\% | 0.06\% |
| Sb | 0.05 | 1.59 | 0.85 | 0.31 | 0.89 | 1.31 | 1.56 | 0.97 | 2.17 | 1.18 | 0.78 | 0.49 | 0.41 | 1.33 | 1.11 | 0.48 | 0.85 | 0.52 |
| Se | 0.01 | 0.13 | 0.09 | 0.02 | 0.15 | 0.33 | 0.21 | 0.09 | 0.25 | 0.39 | 0.04 | 0.04 | $<0.01$ | 0.34 | 0.15 | 0.03 | 0.26 | 0.12 |
| Si | 0.1\% | 25.6\% | 29.2\% | 38.9\% | 29.6\% | 22.5\% | 18.9\% | 26.6\% | 19.1\% | 28.6\% | 37.6\% | 40.4\% | 40.9\% | 29.4\% | 23.1\% | 32.1\% | 26.5\% | 29.8\% |
| Sn | 0.1 | 3.8 | 3.8 | 1.2 | 4.8 | 2.9 | 2.2 | 8.8 | 2.1 | 3.6 | 1.7 | 0.8 | 0.8 | 3.4 | 3.8 | 2.3 | 3 | 4.4 |
| Sr | 0.05 | 12.59 | 25.33 | 29.4 | 157.33 | 33.55 | 20.07 | 118.99 | 25.89 | 38.08 | 42.66 | 27.76 | 22.98 | 33.44 | 19.76 | 109.7 | 115.16 | 57.85 |
| Th | 0.01 | 10.54 | 10.95 | 3.58 | 18.97 | 10.1 | 9.44 | 35.09 | 8.87 | 12.27 | 7.48 | 4.51 | 3.8 | 13.75 | 14.25 | 6.96 | 10.61 | 14.19 |
| U | 0.01 | 4.32 | 2.93 | 0.99 | 4.68 | 2.74 | 4.79 | 11.8 | 3.43 | 4.52 | 2.21 | 1.11 | 1.05 | 5.11 | 3.76 | 1.82 | 2.76 | 3.99 |
| Zn | 1.0 | 45 | 99 | 3 | 104 | 34 | 74 | 165 | 51 | 62 | 31 | 9 | 6 | 66 | 48 | 47 | 80 | 104 |

[^2]Table 5: Geochemical abundance indices (GAI) of selected overburden and floor solid samples.

Table 6: Chemical composition of water extracts of overburden and floor samples

| Parameter |  | Detection Limit | EGi Sample Number and Sample Description |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | 28491 | 28494 | 28495 | 28497 | 28503 | 28508 | 28513 | 28514 | 28542 | 30377 | 30386 | 30390 | 30403 | 30412 | 30418 | 30422 | 30431 |
|  |  | Mudstone | Siltstone | Sandstone | Sandstone | Mudstone | Coal <br> (Moolarben Seam) | Tuff | Mudstone | Intrusion | Weathered Permian | Top Soil | Sandy Clay | Shale | Shale | Sandstone | Shale | Shale |
| pH |  |  | 0.01 | 6.83 | 6.53 | 7.21 | 6.63 | 7.51 | 7.88 | 7.89 | 7.67 | 7.36 | 7.67 | 7.2 | 8.3 | 7.3 | 7.46 | 7.61 | 7.11 | 7.26 |
| EC | dS/m |  | 0.01 | 0.088 | 0.049 | 0.055 | 0.091 | 0.119 | 0.136 | 0.121 | 0.139 | 0.116 | 0.081 | 0.096 | 0.076 | 0.089 | 0.122 | 0.111 | 0.125 | 0.416 |
| Ag | mg/l | 0.00001 | < | < | < | < | < | < | < | < | 0.00001 | < | < | < | < | < | < | < | < |
| Al | mg/l | 0.01 | 1.4 | 1.55 | 1.35 | 0.64 | 0.94 | 0.99 | 2.49 | 0.3 | 0.2 | 3.05 | 3.05 | 1.87 | 0.14 | 3.33 | 0.84 | 2.09 | 0.32 |
| As | mg/l | 0.0001 | 0.0015 | 0.0041 | 0.0021 | 0.0126 | 0.0044 | 0.0013 | 0.0027 | 0.0025 | 0.0009 | 0.0021 | 0.001 | 0.0021 | 0.0027 | 0.0041 | 0.0058 | 0.0102 | 0.0032 |
| B | mg/l | 0.01 | 0.01 | 0.01 | < | 0.01 | 0.01 | 0.03 | 0.01 | 0.02 | < | < | 0.01 | < | 0.01 | 0.01 | 0.01 | 0.02 | 0.01 |
| Ba | mg/l | 0.00005 | 0.02162 | 0.02915 | 0.0431 | 0.01513 | 0.01326 | 0.01901 | 0.00782 | 0.03729 | 0.03581 | 0.00619 | 0.02742 | 0.02786 | 0.14396 | 0.13988 | 0.09798 | 0.17209 | 0.04227 |
| Be | mg/l | 0.0001 | 0.0001 | 0.0002 | 0.0001 | 0.0001 | 0.0003 | 0.0001 | 0.0002 | < | < | 0.0002 | 0.0001 | < | < | 0.0002 | < | 0.0001 | < |
| Ca | mg/l | 0.01 | 3.29 | 5.13 | 4.09 | 9.12 | 2.57 | 3.71 | 9.35 | 5.11 | 75.25 | 1.04 | 3.07 | 2.55 | 14.29 | 9.68 | 15.21 | 12.47 | 4.64 |
| Cd | mg/l | 0.00002 | 0.00003 | 0.00007 | < | 0.00019 | 0.00004 | < | 0.00004 | 0.00003 | 0.00002 | < | 0.00002 | 0.00003 | 0.00002 | < | < | < | < |
| Cl | mg/l | 2 | 29 | 29 | 32 | 29 | 28 | 35 | 30 | 32 | 60 | 37 | 55 | 34 | 34 | 32 | 31 | 29 | 29 |
| Co | mg/l | 0.0001 | 0.0016 | 0.0063 | 0.0011 | 0.2701 | 0.006 | 0.0026 | 0.0031 | 0.0192 | 0.0116 | 0.0007 | 0.0034 | 0.002 | 0.0031 | 0.0007 | 0.0008 | 0.001 | 0.0006 |
| Cr | mg/l | 0.01 | < | < | < | < | < | < | < | < | < | < | < | < | < |  | < | < | < |
| Cu | mg/l | 0.01 | < | < | < | < | < | < | < | < | < | 0.02 | < | < | < | < | < | < | < |
| F | mg/l | 0.1 | 0.7 | 0.9 | 0.9 | 0.5 | 0.5 | 0.9 | 1.0 | 1.0 | 1.2 | 1.4 | 0.6 | 1.4 | 0.8 | 1.6 | 1.8 | 2.0 | 1.8 |
| Fe | mg/l | 0.01 | 0.2 | 0.1 | 0.2 | 0.23 | 0.09 | 0.18 | 0.07 | 0.03 | 0.24 | 1.06 | 1.05 | 0.67 | 0.04 | 0.22 | 0.29 | 0.18 | 0.04 |
| Hg | mg/l | 0.0001 | < | < | < | < | < | < | < | < | < | $<$ | < | < | < | < | < | < | < |
| K | mg/l | 0.1 | 11.2 | 15.6 | 10.8 | 18.6 | 9.3 | 7.7 | 11.6 | 6.8 | 33.4 | 3.1 | 4.2 | 3.8 | 13.3 | 12.5 | 13.4 | 18.7 | 13.3 |
| Mg | mg/l | 0.01 | 3.57 | 5.28 | 4.5 | 10.15 | 3.08 | 6.81 | 8.47 | 6.56 | 47.93 | 4.9 | 4.12 | 1.68 | 8.57 | 4.07 | 6.89 | 5.15 | 2.23 |
| Mn | mg/l | 0.01 | 0.03 | < | 0.02 | 0.18 | 0.08 | 0.03 | < | 0.02 | 0.07 | < | 0.26 | 0.02 | 0.06 | < | 0.02 | < | < |
| Mo | mg/l | 0.00005 | 0.01539 | 0.00359 | 0.00034 | 0.06118 | 0.00211 | 0.01238 | 0.05514 | 0.0214 | 0.00829 | 0.00062 | 0.00024 | 0.00451 | 0.013 | 0.02694 | 0.01047 | 0.02214 | 0.01608 |
| Na | mg/l | 0.1 | 22.2 | 21.4 | 21.9 | 23.8 | 22.4 | 26 | 30.7 | 24.7 | 52.8 | 36.9 | 38.6 | 40.7 | 26.1 | 23.3 | 24.1 | 27 | 32.6 |
| Ni | mg/l | 0.01 | < | < | < | 0.39 | 0.04 | < | < | 0.02 | 0.04 | < | < | 0.01 | 0.03 | < | < | < | < |
| P | mg/l | 0.1 | < | < | < | < | < | < | < | < | < | < | < | < | < | < | < | < | < |
| Pb | mg/l | 0.0005 | 0.0014 | 0.0029 | 0.0018 | 0.0012 | 0.0011 | 0.0012 | 0.0037 | 0.0009 | 0.0008 | < | 0.002 | 0.0012 | 0.001 | 0.0022 | 0.001 | 0.0007 | 0.0007 |
| $\mathrm{SO}_{4}$ | mg/l | 0.3 | 5.6 | 18 | 3.8 | 64.5 | 16 | 12 | 47.2 | 13.7 | 399 | 4.6 | 10.5 | 4.2 | 38.1 | 5.9 | 20.9 | 22.9 | 11.4 |
| Sb | mg/l | 0.00001 | 0.00056 | 0.00055 | 0.00014 | 0.0022 | 0.00045 | 0.00064 | 0.00157 | 0.00119 | 0.00005 | 0.00013 | 0.00012 | 0.00041 | 0.00097 | 0.001 | 0.00068 | 0.0023 | 0.00078 |
| Se | mg/l | 0.0005 | 0.0083 | 0.0129 | 0.0012 | 0.0308 | 0.02 | 0.0079 | 0.0054 | 0.0211 | 0.0099 | 0.001 | 0.0008 | 0.0013 | 0.0019 | 0.0046 | 0.002 | 0.0202 | 0.0055 |
| Si | mg/l | 0.05 | 6.6 | 5.63 | 4 | 3.89 | 5.17 | 3.71 | 7.13 | 2.76 | 1.05 | 8.47 | 8.68 | 12.31 | 2.94 | 8.49 | 2.85 | 7.36 | 3.11 |
| Sn | mg/l | 0.0001 | < | 0.0001 | 0.0002 | < | < | 0.0002 | 0.0002 | 0.0001 | 0.0001 | 0.0001 | 0.0002 | 0.0002 | < | 0.0003 | < | 0.0001 | < |
| Sr | mg/l | 0.00002 | 0.01776 | 0.02972 | 0.02662 | 0.0513 | 0.01308 | 0.02523 | 0.06933 | 0.03106 | 0.64967 | 0.00711 | 0.02895 | 0.01665 | 0.11547 | 0.07286 | 0.09399 | 0.08628 | 0.04105 |
| Th | mg/l | 0.000005 | 0.000276 | 0.000468 | 0.000554 | 0.000192 | 0.000288 | 0.001068 | 0.00136 | 0.000153 | 0.000092 | 0.000168 | 0.000488 | 0.000625 | 0.000082 | 0.000691 | 0.000146 | 0.000191 | 0.00009 |
| U | mg/l | 0.000005 | 0.000249 | 0.000162 | 0.000157 | 0.00013 | 0.000096 | 0.000268 | 0.000583 | 0.000173 | 0.000152 | 0.000134 | 0.000206 | 0.000371 | 0.000699 | 0.000351 | 0.000211 | 0.000244 | 0.000128 |
| Zn | mg/l | 0.01 | < | < | < | 0.02 | < | < | < | < | < | < | < | < | < | < | < | < | < |

[^3]Table 7: Soluble/exchangeable cations and EAT classes of selected overburden and floor samples.

| EGi Sample | Unit | Material | Geological Description | EC | sol Na | sol K | sol Ca | sol Mg | ex Na | ex K | ex Ca | ex Mg | ex AI | $\begin{gathered} \% \\ \text { ECEC } \end{gathered}$ | $\begin{gathered} \% \\ \text { ECEC } \end{gathered}$ | $\begin{gathered} \% \\ \text { ECEC } \end{gathered}$ | $\begin{gathered} \% \\ \text { ECEC } \end{gathered}$ | $\begin{gathered} \% \\ \text { ECEC } \end{gathered}$ | ECEC | $\mathrm{Ca} / \mathrm{Mg}$ | $\begin{array}{r} \mathrm{Em} \\ \text { Aggre } \end{array}$ | merson <br> egate Test <br> Class |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  |  | (dS/m) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (meq\%) | (ESP) |  |  |  |  |  |  | Class | Sub-Class |
| 28492 | Permian | Overburden | Claystone/Tuff | 0.06 | 0.06 | 0.06 | < 0.08 | 0.06 | 0.11 | 0.55 | 3.70 | 5.63 | < 0.02 | 1.1 | 5.5 | 37.0 | 56.4 | 0.0 | 10.0 | 0.70 | 5 | 3 |
| 28497 | Permian | Overburden | Sandstone | 0.09 | 0.07 | 0.12 | 0.16 | 0.22 | 0.03 | 0.36 | 1.27 | 2.01 | < 0.02 | 0.8 | 9.8 | 34.6 | 54.8 | 0.0 | 3.7 | 0.60 | 5 | 1 |
| 28502 | Permian | Overburden | Mudstone | 0.09 | 0.10 | 0.08 | 0.08 | 0.09 | 0.08 | 0.47 | 1.77 | 2.78 | < 0.02 | 1.6 | 9.2 | 34.7 | 54.5 | 0.0 | 5.1 | 0.60 | 5 | 2 |
| 28506 | Permian | Overburden | Siltstone Weathered | 0.08 | 0.86 | 0.12 | < 0.08 | 0.37 | 0.50 | 0.51 | 1.84 | 8.13 | < 0.02 | 4.6 | 4.6 | 16.8 | 74.0 | 0.0 | 11.0 | 0.20 | 5 | 3 |
| 28513 | Permian | Overburden | Claystone/Tuff | 0.11 | 0.20 | 0.06 | 0.11 | 0.15 | 0.15 | 0.42 | 4.43 | 5.88 | < 0.02 | 1.4 | 3.9 | 40.7 | 54.0 | 0.0 | 10.9 | 0.80 | 5 | 2 |
| 28542 | Permian | Overburden | Intrusion - chloritic in part | 0.11 | 0.62 | 0.28 | 2.12 | 2.00 | 0.23 | 0.79 | 8.91 | 7.80 | < 0.02 | 1.3 | 4.5 | 50.3 | 44.0 | 0.0 | 17.7 | 1.10 | 5 | 2 |
| 30374 | Quaternary/Tertiary | Overburden | Alluvium/weathered Permian | 0.09 | 0.20 | 0.01 | < 0.08 | 0.02 | 0.63 | 0.20 | 0.34 | 3.98 | 0.26 | 11.6 | 3.7 | 6.3 | 73.6 | 4.8 | 5.4 | 0.10 | 5 | 2 |
| 30377 | Permian | Overburden | Weathered Permian | 0.06 | 0.71 | 0.01 | < 0.08 | 0.16 | 0.95 | 0.12 | < 0.20 | 5.16 | < 0.02 | 15.0 | 1.9 | 1.4 | 81.6 | 0.0 | 6.3 | 0.00 | 3 |  |
| 30380 | Quaternary/Tertiary | Overburden | Alluvium | 0.07 | 0.81 | 0.06 | < 0.08 | 0.62 | 0.91 | 0.15 | 0.36 | 3.28 | < 0.02 | 19.4 | 3.2 | 7.7 | 69.8 | 0.0 | 4.7 | 0.10 | 2 |  |
| 30383 | Quaternary/Tertiary | Overburden | Alluvium | 0.08 | 1.74 | 0.02 | < 0.08 | 0.07 | 1.16 | 0.23 | < 0.20 | 2.91 | < 0.02 | 26.6 | 5.3 | 1.4 | 66.7 | 0.0 | 4.4 | 0.00 | 5 | 2 |
| 30386 |  |  | Top Soil | 0.09 | 0.29 | 0.02 | < 0.08 | 0.06 | 0.12 | 0.07 | 0.61 | 0.69 | < 0.02 | 8.1 | 4.7 | 40.9 | 46.3 | 0.0 | 1.5 | 0.90 | 3 |  |
| 30388 | Quaternary/Tertiary | Overburden | Silty Clay | 0.07 | 0.28 | 0.01 | < 0.08 | 0.02 | 0.23 | 0.07 | 0.39 | 1.41 | < 0.02 | 11.0 | 3.3 | 18.6 | 67.1 | 0.0 | 2.1 | 0.30 | 5 | 2 |
| 30394 | Permian | Overburden | Weathered Permian Silty Clay | 0.13 | 0.38 | 0.04 | < 0.08 | 0.04 | 0.33 | 0.17 | 1.39 | 2.52 | < 0.02 | 7.5 | 3.9 | 31.5 | 57.1 | 0.0 | 4.4 | 0.60 | 5 | 2 |
| 30399 | Permian | Overburden | Clay | 0.12 | 0.14 | 0.04 | < 0.08 | 0.02 | 0.23 | 0.29 | 1.94 | 3.79 | 0.09 | 3.6 | 4.6 | 30.6 | 59.8 | 1.4 | 6.3 | 0.50 | 5 | 2 |
| 30403 | Permian | Overburden | Carbonaceous Shale | 0.09 | 0.12 | 0.07 | 0.20 | 0.17 | 0.08 | 0.35 | 6.33 | 4.61 | < 0.02 | 0.7 | 3.1 | 55.7 | 40.5 | 0.0 | 11.4 | 1.40 | 5 | 2 |
| 30418 | Permian | Overburden | Sandstone | 0.10 | 0.07 | 0.08 | 0.17 | 0.11 | 0.02 | 0.21 | 3.15 | 1.71 | < 0.02 | 0.4 | 4.1 | 61.9 | 33.6 | 0.0 | 5.1 | 1.80 | 5 | 2 |
| 30422 | Permian | Overburden | Carbonaceous Shale | 0.12 | 0.13 | 0.10 | 0.11 | 0.08 | 0.11 | 0.78 | 4.71 | 2.81 | <0.02 | 1.3 | 9.3 | 56.0 | 33.4 | 0.0 | 8.4 | 1.70 | 5 | 3 |

Table 8: Geochemical characterisation results for Ulan Coal Seam samples from hole TB103.

| EGi Code | Depth (m) |  |  | Geological Description | $\mathrm{pH}_{1: 2}$ | $\mathrm{EC}_{1: 2}$ | ACID-BASE ANALYSIS |  |  |  |  | NAG TEST |  |  | NAGorg TEST |  | ARD <br> Classification |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  | From | To | Interval |  |  |  | Total \%S | MPA | ANC | NAPP | ANC/MPA | NAGpH | NAG ${ }_{(\text {(pH4.5) }}$ | NAG ${ }_{(\text {(pH7.O) }}$ | Extended <br> Boil <br> NAGpH | Calculated NAG |  |
| 30425 | 114.00 | 115.00 | 1.00 | Coal - Ulan Seam | 6.81 | 0.12 | 0.14 | 4 | 11 | -7 | 2.57 | 8.0 | 0 | 0 |  |  | NAF |
| 30426 | 115.00 | 118.00 | 3.00 | Coal - Ulan Seam | 5.21 | 1.46 | 0.30 | 9 | 7 | 2 | 0.76 | 6.7 | 0 | 0 |  |  | UC(NAF) |
| 30427 | 118.00 | 121.00 | 3.00 | Coal - Ulan Seam | 5.11 | 1.31 | 0.41 | 13 | 6 | 7 | 0.48 | 3.0 | 11 | 28 |  |  | PAF-LC |
| 30428 | 121.00 | 124.00 | 3.00 | Coal - Ulan Seam | 4.87 | 1.88 | 0.54 | 17 | 5 | 12 | 0.30 | 2.8 | 20 | 46 | 6.9 | 0 | UC(NAF) |
| 30429 | 124.00 | 127.00 | 3.00 | Coal - Ulan Seam | 4.71 | 1.91 | 0.66 | 20 | 2 | 18 | 0.10 | 2.2 | 54 | 86 | 7.0 | 4 | PAF-LC |
| 30430 | 127.00 | 128.00 | 1.00 | Coal - Ulan Seam | 4.51 | 1.97 | 0.60 | 18 | 3 | 15 | 0.16 | 2.4 | 34 | 57 |  |  | PAF-LC |
| KEY |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{pH}_{1: 2}=\mathrm{pH}$ of 1:2 extract |  |  |  |  | NAGpH $=$ pH of NAG liquor |  |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{EC}_{1: 2}=$ Electrical Conductivity of 1:2 extract ( $\mathrm{dS} / \mathrm{m}$ ) |  |  |  |  | $\mathrm{NAG}_{(\text {PH4.5) }}=$ Net Acid Generation capacity to $\mathrm{pH} 4.5\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |  |
| MPA $=$ Maximum Potential Acidity ( $\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ) |  |  |  |  | $\mathrm{NAG}_{(\text {(H77.0) }}=$ Net Acid Generation capacity to $\mathrm{pH} 7.0\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |  |
| ANC = Acid Neutralising Capacity ( $\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ) |  |  |  |  | $\square$ = NAG Result Affected by High Organic Carbon in Sample |  |  |  |  |  |  |  |  |  |  |  |  |
| NAPP $=$ Net Acid Producing Potential $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  | Extended Boil NAGpH $=\mathrm{pH}$ of NAG liquor after extended heating |  |  |  |  |  |  |  |  |  |  |  |  |
|  |  |  |  |  | Calculated NAG = The net acid potential based on assay of anions and cations released to the NAG solution ( $\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ) |  |  |  |  |  |  |  |  |  |  |  |  |

Table 9: Geochemical characterisation results for coal reject samples.

| Hole No | Open Cut Pit No | $\begin{gathered} \text { EGi } \\ \text { Sample } \\ \text { No } \end{gathered}$ | Carbon Consulting Sample No | Moolarben Samples Used in Coal Composite | Coal Product Description | pH1:2 | $\mathrm{EC}_{1: 2}$ | ACID-BASE ANALYSIS |  |  |  |  | NAG TEST |  |  | NAGorg TEST |  | $\begin{gathered} \text { ARD } \\ \text { Classification } \end{gathered}$ |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  |  |  |  |  |  | $\begin{array}{\|c} \text { Total } \\ \% \mathrm{~S} \end{array}$ | MPA | ANC | NAPP | ANC/MPA | NAGpH | NAG $_{\text {(prat.5) }}$ | NAG $_{\text {(phr } 00}$ | Extended Boil NAGpH | NAG <br> Calculated NAG |  |
| WMLB24 | OC1 | 30372 | 39141 | WML B24-3, 4 \& 6 | High ash domestic thermal product (Ulan Seams A to C) | 5.85 | 0.31 | 0.27 | 8 | 1 | 7 | 0.12 | 2.8 | 14 | 27 | 9 | 6 | PAF |
| WMLB24 | OC1 | 30373 | 39147 | WML B24-7, 8, 9 \& 10 | Export thermal product (Ulan Seams D to E) | 4.21 | 0.37 | 1.06 | 32 | 0 | 32 | 0.00 | 2.5 | 29 | 32 |  | 26 | PAF |
| KEY |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{pH}_{1: 2}=\mathrm{pH}$ of 1:2 extract |  |  |  |  | NAGpH $=\mathrm{pH}$ of NAG liquor |  |  |  |  |  |  |  |  |  |  |  |  |  |
| $\mathrm{EC}_{1: 2}=$ Electrical Conductivity of 1:2 extract (dS/m) |  |  |  |  | $\mathrm{NAG}_{(\text {(r4. } 4 \text { ) }}=$ Net Acid Generation capacity to $\mathrm{pH} 4.5\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  |  |  |  |  |  |  |  |  |  |
| MPA = Maximum Potential Acidity ( $\left.\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  | $\mathrm{NAG}_{(\text {(pH7.0) }}=$ Net Acid Generation capacity to $\mathrm{pH} 7.0\left(\mathrm{kgH}_{2} \mathrm{SO} / \mathrm{l}\right)$ |  |  |  |  |  |  |  |  |  |  |  |  |  |
| ANC = Acid Neutralising Capacity ( $\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ) |  |  |  |  | $\square=$ NAG Result Affected by High Organic Carbon in Sample |  |  |  |  |  |  |  |  |  |  |  |  |  |
| NAPP = Net Acid Producing Potential ( $\left.\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$ |  |  |  |  | Extended Boil NAGpH $=$ pH of NAG liquor after extended heating |  |  |  |  |  |  |  |  |  |  |  |  |  |
|  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |  |

Table 10: Total S(\%) contents of washed coal from drill testing of the proposed open cut development (data provided by Moolarben Coal).

| Seam Section | Open Cut 1 |  |  |  |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  | wmlb22 | dmmld2 | dmmld7 | dmmld8 | dmmlx-1 | wmlb23 | wmlb24 | wmlb25 | wmlb28 | wmlb29 |
| A-CL | 0.43 |  | 0.36 | 0.37 |  | 0.44 | 0.42 | 0.46 | 0.41 |  |
| DTP-ELR A-ELR | 0.44 | 0.44 | 0.47 | 0.54 | 0.45 | 0.48 | 0.44 | 0.50 | 0.45 | 0.50 |


| Seam Section | Open Cut 2 |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  | wmlb11 | wmlb13 | wmlb14 | dmmld3 | dmmld5 | dmmld4 | wmlb16 |
| A-CL | 0.40 | 0.34 | 0.36 | 0.41 | 0.44 | 0.43 | 0.44 |
| DTP-ELR <br> A-ELR | 0.50 | 0.47 | 0.45 |  | 0.36 | 0.44 |  |
| SeamSection | Open Cut 3 |  |  |  |  |  |  |
|  | wmlb3 | wmlb4 | wmlb5 | wmlb6 | wmlb7 | wmlb8 | wmlb10 |
| A-CL | 0.38 | 0.42 |  | 0.39 |  | 0.54 | 0.35 |
| DTP-ELR |  | 0.48 |  | 0.37 |  | 0.36 | 0.40 |
| A-ELR |  |  | 0.44 |  | 0.36 |  |  |

Total S Summary Statististics
Maximum $=0.54 \% \mathrm{~S}$
$\begin{aligned} \text { Minimum } & =0.34 \% \mathrm{~S} \\ \text { Oth Percentile } & =0.50 \% \mathrm{~S}\end{aligned}$
90thedian $=0.44 \% \mathrm{~S}$
10th Percentile $=0.36 \% \mathrm{~S}$

Table 11: Total S(\%) contents of raw coal from drill testing of the proposed underground development (data provided by Moolarben Coal).

| No 4 UG Ulan Seam Raw Ply Sulphur |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| Drillhole | Sample No | From (m) | To (m) | Thickness (m) | Ply | \%S |
| C200 | 11321 | 137.50 | 138.12 | 0.62 | ULA(part) | 0.32 |
| C200 | 11322 | 138.12 | 139.60 | 1.48 | UB | 0.32 |
| C200 | 11323 | 139.60 | 141.80 | 2.20 | UC | 0.33 |
| C200 | 11324 | 141.80 | 142.10 | 0.30 | CMK | 0.60 |
| C200 | 11325 | 142.10 | 142.85 | 0.75 | UCL | 0.75 |
| C200 | 11326 | 142.85 | 143.65 | 0.80 | UD | 0.45 |
| C200 | 11327 | 143.65 | 144.60 | 0.95 | UD | 0.47 |
| C200 | 11328 | 144.60 | 145.50 | 0.90 | UD | 1.71 |
| C200 | 11329 | 145.50 | 146.03 | 0.53 | UD | 1.10 |
| C200 |  | 146.03 | 146.73 | 0.70 |  |  |
| C200 | 11330 | 146.73 | 147.37 | 0.64 | UE | 0.90 |
| C202 | 11391 | 154.40 | 156.20 | 1.80 | ULA | 0.28 |
| C202 | 1 | 156.20 | 158.42 | 2.22 | UB | 0.37 |
| C202 | 2 | 158.42 | 160.00 | 1.58 | UC |  |
| C202 | 3 | 160.00 | 160.30 | 0.30 | CMK |  |
| C202 | 4 | 160.30 | 160.70 | 0.40 | UC | 0.53 |
| C202 | 5 | 160.70 | 163.25 | 2.55 | UD | 0.60 |
| C202 | 6 | 163.25 | 164.25 | 1.00 | UD | 0.72 |
| C202 | 11392 | 164.25 | 166.00 | 1.75 | UE |  |
| C204 | 1 | 91.10 | 93.10 | 2.00 | ULA | 0.32 |
| C204 | 2 | 93.10 | 94.77 | 1.67 | UB | 0.43 |
| C204 | 3+4 | 94.77 | 95.38 | 0.61 | UC1 | 0.43 |
| C204 | 5 | 95.38 | 95.75 | 0.37 | UC2 | 0.27 |
| C204 | 6 | 95.75 | 96.94 | 1.19 | UC2 | 0.51 |
| C204 | 7 | 96.94 | 97.37 | 0.43 | CMK | 0.20 |
| C204 | 8 | 97.37 | 98.01 | 0.64 | UCL | 0.85 |
| C204 | 9 | 98.01 | 99.28 | 1.27 | DTP+DWS | 0.47 |
| C204 | 10 | 99.28 | 100.45 | 1.17 | DWS | 0.75 |
| C204 | 11 | 100.45 | 101.44 | 0.99 | DWS | 0.51 |
| C204 | 12 | 101.44 | 102.32 | 0.88 | ETP | 0.60 |
| C204 | 13 | 102.32 | 103.18 | 0.86 | EBT | 0.49 |
| C204 | 14 | 103.18 | 103.45 | 0.27 | ELW | 0.47 |
| C204 | 15 | 103.45 | 103.99 | 0.54 | ELW | 0.13 |
| C221 | 10189 | 141.86 | 142.92 | 1.06 | ULA | 0.29 |
| C221 | 10190 | 142.92 | 143.65 | 0.73 | UB1 | 0.34 |
| C221 | 10191 | 143.65 | 144.50 | 0.85 | UB2 | 0.46 |
| C221 | 10192 | 144.50 | 145.05 | 0.55 | UC1 | 0.40 |
| C221 | 10193 | 145.05 | 146.55 | 1.50 | UC2 | 0.30 |
| C221 |  | 146.55 | 146.90 | 0.30 | CMK |  |
| C221 | 10194 | 146.90 | 147.66 | 0.76 | UCL | 0.44 |
| C221 | 10195 | 147.66 | 148.16 | 0.50 | DTP | 0.42 |
| C221 | 10196 | 148.16 | 149.60 | 1.44 | DWS1 | 0.50 |
| C221 | 10197 | 149.60 | 150.30 | 0.70 | DWS2 | 0.89 |
| C221 | 10198 | 150.30 | 150.85 | 0.55 | DWS3 | 0.63 |
| C221 | 10199 | 150.85 | 151.95 | 1.10 | ETP | 1.26 |
| C221 | 10200 | 151.95 | 152.70 | 0.75 | EBT | 0.54 |
| C223 | 10001 | 57.80 | 59.87 | 2.00 | ULA | 0.28 |
| C223 | 10002 | 59.87 | 60.70 | 0.90 | UB1 | 0.41 |
| C223 | 10003 | 60.70 | 61.50 | 0.80 | UB2 | 0.35 |
| C223 | 10004 | 61.50 | 62.08 | 0.58 | UC1 | 0.40 |
| C223 | 10005 | 62.08 | 63.68 | 1.60 | UC2 | 0.34 |
| C223 |  | 63.68 | 63.96 | 0.28 | CMK |  |
| C223 | 10006 | 63.96 | 64.90 | 0.94 | UCL | 0.33 |
| C223 | 10007 | 64.90 | 65.20 | 0.30 | DTP | 0.44 |
| C223 | 10008 | 65.20 | 66.50 | 1.30 | DWS1 | 0.50 |
| C223 | 10009 | 66.50 | 67.05 | 0.55 | DWS2 | 0.43 |

Table 11: Total S(\%) contents of raw coal from drill testing of the proposed underground development (data provided by Moolarben Coal).

| No 4 UG Ulan Seam Raw Ply Sulphur |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| Drillhole | Sample No | From (m) | To (m) | Thickness (m) | Ply | \%S |
| C223 | 10010 | 67.05 | 68.05 | 1.00 | DWS3 | 0.46 |
| C223 | 10011 | 68.05 | 69.00 | 0.95 | ETP | 0.39 |
| C223 | 10012 | 69.00 | 69.90 | 0.90 | EBT | 0.53 |
| C224 | 10014 | 74.90 | 77.25 | 2.35 | ULA | 0.33 |
| C224 | 10015 | 77.25 | 77.90 | 0.65 | UB1 | 0.41 |
| C224 | 10016 | 77.90 | 78.90 | 1.00 | UB2 | 0.41 |
| C224 | 10017 | 78.90 | 79.40 | 0.50 | UC1 | 0.41 |
| C224 | 10018 | 79.40 | 80.90 | 1.50 | UC2 | 0.40 |
| C224 | 10019 | 81.25 | 82.00 | 0.75 | UCL | 0.57 |
| C224 |  | 81.90 | 81.25 | 0.35 | CMK |  |
| C224 | 10020 | 82.00 | 82.25 | 0.25 | DTP | 0.58 |
| C224 | 10021 | 82.25 | 82.90 | 0.65 | DWS1 | 0.59 |
| C224 | 10022 | 82.90 | 84.60 | 1.70 | DWS2 | 0.64 |
| C224 | 10023 | 84.60 | 85.20 | 0.60 | DWS3 | 0.47 |
| C224 | 10024 | 85.20 | 85.94 | 0.74 | ETP | 0.56 |
| C224 | 10025 | 85.94 | 86.85 | 0.91 | EBT | 0.71 |
| C224 | 10026 | 86.85 | 87.45 | 0.60 | ELW | 0.67 |
| C225 | 10027 | 123.68 | 124.30 | 0.62 | UCL | 0.48 |
| C225 | 10028 | 124.30 | 124.60 | 0.30 | DTP | 0.69 |
| C225 | 10029 | 124.60 | 125.60 | 1.00 | DWS1 | 0.42 |
| C225 | 10030 | 125.60 | 126.70 | 1.10 | DWS2 | 0.50 |
| C225 | 10031 | 126.70 | 127.65 | 0.95 | DWS3 | 0.44 |
| C225 | 10032 | 127.65 | 128.60 | 0.95 | ETP | 0.54 |
| C225 | 10033 | 128.60 | 129.50 | 0.90 | EBT | 0.51 |
| C225 | 10034 | 129.50 | 130.25 | 0.75 | ELW | 0.47 |
| C227 | 10048 | 138.20 | 140.65 | 2.45 | ULA | 0.29 |
| C227 | 10049 | 140.65 | 141.30 | 0.65 | UB1 | 0.35 |
| C227 | 10050 | 141.30 | 142.10 | 0.80 | UB2 | 0.38 |
| C227 | 10051 | 142.10 | 142.80 | 0.70 | UC1 | 0.40 |
| C227 | 10052 | 142.80 | 144.35 | 1.55 | UC2 | 0.46 |
| C227 |  | 144.35 | 144.63 | 0.28 | CMK |  |
| C227 | 10053 | 144.63 | 145.40 | 0.77 | UCL | 0.38 |
| C227 | 10054 | 145.40 | 145.66 | 0.26 | DTP | 0.51 |
| C227 | 10055 | 145.66 | 146.54 | 0.88 | DWS1 | 0.67 |
| C227 | 10056 | 146.54 | 147.70 | 1.16 | DWS2 | 0.59 |
| C227 | 10057 | 147.70 | 148.60 | 0.90 | DWS3 | 0.48 |
| C227 | 10058 | 148.60 | 149.61 | 1.01 | ETP | 0.78 |
| C227 | 10059 | 149.61 | 150.45 | 0.84 | EBT | 0.65 |
| C229 | 10087 | 130.50 | 131.00 | 0.50 | UB1 | 0.32 |
| C229 | 10088 | 131.00 | 132.13 | 1.13 | UB2 | 0.34 |
| C229 | 10089 | 132.13 | 132.75 | 0.62 | UC1 | 0.32 |
| C229 | 10090 | 132.75 | 134.20 | 1.45 | UC2 | 0.39 |
| C229 |  | 134.20 | 134.48 | 0.28 | CMK |  |
| C229 | 10091 | 134.48 | 135.30 | 0.82 | UCL | 0.58 |
| C229 | 10092 | 135.30 | 135.55 | 0.25 | DTP | 2.01 |
| C229 | 10093 | 135.55 | 136.43 | 0.88 | DWS1 | 0.48 |
| C229 | 10094 | 136.43 | 137.55 | 1.12 | DWS2 | 0.51 |
| C229 | 10095 | 137.55 | 138.40 | 0.85 | DWS3 | 0.51 |
| C229 | 10096 | 138.40 | 139.50 | 1.10 | ETP | 0.94 |
| C229 | 10097 | 139.50 | 140.35 | 0.85 | EBT | 0.62 |
| C229 | 10098 | 140.35 | 140.85 | 0.50 | ELW | 0.45 |
| C230 | 10099 | 137.35 | 139.80 | 2.45 | ULA | 0.31 |
| C230 | 10100 | 139.80 | 140.55 | 0.75 | UB1 | 0.36 |
| C230 | 10179 | 140.55 | 141.36 | 0.81 | UB2 | 0.44 |
| C230 | 10180 | 141.36 | 142.00 | 0.64 | UC1 | 0.33 |
| C230 | 11393 | 142.00 | 143.55 | 1.55 | UC2 | 0.40 |

Table 11: Total S(\%) contents of raw coal from drill testing of the proposed underground development (data provided by Moolarben Coal).

| No 4 UG Ulan Seam Raw Ply Sulphur |  |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| Drillhole | Sample No | From (m) | To (m) | Thickness (m) | Ply | \%S |
| C230 |  | 143.55 | 143.91 | 0.28 | CMK |  |
| C230 | 11394 | 143.91 | 144.68 | 0.77 | UCL | 0.43 |
| C230 | 11395 | 144.68 | 144.90 | 0.22 | DTP | 0.97 |
| C230 | 11396 | 144.90 | 146.20 | 1.30 | DWS1 | 0.67 |
| C230 | 11397 | 146.20 | 147.33 | 1.13 | DWS2 | 0.74 |
| C230 | 11398 | 147.33 | 147.93 | 0.60 | DWS3 | 0.49 |
| C230 | 11399 | 147.93 | 148.95 | 1.02 | ETP | 0.72 |
| C230 | 11400 | 148.95 | 149.80 | 0.85 | EBT | 0.66 |
| C238 |  | 169.75 | 170.09 | 0.34 | dtp | 0.38 |
| C238 |  | 170.09 | 171.15 | 1.06 | dws1 | 0.53 |
| C238 |  | 171.15 | 172.15 | 1.00 | dws2 | 0.69 |
| C238 |  | 172.15 | 173.10 | 0.95 | dws3 | 0.60 |
| C238 |  | 173.10 | 173.95 | 0.85 | ETP | 0.60 |
| C238 |  | 173.95 | 174.70 | 0.75 | ebt | 0.42 |
| C238 |  | 174.70 | 175.65 | 0.95 | ELW | 0.38 |
| C239 |  | 162.85 | 163.15 | 0.30 | dtp | 0.40 |
| C239 |  | 163.15 | 164.45 | 1.30 | dws1 | 0.49 |
| C239 |  | 164.45 | 165.10 | 0.65 | dws2 | 0.50 |
| C239 |  | 165.10 | 166.10 | 1.00 | dws3 | 0.44 |
| C239 |  | 166.10 | 167.00 | 0.90 | ETP | 1.07 |
| C239 |  | 167.00 | 167.80 | 0.80 | ebt | 0.75 |
| C240 | 10346 | 144.20 | 146.35 | 2.15 | ULA | 0.35 |
| C240 | 10347 | 146.35 | 146.95 | 0.60 | UB1 | 0.40 |
| C240 | 10348 | 146.95 | 147.85 | 0.90 | UB2 | 0.39 |
| C240 | 10349 | 147.85 | 148.40 | 0.55 | UC1 | 0.45 |
| C240 | 10350 | 148.40 | 149.90 | 1.50 | UC2 | 0.33 |
| C240 |  | 149.90 | 150.20 | 0.30 | CMK |  |
| C240 |  | 150.20 | 151.00 | 0.80 | UCL |  |
| C240 |  | 151.00 | 151.60 | 0.60 | DTP + DWS |  |
| C240 |  | 151.60 | 154.15 | 2.65 | DWS |  |
| C240 | 10356 | 154.15 | 155.10 | 0.95 | ETP | 1.06 |
| C240 | 10357 | 155.10 | 155.95 | 0.85 | EBT | 0.57 |
| C240 | 10358 | 155.95 | 156.65 | 0.70 | ELW | 0.93 |
| C245 | 10878 | 113.14 | 115.61 | 2.47 | ULA | 0.34 |
| C245 | 10879 | 115.61 | 116.20 | 0.59 | UB1 | 0.35 |
| C245 | 10880 | 116.20 | 117.20 | 1.00 | UB2 | 0.37 |
| C245 | 10899 | 117.20 | 117.70 | 0.50 | UC1 | 0.55 |
| C245 | 10900 | 117.70 | 119.25 | 1.55 | UC2 | 0.35 |
| C245 | 10501 | 119.57 | 120.73 | 1.16 | UCL | 0.45 |
| C245 | 10502 | 120.73 | 121.03 | 0.30 | DTP | 0.45 |
| C245 | 10503 | 121.03 | 122.07 | 1.04 | DWS1 | 0.59 |
| C245 | 10504 | 122.07 | 123.00 | 0.93 | DWS2 | 0.46 |
| C245 | 10505 | 123.00 | 123.55 | 0.55 | DWS3 | 0.62 |
| C245 | 10506 | 123.55 | 124.41 | 0.86 | ETP | 1.19 |
| C245 | 10507 | 124.41 | 125.04 | 0.63 | EBT | 0.60 |
| C268 | 11070 | 205.40 | 206.30 | 0.90 | DWS1 | 0.42 |
| C268 | 11071 | 206.30 | 207.21 | 0.91 | DWS2 | 0.47 |
| C268 | 11072 | 207.21 | 208.09 | 0.88 | DWS3 | 0.42 |
| C268 | 11073 | 208.09 | 208.38 | 0.29 | DWS4 | 0.46 |

Total S Summary Statististics

| Maximum $=$ | $2.01 \% \mathrm{~S}$ |
| ---: | ---: |
| Minimum $=$ | $0.13 \% \mathrm{~S}$ |
| 90th Percentile $=$ | $0.75 \% \mathrm{~S}$ |
| Median $=$ | $0.47 \% \mathrm{~S}$ |
| 10th Percentile $=$ | $0.33 \% \mathrm{~S}$ |

Table 12: Multi-element composition (mg/kg except where shown) and geochemical abundance index (GAI) of coal and reject sample solids.

| Element | Detection Limit | EGi Sample Number and Sample Description |  |  |  | Element | Median Soil Abundance* | EGi Sample Number and Sample Description |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | 30428 | 30430 | 30372 | 30373 |  |  | 30428 | 30430 | 30372 | 30373 |
|  |  | Coal (Ulan Seam) | Coal (Ulan Seam) | High ash domestic thermal product (Ulan Seams A to C) | Export thermal product (Ulan Seams D to E) |  |  | Coal (Ulan Seam) | Coal (Ulan Seam) | High ash domestic thermal product (Ulan Seams A to C) | Export thermal product <br> (Ulan Seams D to E) |
| Ag | 0.1 | <0.1 | <0.1 | <0.7 | <0.3 | Ag | 0.05 | - | - | 3 | 2 |
| Al | 0.002\% | 2.88\% | 4.88\% | 7.40\% | 14.30\% | Al | 7.1\% | - | - | - | - |
| As | 1 | 2 | 3 | 4 | 8 | As | 6 | - | - | - | - |
| Ba | 0.1 | 52 | 93 | 57 | 41 | Ba | 500 | - | - | - | - |
| Be | 0.1 | 1.5 | 3.9 | 1 | 3 | Be | 0.3 | 2 | 3 | 1 | 3 |
| Ca | 0.001\% | 0.09\% | 0.05\% | 0.02\% | 0.03\% | Ca | 1.5\% | - | - | - | - |
| Cd | 0.1 | <0.1 | <0.1 | <0.6 | <0.8 | Cd | 0.35 | - | - | - | 1 |
| Co | 0.1 | 2.0 | 3.4 | 6.9 | 3.5 | Co | 8 | - | - | - | - |
| Cr | 2.0 | 9 | 17 | 6 | 2 | Cr | 70 | - | - | - | - |
| Cu | 1.0 | 8 | 12 | 174 | 83 | Cu | 30 | - | - | 2 | 1 |
| F | 50 | 228 | 230 | 342 | 534 | F | 200 | - | - | - | 1 |
| Fe | 0.01\% | 0.88\% | 0.59\% | 0.26\% | 0.74\% | Fe | 4.0\% | - | - | - | - |
| Hg | 0.01 | <0.10 | <0.11 | <0.13 | 0.26 | Hg | 0.06 | - | - | 1 | 2 |
| K | 0.002\% | 0.23\% | 0.55\% | 0.15\% | 0.25\% | K | 1.4\% | - | - | - | - |
| Mg | 0.002\% | 0.08\% | 0.07\% | 0.03\% | 0.03\% | Mg | 0.5\% | - | - | - | - |
| Mn | 1.0 | 174 | 65 | 18 | 7 | Mn | 1000 | - | - | - | - |
| Mo | 0.1 | 0.8 | 1.2 | 1.3 | 2.7 | Mo | 1.2 | - | - | - | 1 |
| Na | 0.002\% | 0.01\% | 0.02\% | 0.01\% | 0.00\% | Na | 0.5\% | - | - | - | - |
| Ni | 1.0 | 9 | 9 | 4 | 7 | Ni | 50 | - | - | - | - |
| P | 20 | 72 | 76 | 103 | $<175$ | P | 800 | - | - | - | - |
| Pb | 2.0 | 10 | 15 | 20 | <42 | Pb | 35 | - | - | - | - |
| S | 0.001\% | 0.33\% | 0.52\% | 0.28\% | 1.04\% | S | 0.07\% | 2 | 2 | 1 | 3 |
| Sb | 0.05 | 0.35 | 0.61 | 0.44 | 0.59 | Sb | 1 | - | - | - | - |
| Se | 0.01 | 0.31 | <0.39 | 0.6 | 2.25 | Se | 0.4 | - | - | - | 2 |
| Si | 0.1\% | 11.9\% | 12.7\% | 27.2\% | 20.0\% | Si | 33.0\% | - | - | - | - |
| Sn | 0.1 | 1.4 | 2.8 | 19.1 | 19.2 | Sn | 4 | - | - | 2 | 2 |
| Sr | 0.05 | 19.4 | 25.1 | 24.6 | 51.1 | Sr | 250 | - | - | - | - |
| Th | 0.01 | 5.56 | 9.93 | 11.54 | 30.88 | Th | 9 | - | - | - | 1 |
| U | 0.01 | 1.77 | 2.96 | 3.23 | 6.58 | U | 2 | - | - | - | 1 |
| Zn | 1.0 | 19 | 34 | 103 | 118 | Zn | 90 | - | - | - | - |

Table 13: Chemical composition of coal and reject sample water extracts.

| Parameter |  | Detection Limit | EGi Sample Number and Sample Description |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | 30428 | 30430 | 30372 | 30373 |
|  |  | Coal (Ulan Seam) | Coal (Ulan Seam) | High ash domestic thermal product (Ulan Seams A to C) | Export thermal product (Ulan Seams D to E) |
| pH |  |  | 0.01 | 5.11 | 5.02 | 5.9 | 4.2 |
| EC | dS/m |  | 0.01 | 1.92 | 1.99 | 0.307 | 0.366 |
| Ag | mg/l | 0.00001 | < | < | 0.00004 | 0.00003 |
| Al | mg/l | 0.01 | 0.06 | 0.03 | 0.16 | 0.85 |
| As | mg/l | 0.0001 | 0.0012 | 0.0009 | 0.1034 | 0.111 |
| B | mg/l | 0.01 | 0.02 | 0.01 | < | < |
| Ba | mg/l | 0.00005 | 0.14777 | 0.07402 | 0.05102 | 0.04673 |
| Be | mg/l | 0.0001 | < | < | 0.0004 | 0.0089 |
| Ca | mg/l | 0.01 | 10.72 | 16.74 | 10.28 | 26.27 |
| Cd | mg/l | 0.00002 | < | 0.00004 | 0.00091 | 0.00238 |
| Cl | mg/l | 2 | 29 | 28 | 166 | 121 |
| Co | mg/l | 0.0001 | 0.0002 | 0.0018 | 0.2536 | 0.0760 |
| Cr | mg/l | 0.01 | < | < | < | < |
| Cu | mg/l | 0.01 | $<$ | $<$ | 1.00 | 0.25 |
| F | mg/l | 0.1 | 1.6 | 1.2 | 0.1 | 0.4 |
| Fe | mg/l | 0.01 | 0.01 | < | 0.48 | 1.16 |
| Hg | mg/l | 0.0001 | < | < | 0.0003 | 0.0003 |
| K | mg/l | 0.1 | 7.4 | 10.5 | 10.4 | 9.9 |
| Mg | mg/l | 0.01 | 6.34 | 8.76 | 10.00 | 12.62 |
| Mn | mg/l | 0.01 | 0.02 | 0.14 | 0.39 | 0.43 |
| Mo | mg/l | 0.00005 | 0.00408 | 0.00151 | 0.00126 | 0.00025 |
| Na | mg/l | 0.1 | 26.2 | 29.5 | 31.3 | 28.6 |
| Ni | mg/l | 0.01 | < | < | 0.07 | 0.13 |
| P | mg/l | 0.1 | < | < | 0.2 | 0.4 |
| Pb | mg/l | 0.0005 | < | < | 0.0044 | 0.0525 |
| $\mathrm{SO}_{4}$ | mg/l | 0.3 | 17.2 | 74.1 | 31.5 | 65.5 |
| Sb | mg/l | 0.00001 | 0.00027 | 0.00032 | 0.00053 | 0.00020 |
| Se | mg/l | 0.0005 | 0.004 | 0.0056 | 0.4113 | 0.4466 |
| Si | mg/l | 0.05 | 1.62 | 1.98 | 1.97 | 3.52 |
| Sn | mg/l | 0.0001 | < | < | 0.0022 | 0.0003 |
| Sr | mg/l | 0.00002 | 0.061 | 0.0977 | 0.06838 | 0.16459 |
| Th | mg/l | 0.000005 | 0.000037 | 0.000027 | 0.000472 | 0.000561 |
| U | mg/l | 0.000005 | 0.00011 | 0.000085 | 0.000228 | 0.000670 |
| Zn | mg/l | 0.01 | < | < | 0.52 | 0.32 |

< element at or below analytical detection limit.


Figure 1: Acid-base account (ABA) plot showing ANC versus total S for overburden and floor samples.


Figure 2: Geochemical classification plot for overburden and floor samples showing NAGpH versus NAPP, with ARD classification domains indicated.


Figure 3: Box plot showing distribution of MPA and ANC values for open cut and underground samples. 10th, 25th, 50th (median), 75th and 90 th percentiles are marked.


Figure 4: ABCC profile for sandstone sample 28523 with an ANC value of $94 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. Carbonate standard curves are included for reference.


Figure 5: ABCC profile for sandstone sample 30418 with an ANC value of $82 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. Carbonate standard curves are included for reference.


Figure 6: ABCC profile for siltstone sample 28533 with an ANC value of $37 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, and shale sample 30424 with an ANC value of $47 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. Carbonate standard curves are included for reference.


Figure 7: ABCC profile for sandstone sample 30411 with an ANC value of $24 \mathrm{~kg} \mathrm{H} 2 \mathrm{SO} 4 / \mathrm{t}$. Carbonate standard curves are included for reference.


Figure 8: ABCC profile for sandstone sample 28517 with an ANC value of $12 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$, and shale sample 30405 with an ANC value of $14 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. Carbonate standard curves are included for reference.


Figure 9: Acid-base account (ABA) plot showing ANC versus total S for coal and reject samples.


Figure 10: Geochemical classification plot for coal and reject samples showing NAGpH versus NAPP, with ARD classification domains indicated.


Figure 11: Kinetic NAG graph for sample 30373.


Figure 12: ABCC profile for sandstone sample 30425 with an ANC value of $11 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$. Carbonate standard curves are included for reference.

## APPENDIX A

## Assessment of Acid Forming Characteristics

## Assessment of Acid Forming Characteristics

## Introduction

Acid rock drainage (ARD) is produced by the exposure of sulphide minerals such as pyrite to atmospheric oxygen and water. The ability to identify in advance any mine materials that could potentially produce ARD is essential for timely implementation of mine waste management strategies.

A number of procedures have been developed to help assess the acid forming characteristics of mine waste materials. The most widely used assessment methods for ARD characterisation are the Acid-Base Account (ABA) and the Net Acid Generation (NAG) test. These methods are referred to as static procedures because each involves a single measurement in time.

## Acid-Base Account

The acid-base account involves static laboratory procedures that evaluate the balance between acid generation processes (oxidation of sulphide minerals) and acid neutralising processes (dissolution of alkaline carbonates, displacement of exchangeable bases, and weathering of silicates).

The values arising from the acid-base account are referred to as the maximum potential acidity (MPA) and the acid neutralising capacity (ANC), respectively. The difference between the MPA and ANC value is referred to as the net acid producing potential (NAPP).

The chemical and theoretical basis of the ABA are discussed below.

## Maximum Potential Acidity

The MPA that can be generated by a sample is determined from the sample sulphur content. The total sulphur content of a sample is commonly determined by the Leco high temperature combustion method. The calculation assumes that all the sulphur measured in the sample occurs as pyrite $\left(\mathrm{FeS}_{2}\right)$ and that the pyrite reacts under oxidising conditions to generate acid according to the reaction:

$$
\mathrm{FeS}_{2}+15 / 4 \mathrm{O}_{2}+7 / 2 \mathrm{H}_{2} \mathrm{O} \Rightarrow \mathrm{Fe}(\mathrm{OH})_{3}+2 \mathrm{H}_{2} \mathrm{SO}_{4}
$$

According to this reaction, the MPA of a sample containing $1 \% \mathrm{~S}$ as pyrite would be 30.6 kilograms of $\mathrm{H}_{2} \mathrm{SO}_{4}$ per tonne of material (i.e. $\mathrm{kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ ). Hence the MPA of a sample is calculated from the total sulphur content using the following formula:

$$
\text { MPA }\left(\mathrm{kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)=(\text { Total } \% \mathrm{~S}) \times 30.6
$$

The use of the total sulphur assay to estimate the MPA is a conservative approach because some sulphur may occur in forms other than pyrite. Sulphate-sulphur and native sulphur, for example, are non-acid generating sulphur forms. Also, some sulphur may occur as other metal sulphides (e.g. covellite, chalcocite, sphalerite, galena) which yield less acidity than pyrite when oxidised or, in some cases, may be non-acid generating.

The total sulfur content is commonly used to assess MPA because of the difficulty and costs involved in routinely determining the speciation of sulfur forms within samples and determining reactive sulphide-sulfur contents. However, if the sulphide mineral forms are known then allowance can be made for non- and lesser acid generating sulfur forms to provide a better estimate of the MPA.

## Acid Neutralising Capacity

The acid formed from pyrite oxidation will to some extent react with acid neutralising minerals contained within the sample. This inherent acid buffering is quantified in terms of the ANC.

The ANC is commonly determined by the Modified Sobek method. This method involves the addition of a known amount of standardised hydrochloric acid $(\mathrm{HCl})$ to an accurately weighed sample, allowing the sample time to react (with heating), then back-titrating the mixture with standardised sodium hydroxide $(\mathrm{NaOH})$ to determine the amount of unreacted HCl . The amount of acid consumed by reaction with the sample is then calculated and expressed in the same units as the MPA, that is $\mathrm{kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$.

## Net Acid Producing Potential

This is a theoretical calculation commonly used to indicate if a material has potential to produce acidic drainage. It represents the balance between the capacity of a sample to generate acid (MPA) and its capacity to neutralise acid (ANC). The NAPP is also expressed in units of $\mathrm{kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$ and is calculated as follows:

$$
\text { NAPP = MPA }-\mathrm{ANC}
$$

If the MPA is less than the ANC then the NAPP is negative, which indicates that the sample may have sufficient ANC to prevent acid generation. Conversely, if the MPA exceeds the ANC then the NAPP is positive, which indicates that the material may be acid generating.

## ANC/MPA Ratio

The ANC/MPA ratio is frequently used as a means of assessing the risk of acid generation from mine waste materials. The ANC/MPA ratio is another way of looking at the acid base account. A positive NAPP is equivalent to an ANC/MPA ratio less than 1, and a
negative NAPP is equivalent to an ANC/MPA ratio greater than 1. A NAPP of zero is equivalent to an ANC/MPA ratio of 1 .

The purpose of the ANC/MPA ratio is to provide an indication of the relative margin of safety (or lack thereof) within a material. Various ANC/MPA values are reported in the literature for indicating safe values for prevention of acid generation. These values typically range from 1 to 3 . As a general rule, a ANC/MPA ratio of 2 or more generally signifies that there is a high probability that the material will remain circum-neutral in pH and thereby should not be problematic with respect to acid rock drainage.

## Acid-Base Account Plot

Sulphur and ANC data are often presented graphically in a format similar to that shown in Figure 1. This figure includes a line indicating the division between NAPP positive samples from NAPP negative samples. Also shown are lines corresponding to ANC/MPA ratios of 2 and 3 .


Figure A-1. Acid-base account (ABA) plot

## Net Acid Generation (NAG) Test

The NAG test is used in association with the NAPP to classify the acid generating potential of a sample. The NAG test involves reaction of a sample with hydrogen peroxide to rapidly oxidise any sulphide minerals contained within a sample. During the NAG test both acid generation and acid neutralisation reactions can occur simultaneously. Therefore, the end result represents a direct measurement of the net amount of acid generated by the sample. This value is commonly referred to as the NAG capacity and is expressed in the same units as NAPP, that is $\mathrm{kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$.

Several variations of the NAG test have been developed to accommodate the wide geochemical variability of mine waste materials. The three main NAG test procedures currently used by EGi are the single addition NAG test, the sequential NAG test, and the kinetic NAG test.

## Single Addition NAG Test

The single addition NAG test involves the addition of 250 mL of $15 \%$ hydrogen peroxide to 2.5 gm of sample. The peroxide is allowed to react with the sample overnight and the following day the sample is gently heated to accelerate the oxidation of any remaining sulphides, then vigorously boiled for several minutes to decompose residual peroxide. When cool, the pH and acidity of the NAG liquor are measured. The acidity of the liquor is then used to estimate the net amount of acidity produced per unit weight of sample.

An indication of the form of the acidity is provided by initially titrating the NAG liquor to pH 4.5 , then continuing the titration up to pH 7 . The titration value at pH 4.5 includes acidity due to free acid (i.e. $\mathrm{H}_{2} \mathrm{SO}_{4}$ ) as well as soluble iron and aluminium. The titration value at pH 7 also includes metallic ions that precipitate as hydroxides at pHs between 4.5 and 7.

## Sequential NAG Test

When testing samples with high sulphide contents it is not uncommon for oxidation to be incomplete in the single addition NAG test. This can sometimes occur when there is catalytic breakdown of the hydrogen peroxide before it has had a chance to oxidise all of the sulphides in a sample. To overcome this limitation, a multi-stage sequential NAG test is often carried out. This test may also be used to assess the relative geochemical lag of PAF samples with high ANC.

The sequential NAG test is a multi-stage procedure involving a series of single addition NAG tests on the one sample (i.e. 2.5 g of sample is reacted two or more times with 250 mL aliquots of $15 \%$ hydrogen peroxide). At the end of each stage, the sample is filtered and the solution is used for measurement of NAGpH and NAG capacity. The NAG test is then repeated on the solid residue. The cycle is repeated until such time that there is no further catalytic decomposition of the peroxide, or when the NAGpH is greater than pH 4.5. The overall NAG capacity of the sample is then determined by summing the individual acid capacities from each stage.

## Kinetic NAG Test

The kinetic NAG test is the same as the single addition NAG test except that the temperature, pH and sometimes EC of the liquor are recorded. Variations in these parameters during the test provide an indication of the kinetics of sulphide oxidation and acid generation during the test. This, in turn, can provide an insight into the behaviour of the material field under field conditions. For example, the pH trend gives an estimate of
relative reactivity and may be related to prediction of lag times and oxidation rates similar to those measured in leach columns. Also, sulphidic samples commonly produce a temperature excursion during the NAG test due to the decomposition of the peroxide solution, catalysed by sulphide surfaces and/or oxidation products.

## Sample Classification

The acid forming potential of a sample is classified on the basis of the acid-base and NAG test results into one of the following categories:

- Barren,
- Non-acid forming (NAF),
- Potentially acid forming (PAF), and
- Uncertain (UC).


## Barren

A sample classified as barren essentially has no acid generating capacity and no acid buffering capacity. This category is most likely to apply to highly weathered materials. In essence, it represents an 'inert' material with respect to acid generation. The criteria used to classify a sample as barren may vary between sites, but for hard rock mines it generally applies to materials with a total sulfur content $\leq 0.1 \%$ S and an ANC $\leq 5 \mathrm{~kg} \mathrm{H}_{2} \mathrm{SO}_{4} / \mathrm{t}$.

## Non-acid forming (NAF)

A sample classified as NAF may, or may not, have a significant sulfur content but the availability of ANC within the sample is more than adequate to neutralise all the acid that theoretically could be produced by any contained sulphide minerals. As such, material classified as NAF is considered unlikely to be a source of acidic drainage. A sample is usually defined as NAF when it has a negative NAPP and the final NAG $\mathrm{pH} \geq 4.5$.

## Potentially acid forming (PAF)

A sample classified as PAF always has a significant sulfur content, the acid generating potential of which exceeds the inherent acid neutralising capacity of the material. This means there is a high risk that such a material, even if pH circum-neutral when freshly mined or processed, could oxidise and generate acidic drainage if exposed to atmospheric conditions. A sample is usually defined as PAF when it has a positive NAPP and a final NAGpH $<4.5$.

## Uncertain (UC)

An uncertain classification is used when there is an apparent conflict between the NAPP and NAG results (i.e. when the NAPP is positive and $\mathrm{NAGpH}>4.5$, or when the NAPP is
negative and $\mathrm{NAGpH} \leq 4.5$ ). Uncertain samples are generally given a tentative classification that is shown in brackets e.g. UC(NAF).

Figure A-2 shows the format of the classification plot that is typically used for presentation of geochemical data. Marked on this plot are the quadrats representing the NAF, PAF and UC classifications.


Figure A-2 Geochemical classification plot

## Other Methods

Other test procedures may be used to define the acid forming characteristics of a sample.
pH and Electrical Conductivity
The pH and electrical conductivity (EC) of a sample is determined by equilibrating the sample in deionised water for a minimum of 1 hour, typically at a solid to water ratio of $1: 2(\mathrm{w} / \mathrm{w})$. This gives an indication of the inherent acidity and salinity of the waste material when initially exposed in a waste emplacement area.

Acid Buffering Characteristic Curve (ABCC) Test
The ABCC test involves slow titration of a sample with acid while continuously monitoring pH . This data provides an indication of the portion of ANC within a sample that is readily available for acid neutralisation.


[^0]:    ${ }^{1}$ Bowen, H.J.M. (1979) Environmental Chemistry of the Elements. Academic Press, New York, p 36-37.

[^1]:    MPA $=$ Maximum Potential Acidity $\left(\mathrm{kgH}_{2} \mathrm{SO}_{4} / \mathrm{t}\right)$

[^2]:    < element at or below analytical detection limit.

[^3]:    < element at or below analytical detection limit.

