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**Assessment Report on the Missy Drilling & Mapping
as well as
the Sampling of the Missy and Magnum Properties**

Laird Mining Division
British Columbia, Canada

March 14, 2008

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by

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BC GEOL SURVEY BRANCH
ASSESSMENT REPORT
29756

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1.0 Introduction

This Assessment Report outlines drilling and other work carried out in 2007 on the Missy Property (the "Claim"), tenure number 501534 and the Angel claim (the Churchill mine) tenure number 501416 which are part of the group of 580 mineral claims comprising the Trident Copper Project.

At the request of Aries Resource Corp and Action Minerals Inc. (the "Companies" or "Action" or Aries"), the accompanying assessment report was prepared on the Trident Copper Project properties (the "Property"), Fort Nelson Area, Laird Mining Division, British Columbia Canada to summarize previous work, appraise the exploration potential of the Property, and to make recommendations for future work. The trident Copper Project comprises a group of 580 un-surveyed mineral claims totalling over 223,595 hectares (ha).

2.0 Descriptions, Locations and Ownership of Claims

The Trident Copper Project comprises a group of 580 contiguous mineral claims totalling 223,595 hectares (ha). The claims are located in the Laird Mining Division, British Columbia, Canada, and is shown on Map Sheets NTS 94K/4, 5, 6, 11, and 12.

The Property area is centered at latitude 58°23' North, longitude 125°24' West, and UTM 6476000 m North, and UTM 360000 m East. Detailed claim information is provided in Appendix A.

Aries holds an interest in claims through option agreements with seven arms-length parties: Twenty-Seven Capital Corp., GWN Investment Ltd., Saints Investment Ltd. Laird Rice, Ryan Gibson, Seguro Projects Inc and Doctors Investment Group Ltd. Action has acquired an interest in the Missy, Okey, Sox, and Talus claims through a non arm's length agreement with Aries. Action also holds an interest in claims through option agreements with six arms-length parties: Minero Majestuoso Limitado, GWN Investment Ltd., Saints Investment Ltd. Laird Rice, Ryan Gibson, and Doctors Investment Group Ltd. The following is a summary of the Trident Copper project acquisitions:

Property	Location	Nature of Ownership Claim Numbers	Current Use or Operations Conducted on the Property	Financial Terms Related to the Company's Ownership of its Interest in the Properties
Neil, Talus, Sox Joint Venture Property	Liard Mining Division, British Columbia	50% 504054 501462 510008	exploration	<p>The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided fifty (50%) per cent interest in the Mineral Claims by making the following payments/commitments (the "Option Payments") to the Owner:</p> <p>the issuance of 500,000 common shares and a cash payment of \$50,000 to be paid within 10 days of exchange approval;</p> <p>a cash payment of \$75,000 on or before 180 days of exchange approval;</p> <p>Before the first (1st) anniversary of this Agreement 500,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$400,000 on the Mineral Claims or any properties forming part of the Mineral Claims (including any properties acquired with borders within thirty kilometres of the nearest portion of the Mineral Claims ["Proximate Properties"]), subject to Aries having previously received a National Instrument 43-101 compliant property report recommending such work;</p> <p>On the second (2nd) anniversary of this Agreement 500,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$1,100,000 on the Mineral Claims or any properties forming part of the Mineral Claims (including any properties acquired with borders within thirty kilometres of the nearest portion of the Mineral Claims ["Proximate Properties"]), subject to Aries having previously received a National Instrument 43-101 compliant property report recommending such work; and</p> <p>On the third (3rd) anniversary of this Agreement 1,000,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$1,500,000 on the Mineral Claims or any properties forming part of the Mineral Claims (including any properties acquired with borders within thirty kilometres of the nearest portion of the Mineral Claims ["Proximate Properties"]), subject to Aries having previously received a National Instrument 43-101 compliant property report recommending such work.</p> <p>Aries shall have the right at any time to accelerate the Option Payments for the purpose of shortening the time period for exercising the Option.</p>
Missy Property	Liard Mining Division, British Columbia	50% 501534	exploration	<p>(i) the issuance of 500,000 common shares and a cash payment of \$100,000 to be paid within 10 days of exchange approval;</p> <p>(ii) On or after the first (1st) anniversary of this Agreement 500,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$400,000 on the Mineral Claim or any properties forming part of the Mineral Claim (including any properties acquired with borders within thirty kilometres of the nearest portion of the Mineral Claim ["Proximate Properties"]),</p> <p>(iii) On the second (2nd) anniversary of this Agreement 1,000,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$400,000 on the Mineral Claim or any properties forming part of the Mineral Claim (including any properties acquired with borders within thirty kilometres of the nearest portion of the Mineral Claim ["Proximate Properties"]),</p> <p>(iv) On the third (3rd) anniversary of this Agreement 1,000,000 common shares shall be issued to the Owner and, by such time, Action shall have performed exploration and development work costing \$400,000 on the Mineral Claim or any properties forming part of the Mineral Claim (including any properties acquired with borders within thirty kilometres</p>

				of the nearest portion of the Mineral Claim ["Proximate Properties"]).
Yedhe Mountain Property	Liard Mining Division, British Columbia	100% 519444 519445 519446 519447 519448 519449 519450 519451 519452 519453 519454 519455 519456 519457 519458	exploration	The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner: A cash payment of \$20,000 and 400,000 Common shares to be paid and issued within 30 days of TSX Venture Exchange approval. A 1% NSR shall be reserved unto the Owner hereunder which may be purchased at any time by Action paying to the Owner \$1,000,000, less all amounts previously received by Owner as NSR payments.
Nelson Property	Liard Mining Division, British Columbia	100% 520701 520702 520703 520704 520707	exploration	a) The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner: (v) A cash payment of \$10,000, and (vi) 500,000 Common shares shall be issued to the Owner no later than 10-business days after the receipt of regulatory approval to this Agreement. b) A 1% NSR shall be reserved unto the Owner hereunder which may be purchased at any time by Action paying to the Owner \$1,000,000, less all amounts previously received by Owner as NSR payments.
Goliath Property	Liard Mining Division, British Columbia	100% 529843 529844 529845 529846 529847 529848 529849 529850 529851	exploration	a) The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner: (vii) A cash payment of \$20,000, and (viii) 600,000 Common shares shall be issued to the Owner no later than 10-business days after the receipt of regulatory approval to this Agreement. b) A 1% NSR shall be reserved unto the Owner hereunder which may be purchased at any time by Action paying to the Owner \$1,000,000, less all amounts previously received by Owner as NSR payments.
Tusk	Liard	100%	exploration	a) The Owner hereby grants Action an exclusive and irrevocable

Property	Mining Division, British Columbia	537943 537945 537947 537948 537950 537951 537952 537953 537954 537955		<p>option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner:</p> <p>(ix) 2,000,000 Common shares shall be issued to the Owner no later than 10 days after exchange acceptance,</p> <p>(x) A cash consideration of \$25,000 upon exchange acceptance.</p>
Peace River Property	Liard Mining Division, British Columbia	100% 537944 538056 538054 538053 538050 538047 538052 538066 538064 538063 538061 538058 538057 538048 538045 537941 538069 538078 538083 538088 538090 538093 538095 538098 538076 538075 538072 538071 538055 538038 538036 538081 538080 538067 538065 538062 538060 538070 538073 538084	exploration	<p>a) The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner:</p> <p>(xi) A cash payment of \$20,000, and</p> <p>(xii) 4,000,000 Common shares shall be issued to the Owner no later than 10-business days after the receipt of regulatory approval to this Agreement.</p> <p>b) A 1% NSR shall be reserved unto the Owner hereunder which may be purchased at any time by Action paying to the Owner \$1,000,000, less all amounts previously received by Owner as NSR payments.</p>

		538091 538085 538092 538097 538087 538089 538099 538096 538082 538079 538077		
Summit Property	Liard Mining Division, British Columbia	100% 517930 517932 517931 517929 517928 517927 517926 517925 517924 517878 517877 517875 517882 517893 517879 517891 517890 517888 517886 517885 517892 517894 517895 517898 517899 517900	exploration	a) The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner: (xiii) 2,000,000 Common shares shall be issued to the Owner within 10 days of TSX Venture Exchange acceptance, (xiv) A cash consideration of \$25,000 within 10 days of TSX Venture Exchange acceptance.
Racing River Property	Liard Mining Division, British Columbia	50% (claim numbers attached as schedule "B")	exploration	a) The Optionor hereby grants the Optionees an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments and performing the following work programs (collectively the "Option Payments") to the Optionor: (i) No later than 2 (two) business days after signing of the agreement, a cash deposit of \$150,000 (one hundred and fifty thousand dollars) shall be paid to the Optionor. The deposit shall be refundable to the Optionees in the event that this agreement, in this or any amended form, is not accepted for filing with the TSX Venture Exchange; (ii) No later than 180 days after the receipt of regulatory approval of this Agreement, an additional cash payment

				<p>of \$300,000 (three hundred thousand dollars) shall be paid to the Optionor;</p> <p>(iii) No later than 270 days after the receipt of regulatory approval of this Agreement, an additional cash payment of \$300,000 (three hundred thousand dollars) shall be paid to the Optionor;</p> <p>(iv) On or before the third (3rd) anniversary of regulatory approval of this Agreement, the Optionees shall have performed an aggregate amount of \$5,000,000 (five million dollars) in exploration work (the "exploration commitment") on the Mineral Claims, or on any claims within the Trident Copper Project area;</p> <p>(v) On the third (3rd) anniversary date, and with the Optionees having successfully performed the exploration commitment described in 2.01a.(iv), the Optionor shall, at its sole discretion, through the process of giving written notice to the Optionees, receive one of the following:</p> <ul style="list-style-type: none"> i. 2,500,000 common shares in Action Minerals Inc. and 2,500,000 common shares in Aries Resource Corp., to be issued within 5 business days of having given the aforementioned written notice. In the event that the Optionees vend, joint venture or otherwise dispose of the Mineral Claims to an "area partner" during the term of this agreement, the Optionor shall have the right to receive its aggregate 5,000,000 shares from each of the area partners on a pro-rata basis based upon claim area, or ii. \$5,000,000 in cash, to be paid within 90 days of having given the aforementioned written notice. <p>b) A 1% NSR shall be reserved unto the Optionor hereunder.</p> <p>c) The Optionees shall have the right at any time to accelerate the Option Payments for the purpose of shortening the time period for exercising the Option.</p> <p>d) The Optionor shall be solely responsible for making all obligations, payments and keeping current the Klassen Agreement, as defined in Schedule "C" hereto, on behalf and to the benefit of the Optionees.</p>
Rush Property	Liard Mining Division, British Columbia	100% 534724 534725 534726	exploration	<p>a) The Owner hereby grants Action an exclusive and irrevocable option (the "Option") to acquire an undivided one hundred (100%) per cent interest in the Mineral Claims by making the following payments (the "Option Payments") to the Owner:</p> <ul style="list-style-type: none"> (xv) A cash payment of \$20,000, and (xvi) 2,500,000 Common shares shall be issued to the

				<p>Owner no later than 10-business days after the receipt of regulatory approval to this Agreement.</p> <p>b) A 1% NSR shall be reserved unto the Owner hereunder which may be purchased at any time by Action paying to the Owner \$1,000,000, less all amounts previously received by Owner as NSR payments.</p>
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Aries Resource Corp and Seguro Projects Inc:

Key Property and Okey Claim

This option agreement (Agreement) between Aries Resource Corp, 1255 West Pender Street, Vancouver, B.C. (Aries), and Seguro Projects Inc, 330 East 23rd Street, North Vancouver, B.C. (Seguro), includes the Key Property and the Okey claim and is effectively dated December 14, 2004. The Agreement is subject to approval, which has been obtained, of the TSX Venture Exchange of both this Agreement and the agreement between Seguro and Senator Minerals Inc, 418 East 14th Street, North Vancouver, B.C. (Senator), cancelling the option agreement held by Senator to acquire a 50% interest in the Key Property and the Okey claim. The following table details Aries' payments under the Agreement.

Monetary Payments CAN\$	
To be paid within 2 days of TSX Venture Exchange Agreement approval	\$10,000
To be paid within 30 days of TSX Venture Exchange Agreement approval	\$32,500
To be paid within 60 days of TSX Venture Exchange Agreement approval	\$32,500
To be paid within 6 months of TSX Venture Exchange Agreement approval	\$75,000
Total	\$150,000
Payments of Aries Common Stock	
To be issued within 10 days of TSX Venture Exchange Agreement approval	300,000 shares
To be issued within 10 days of receipt of independent report of first work program or no later than 30 Nov 2005.	300,000 shares
To be issued upon commencement of commercial production	500,000 shares
Total	1,100,000 shares

The Agreement gives Aries an option to control 100% of the properties, net of a 3% Net Smelter Return Royalty (NSR). Commencing with the date of the Agreement and continuing until the date of commercial production, Aries is to pay a retainer for consulting and operating activities to Seguro, in the amount of CAN\$12,000, by the end of the first month in each quarter.

For the duration of the Agreement, Aries has the right to designate an Operator entitled to charge an Operator fee equal to 9% of Exploration and Development Expenditures. In the event that Seguro is the designated Operator, 50% of Seguro's retainer fee will be applied as a payment toward the total Operator fee.

Under the Agreement, Aries must keep the claims in good standing and ensure that all exploration work is carried out by qualified parties paid at industry standard rates.

**Seguro Projects Inc, Donald A. Simon, and Doctors Investment Group Ltd:
NBC Copper Properties Acquisition Agreement**

Donald A. Simon, 330 East 23rd Street, North Vancouver, B.C. (Simon), registered with the British Columbia Ministry of Energy and Mines, Mineral Titles branch, as Free Miner Certificate #124708, holds title on behalf of Seguro to the following ten mineral claims with Tenure Numbers 501389, 501321, 501416, 501446, 501462, 501482, 501497, 501523, 501534, and 510811 (Simon Claims).

The acquisition agreement (Agreement) between Doctors Investment Group Ltd , 29 Retirement Road, PO Box N-7777, Nassau, Bahamas (Doctors) and Seguro includes the Simon Claims and is effectively dated January 5, 2005. The Agreement between Doctors and Seguro allows Doctors to acquire an undivided 100% interest in the Simon Claims, net of a 1% Net Smelter Return Royalty (NSR), for the following considerations:

- Upon confirmation of the value of any of the Simon Claims through the acceptance by any recognized stock exchange of any option agreement by a listed company to earn an interest in any of the claims, Doctors will pay to Simon \$1,000 for each claim so approved;
- If work is commenced on any of the Simon Claims, Seguro is to be retained as the operator, and if circumstances preclude Seguro from being the operator, Doctors will retain Seguro on a consulting basis at industry standard rates; and
- If any claim is dropped by Doctors or any optionee, Seguro will be notified thirty (30) days in advance, and Seguro will be allowed first right of ownership of said claim or partial claim at no cost to Seguro.

All Simon Claims are registered in the name of Simon, who acts as registered claimholder only. Upon written request and providing that all above considerations have been met, Simon will provide Doctors and Seguro with executed registerable transfers of interests in the claims.

Doctors and Seguro may assign rights and obligations without the prior written consent of the other party. Any assignee chosen by Doctors must assume all Agreement obligations, and Doctors retains any liabilities and obligations occurring prior to such assignment.

Doctors may terminate the Agreement at any time upon written notice to Seguro thirty (30) days prior to the termination date. Upon termination, Seguro is entitled to retain all payments made by Doctors to the date of termination, and, at Seguro's option, is entitled to beneficial ownership of all terminated claims.

**Gilbert Santos and Doctors Investment Group Ltd:
NBC Copper Properties Acquisition Agreement**

Gilbert Santos, 2795 East 18th Avenue, Vancouver, B.C. (Santos), registered with the British Columbia Ministry of Energy and Mines, Mineral Titles branch, as Free Miner Certificate #146887, holds title to twelve mineral claims with Tenure Numbers 504049, 504054, 504060, 504064, 504085, 509540, 509544, 509549, 509553, 509563, 509567, and 509576 (Santos Claims).

The acquisition agreement (Agreement) between Doctors and Santos includes the Santos Claims and is effectively dated January 5, 2005. The Agreement allows Doctors to acquire an undivided 100% interest, net of a 1% Net Smelter Return Royalty (NSR), in the Santos Claims for the following considerations:

- Upon confirmation of the value of any of the Santos Claims through the acceptance by any recognized stock exchange of any option agreement by a listed company to earn an interest in any of the claims, Doctors will pay to Santos \$1,000 for each claim so approved;
- If work is commenced on any of the Santos Claims, Santos is to be retained as operator, and if circumstances preclude Santos from being the operator, Doctors will retain Santos on a consulting basis; and

- If any claim is dropped by Doctors or any optionee, Santos will be notified within thirty (30) days, and Santos will be allowed first right of ownership of said claim or partial claim at no cost to Santos.

**Aries Resource Corp and Seguro Projects Inc:
Churchill Property Option Agreement**

This option agreement (Agreement) includes the Cisco and Angel claims and is effectively dated February 24, 2005.

The Agreement is subject to approval of the TSX Venture Exchange. The Agreement gives Aries an option to control 100% of the claims, net of a 1% Net Smelter Return Royalty (NSR). The following table details Aries' payments under the Agreement.

Timing	Payment	Aries Work Requirement
To be issued within 10 business days of TSX Venture Exchange Agreement approval	500,000 shares	none
To be issued on the 1 st anniversary of the Agreement	1,000,000 shares	\$250,000 of NI 43-101 recommended work
To be issued on the 2 nd anniversary of the Agreement	2,500,000 shares	\$500,000 of NI 43-101 recommended work
To be issued on the 5 th anniversary of the Agreement	5,000,000 shares	\$500,000 and bankable feasibility study recommending production
Total	9,000,000 shares	CAN\$1,250,000

Share issuance requirements are subject of additional regulatory and shareholder approvals, as might be required from time to time, in the event that the share issuances will result in the creation of new insiders or control positions.

Seguro's 1% NSR can be purchased by Aries at any time for CAN\$1,000,000, less any prepaid NSR amounts. At any time, Aries may accelerate the Option Payments,

shortening the time period for exercising the Agreement. If Aries fails to make any of the payments, Aries will not be entitled to a partial interest in the claims.

Aries may install, maintain, replace, and remove any machinery, equipment, tools, and facilities on the claims. Upon termination of the Agreement, Aries has a period of six (6) months in which to remove its equipment at its sole expense.

During the Agreement period, Aries shall at all times occupy, manage, and use the subject claims in full compliance with all environmental laws. Aries will be responsible for prompt performance of any reclamation, remediation, or pollution control required for its operations carried out during the Agreement term.

There is an area of interest (AOI) extending one (1) mile from the outer boundaries of the claims. The AOI applies to any additional properties acquired by Seguro, and Aries may acquire a 100% interest in the AOI properties without additional consideration. AOI properties will be included in the Agreement upon Aries reimbursing Seguro for reasonable acquisition costs.

Aries may terminate the Agreement at any time upon written notice to Seguro thirty (30) days prior to the termination date. Upon termination, Seguro is entitled to retain all payments made by Aries to such date. If Aries fails to duly pay or cure any obligation default within thirty (30) days after receipt of a default notice from Seguro, Seguro may terminate the Agreement.

**Doctors Investment Group Ltd and Aries Resource Corp:
Liard Property Option Agreement**

This option agreement (Agreement) effectively dated May 16, 2005, grants Aries an option to acquire up to an undivided 100% interest in the following twenty claims with the Tenure Numbers, 504049, 504054, 504060, 504064, 504085, 509540, 509544, 509549,

509553, 509563, 509567, 509576, 510811, 501321, 501446, 501462, 501482, 501497, 501523, and 501534.

The Agreement gives Aries a yearly option to control 100% of the claims, net of a 2% Net Smelter Return Royalty (NSR). The following table details Aries' payments under the Agreement.

Timing	Payment	Work Requirement
To be issued within 10 business days of TSX Venture Exchange Agreement approval	2,000,000 shares (100,000/claim)	none
To be issued on the 1 st anniversary of the Agreement	2,000,000 shares	\$750,000 of NI 43-101 recommended work
To be issued on the 2 nd anniversary of the Agreement	2,500,000 shares	\$750,000 of NI 43-101 recommended work
To be issued on the 3 rd anniversary of the Agreement	5,000,000 shares	\$1,000,000 of NI 43-101 recommended work
To be issued on the 4 th anniversary of the Agreement	5,000,000 shares	\$1,000,000 of NI 43-101 recommended work
Total	16,500,000 shares	CAN\$3,500,000

Share issuance requirements are subject of additional regulatory and shareholder approvals, as might be required from time to time, in the event that the share issuances will result in the creation of new insiders or control positions.

Doctors' 2% NSR may be purchased by Aries at any time for CAN\$2,000,000, less any prepaid NSR amounts. At any time, Aries may accelerate the Option Payments shortening the time period for exercising the Agreement. If Aries fails to make any of the payments, Aries will not be entitled to a partial interest in the claims. If a bankable feasibility study is prepared in favour of the claims, either before or after exercising the Agreement, Aries will issue an additional 5,000,000 common shares to Doctors within five (5) working days of receipt of share issuance regulatory approval.

Concurrently with each of the aforementioned Common Share issuances, Doctors will execute a Voting Trust document which will allow Aries' current management or their

assigns to vote such Common Shares as they deem fit. The Voting Trust does not restrict Doctors from selling Common Shares to unrelated third parties from time to time as it sees fit.

Aries Resource Corp and Action Minerals Inc:

Neil Property Option Agreement

The non-arm's length option agreement (Agreement) between Aries and Action Minerals Inc, 1255 West Pender Street, Vancouver, B.C. (Action), effectively dated July 11, 2005 and amended August 10, 2005, includes the Okey (TN: 510008), Sox (TN: 501462), and the Talus (TN: 504054) claims. The Agreement grants Action an exclusive and irrevocable option to acquire an undivided 50% interest in the Okey, Sox, and Talus claims. The following table details Action's payments.

Timing	Payment	Action Work Requirements
To be issued within 10 business days of TSX Venture Exchange Agreement approval	500,000 common shares CAN\$50,000 cash payment	none
On or before 180 days of TSX Venture Exchange Agreement approval	CAN\$75,000 cash payment	none
To be issued before the 1 st anniversary of the Agreement	500,000 common shares	\$400,000 of NI 43-101 recommended work
To be issued on the 2 nd anniversary of the Agreement	500,000 common shares	\$1,100,000 of NI 43-101 recommended work
To be issued on the 3 rd anniversary of the Agreement	1,000,000 common shares	\$1,500,000 of NI 43-101 recommended work
Total	2,500,000 common shares CAN\$125,000	CAN\$3,000,000

Exploration and development work by Action may be carried out on the subject claims as well as on acquired properties having borders within thirty (30) kilometres of the nearest portion of the subject claims.

Share issuance requirements are subject of additional regulatory and shareholder approvals, as might be required from time to time, in the event that the share issuances will result in the creation of new insiders or control positions.

At any time, Action may accelerate the Option Payments shortening the time period for exercising the Agreement.

3.0 Accessibility, Climate and Physiography

Access to the Trident Copper Project area is by helicopter from Fort Nelson. Helicopter access can also be based from Toad River (Mile 422 Alaska Highway) or Muncho Lake (Mile 462 Alaska Highway), where hotel accommodations are available. Ground access to the north-eastern portion of the Trident area is possible by two-track dirt road extending thirty kilometres from a point approximately thirteen kilometres west of Summit Lake (Mile 401 Alaska Highway) to the Churchill mill site situated at the confluence of Delano Creek and the Racing River. A temporary exploration camp was located at the Churchill mill site. The road is in good condition and well used, but entails fording MacDonald Creek, Wokkpash Creek, and Delano Creek/Racing River.

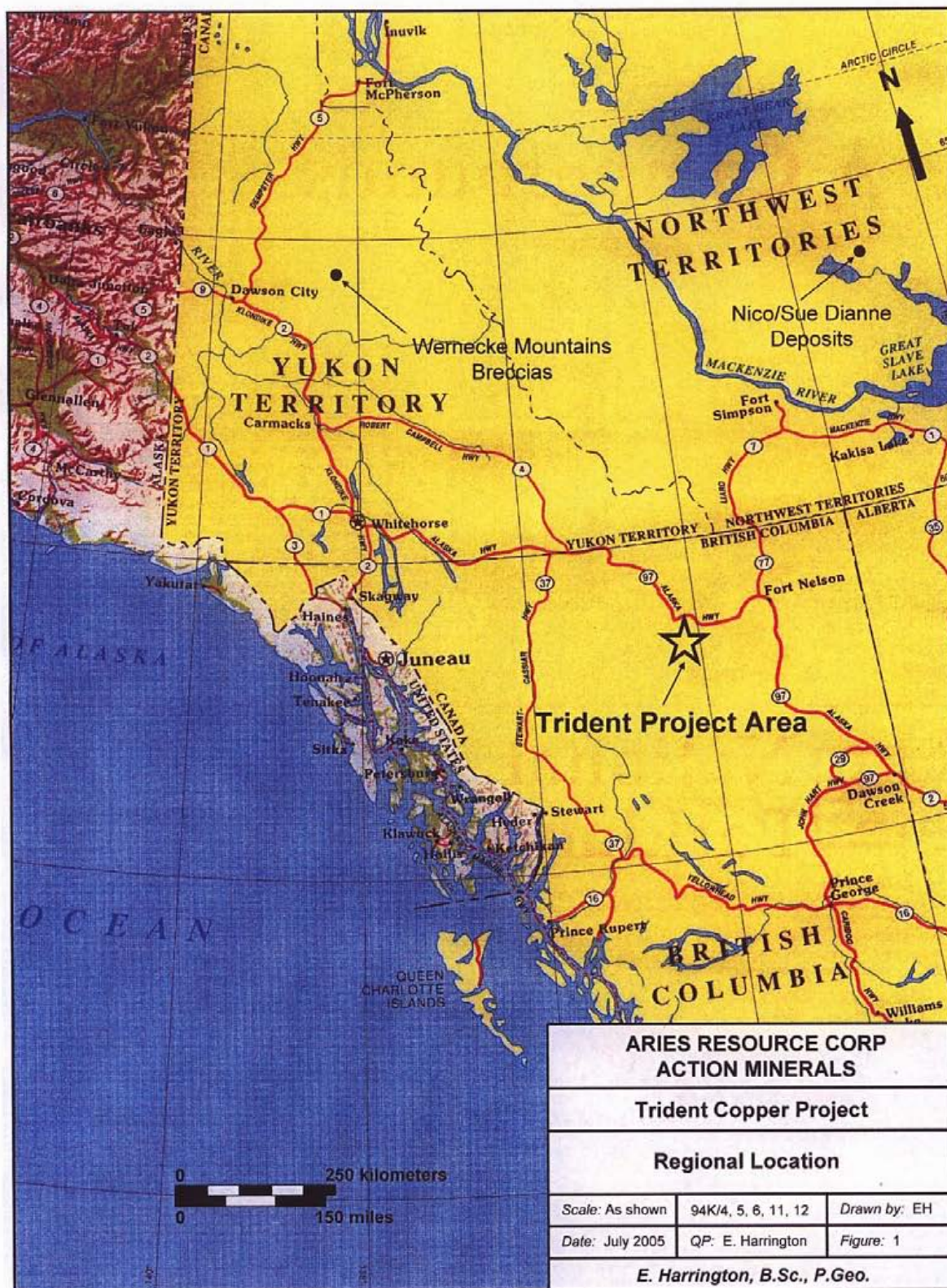
Access to the north-western portion of the Trident Copper Project area is by road from Mile 442 on the Alaska Highway, where a dirt road leads south along the Toad River and Yedhe Creek for approximately 30 kilometres to the area of the Key property. The bridge located 1.5 kilometres south of the Alaska Highway, where the Toad River road crosses the Toad River, has a resurfaced width only allowing motorized quad bikes or smaller vehicles. The roads along the Toad River, Yedhe Creek, and the turnoff into the Key property are subject to periodic washouts.

The project area is on moderate to very steep mountainous glaciated terrain with elevations ranging from 1,100 and 2,680 meters. Except for creek and river valleys showing coniferous tree growth, most of the claims are above the tree-line where vegetation is restricted to shrubs and grasses, or is nonexistent. Climate is variable, with higher elevations receiving

precipitation almost daily during the summer. Winters are cold, with snow that stays from September to May. The work season is mid- or late-June to mid-September.

Rocks in the Trident Copper Project area are predominantly Proterozoic Helikian-age Aida Formation marine sediments consisting of calcareous and dolomitic mudstone, siltstone, and minor sandstone. Upper and lower Aida Formation contacts are conformable. The overlying Gataga Formation consists of mudstone, siltstone, and sandstone, and the underlying Tuchodi Formation consists of quartzite, dolomite, siltstone, and red shale. There are a number of other marine sediments occurring within the project area ranging in age from Cambrian to Silurian. While known copper deposits in the project area are vein-type, trace element results from 2005 rock sampling suggest that iron-oxide copper gold (IOCG) mineralization, similar to the polymetallic Olympic Dam deposit in Australia and the Nico deposit in the Northwest Territories, may be present.

Figure 1: Regional Location of the Trident Project



4.0 History

4.1 Area History

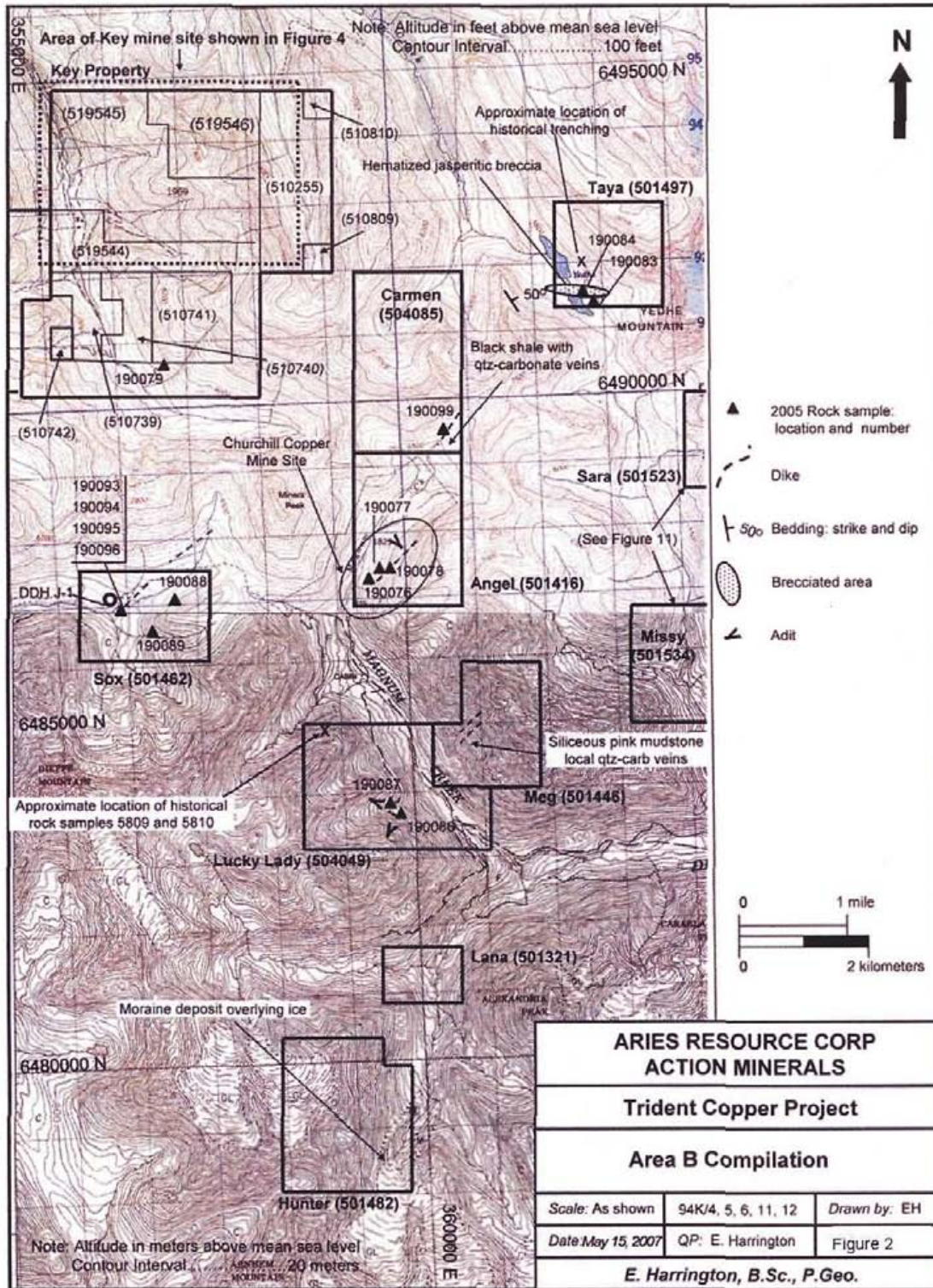
During the 1940s, copper was discovered in the area while the Alaska Highway was being built. Exploration activity took place during the 1950s and early 1960s, but was most active during the late 1960s and early 1970s. The two main deposits identified were the Davis-Keays (the Eagle Vein located on the Key property), discovered in August 1967, by prospectors Harris Davis and Robert Keays of Fort Nelson, BC, and the Churchill Copper deposit (the Magnum Vein located on Aries' Angel claim).

4.2 Previous Work

4.2.1 Missy Claim TN: 501534

As no assessment reports are listed for previous work on the Missy Claim (Figure 2), historical information is limited to Minfile Master Report 094K 005 of the Geological Survey Branch, Ministry of Energy & Mines. The historical Bill copper showing lies close to a thrust fault within the Muskwa Assemblage's Aida Formation. The Bill showing is located on Aries' current Missy claim and consists of four copper-bearing quartz carbonate veins, striking 020 degrees, in dolostone and carbonaceous shale (Figure3). The veins, each about one meter thick, are adjacent to a small shear in the footwall of the thrust, and are generally poorly and sporadically mineralized with chalcopyrite.

Figure 2: Missy and Angel Claim Locations

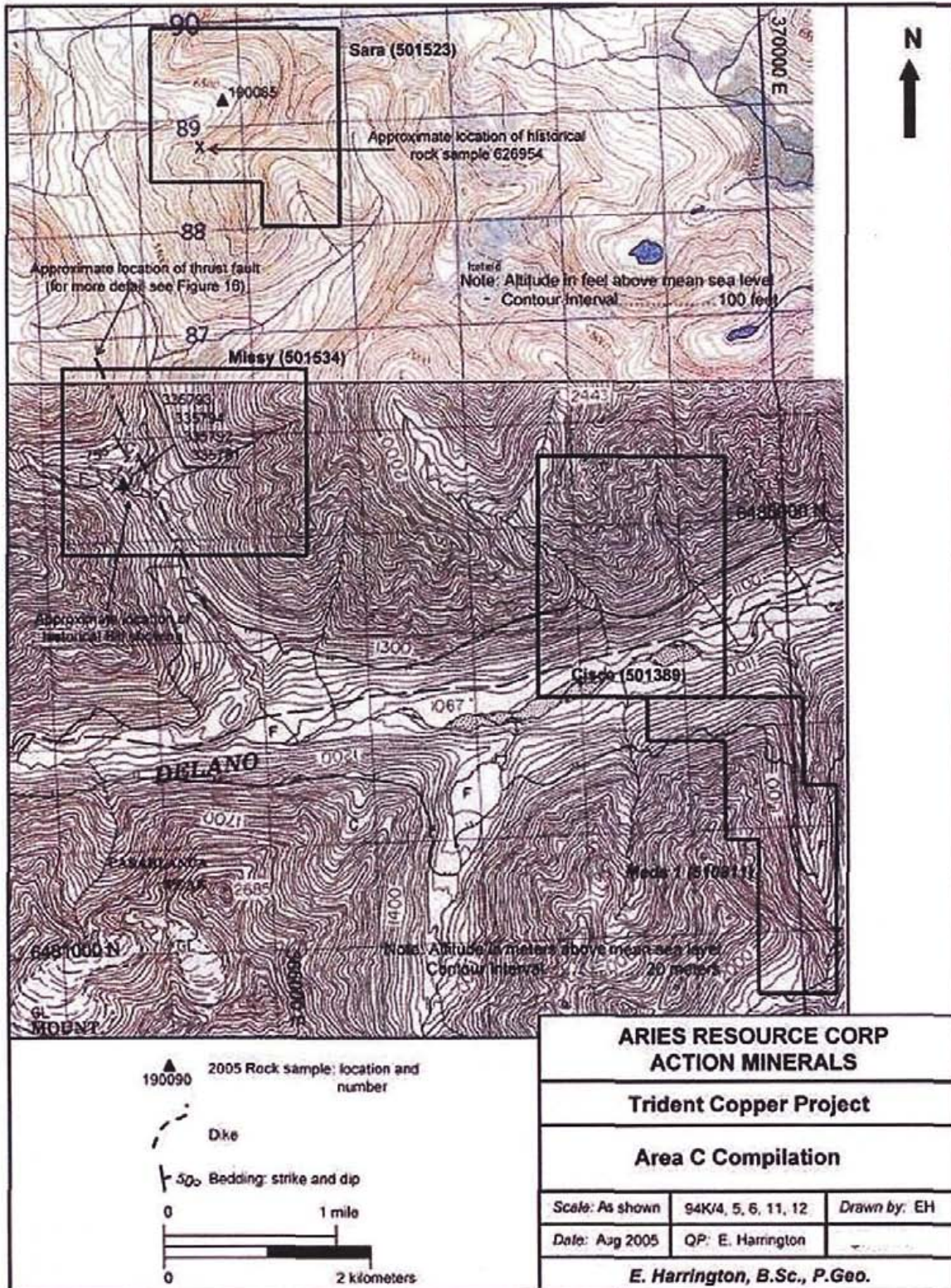


On the Missy claim, four rock chip samples were taken from quartz-carbonate veins associated with mafic dykes (Harrington 2005). Quartz-carbonate veining contained malachite staining, massive, and disseminated chalcopyrite. All samples returned anomalous copper values. Sample 335791 was anomalous in silver while the other three samples returned elevated silver values.

Table 1: Missy Rock Sampling (2005)

Sample	Type	Width m	Au g/t	Ag g/t	Ba ppm	Ce ppm	Co ppm	Cu %	La ppm	P ppm
335791	chip	1	<.001	1.08	30	16.1	1.7	0.85%	7.4	90
335792	chip	2	0.012	0.35	20	16.8	24.9	0.48%	6.7	240
335793	chip	1.5	0.005	0.42	30	30.5	4.5	0.61%	12.9	210
335794	chip	0.5	0.013	0.61	30	24.1	4.1	1.54%	10.1	320

Figure 3: Missy Sample Locations (2005)



4.2.2 Angel Claim TN: 501416 (Magnum Vein; Churchill Copper Mine)

The Magnum vein was discovered in 1943. In 1958 and 1959, Canex Aerial Exploration Ltd carried out a work program of rock sampling and diamond drilling for Magnum Consolidated Mining Company Ltd. (Figure 2)

Mineralization, described as being epigenetic hydrothermal vein-type, consists of chalcopyrite, bornite, and malachite, with gangue of pyrite, quartz, carbonate, graphite, and ankerite. The deposit occurs in Aida Formation sediments consisting of calcareous shale, dolomite, and limestone, cut by a large number of northeast- to east-trending diabase dikes (Figure 4). Copper mineralization occurs in quartz-carbonate veins.

The diabase dykes and quartz-carbonate veining are generally parallel but dikes are post-mineralization, truncating the veins. A series of northwest-trending trachytic composition dikes cuts across mineralized veins.

Host rock Aida Formation deformation and northwest-trending folding (regionally forming the Muskwa Anticlinorium) are pre-mineralization. At Magnum Creek, dykes, fracture zones, and mineralized veins all cut across the regional folding structure suggesting that both the dykes and veins were emplaced in fracture system that developed after regional folding. The northeast-trending and steeply dipping fracture system and mineralized veining at Magnum Creek was explored for a length of 1,375 meters, 90 meters wide, and to a depth of 365 meters. Veins range from less than 1 meter up to 7.6 meters, and ten veins have been identified.

While the reserve calculation reported for Churchill Copper Corporation Limited (Churchill) by Chapman, Wood, and Griswold (feasibility report, 1969 (as reported by Glenn (1991)) is considered relevant, it is historical, and does not meet NI 43-101 standards. Aries is not treating the reserve calculation as a NI 43-101 defined resource or reserve verified by the writer. The writer has not verified assay results or the resource calculation. Aries has not done the work necessary to verify the classification of the resource or reserve. Aries is not treating this historical amount and classification as a NI 43-101-compliant defined resource or reserve as a qualified person has not verified the figures. Therefore, the historical estimate should not be relied upon. No estimates have been made since that date. In addition, the mineral resource cannot be converted to mineral reserves without further drilling and engineering studies.

From 1967 to 1969, Churchill conducted drilling at 100-foot centers and some crosscutting and raising on the Magnum vein. Prior to production, Churchill reported proven and probable reserves totaling 1.178 million tons grading 3.92% copper, including a 20% dilution factor, were delineated. From 1970-1974, the Churchill mine processed 598,000 tons of copper ore grading 3.0% copper (Harrington E, 16 August 2005).

5.0 Regional Geology

(Taken from Chapman et al, 1971)

“The Missy property lies within the eastern edge of the Rocky Mountains in an area of rugged topography. Excellent exposures exist above timberline revealing flat to locally contorted sedimentary rock formations dislocated by extensive regional faulting.

Proterozoic argillites, quartzites, and limestones contain all the known copper deposits, possess generally low dips, are intruded by post-ore diabase dykes of Proterozoic age, and are overlain by un-mineralized Palaeozoic formations of Cambrian and later ages. The Proterozoic strata occupy nearly the full width (40-50 miles) of the Rocky Mountains in the south part of the area. Northward they become separated into a north-trending eastern belt (mainly east of upper MacDonald Creek) and wider central and western belts which trend northwest and reach the Alaska Highway west of about Mile 436.

The presently known quartz-carbonate veins, many of which contain chalcopyrite, occur mainly in the western half of the Precambrian with a more or less similar distribution to the subsequent diabase dykes.

The dykes cut the veins and are themselves only weakly mineralized on fractures containing carbonates (principally calcite) and quartz. In places dykes are more strongly mineralized by barren pyrite.

Veins may be much less numerous than dykes, many of which are discernible at a distance on the hill slopes. Dykes and veins generally have more or less similar attitudes, which are relatively constant in certain zones, belts, or parts of the area. Dykes and veins

probably occur in, and may be virtually restricted to, these so-called mineral belts.

The best recognized to date is belt 1 approximately 6 miles wide and 40 miles long that trends north 35 degrees west and contains, from north to south, the known copper deposits of the Davis-Keays, Magnum, John, Lady, Churchill Creek, Ed, and Anne properties, (Figure 4; block 2) Most of the known mineralized veins of the region have strikingly similar mineral composition and structural characteristics. The Missy property is located on the border of block 2 and 3. The dykes/ veins trend in a south- west direction in block 1 until covered by over-thrusted younger rocks. The dykes/ veins trend in a northeast direction on block 2 for about 4 km then they are covered by younger rocks.

This belt, which is further marked by a pattern of sporadically developed northwest trending asymmetric folds with steep east limbs and by the occurrence within it of a huge local pile of Cambrian conglomerate that forms Mt. Roosevelt, contains dykes and veins that mostly strike east of north and possess steep westerly dips.

6.0 Regional Structure

Middle Proterozoic sediments of the Muskwa Assemblage (Wheeler et al, 1991) include the Tetsa, George, Henry Creek, Tuchodi, Aida, and Gataga formations described by Taylor et al, 1973.

The Muskwa Assemblage is cut by gabbroic dykes and is overlain unconformably by Cambrian (Atan Group) and Ordovician (Kechika Group) rocks. These Ordovician and older rocks, termed pseudo-basement by Taylor, were intensely and repeatedly deformed during pre-Laramide periods of tectonism, and also later during the Laramide Orogony, which occurred between 89 and 43 Ma. Laramide compression deformation created large asymmetrical northwest-trending folds, thrust faults, and anticlinal structures which form the Muskwa Anticlinorium.

Uplift in the Rocky Mountains resulted principally from generally northeast-southwest shortening and thrust faulting that penetrated basement rocks, bringing the basement

and overriding younger strata to relatively high levels in the crust. The Laramide thrusts likely followed older zones of weakness.

A fracture zone of normal faults, later than Laramide deformation, extends southward from Muncho Lake into the Toad River valley. The normal faults have a vertical displacement of up to 2,000 feet (600 meters).

Figure 4: Regional Geology of the Trident Project

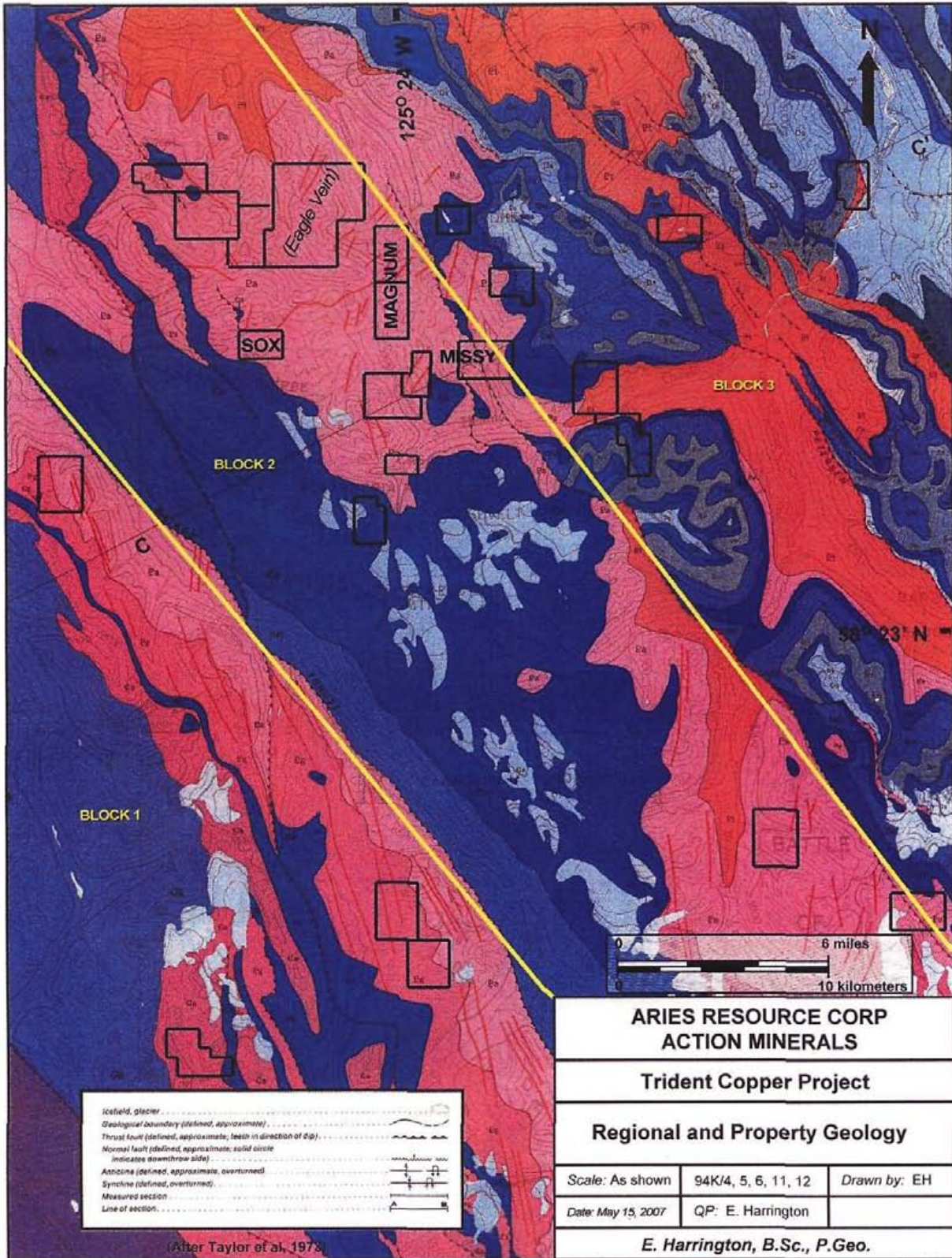


Table 2: Geology Legend

Phanerozoic	Paleozoic	
	Carboniferous and Devonian	
	Db	- Besa River Formation: dark pyritic siliceous shale
	Devonian	
	Dd	- Dunedin Formation: dark grey limestone
	<i>Local Disconformity</i>	
	Ds	- Stone Formation: light grey dolomite; dolomite breccia
	<i>Disconformity</i>	
	Dw	- Wokkpash Formation: sandstone, minor dolomite, shale
	Dm	- Muncho-McConnell Formation: dolomite
	<i>Disconformity</i>	
	Silurian	
	Sn	- Nonda Formation: dark grey dolomite, basal sandstones; minor limestone
	<i>Angular unconformity</i>	
	Ordovician - Ketchica Group	
Ok	- argillaceous limestone	
Okg	- graptolitic shale	
Okt	- turbidites	
Okl	- limestone, minor sandstone	
<i>Angular unconformity</i>		
Cambrian - Atan Group		
Ca	- limestone, dolomite; minor sandstone and shale	
Cs	- conglomerate, sandstone, shale; minor limestone	
<i>Disconformity</i>		
Proterozoic	Hadrynian	
	Pv	- quartz-chlorite phyllite, meta-sandstone, quartz-pebble conglomerate
	<i>Angular unconformity</i>	
	Helikian	
		- gabbroic dykes
	Pg	- Gataga Formation: mudstone, siltstone; minor sandstone
	Pa	- Aida Formation: mudstone, siltstone; minor chamositic and carbonaceous mudstone, dolomite, and limestone
	Pt	- Tuchodi Formation: quartzite, dolomite, siltstone; minor red shale
	Ph	- Henry Creek Formation: calcareous mudstone, siltstone; minor sandstone
	Pd	- George Formation: limestone, dolomite
Ps	- Tetsa Formation: dark grey mudstone, sandstone; minor quartzite	
<i>Disconformity</i>		
Pc	- Chisma Formation: dolomite, quartzite; minor siltstone	

7.0 General Mineralization Types

General mineralization types discussed in this report are:

Mineralization Type 1; Chalcopyrite bornite pyrite quartz - carbonate veins

This is the most pervasive epithermal vein type of mineralization, encountered at the Magnum, Eagle, Toro, Neil, Missy and Sox projects. The near vertical vein mineralization and associated dykes, crosscut shallow dipping, folded and thrust faulted well foliated argillites and limestone.

Mineralization Vein type 1A; Brecciated mineralized (veined) zones

The brecciated mineralized zones occur where mineralized quartz carbonate vein feeders are intersected by faulting and thus the host rocks were fractured. The mineralization is then trapped in the open cavities. At the Churchill mine, a breccia zone, 20 to 30 metre thick over 200m in length, is exposed on surface just north of the exploited veins. Malachite (on surface) is clearly visible within the Churchill brecciated zone. The brecciated zone occurs in close vicinity to a north- west trending fault zone. The northern extension of the Neil vein displays a similar brecciated mineralized structure after being cut, by a north west striking fault zone. These brecciated (veined) zones normally carrier relatively large tonnages and high to low copper grades (as was mined on the Toro property). The Neil breccia assayed at 6.1 % copper over 20.8m.

8.0 Regional Geophysical Surveys (2006)

In April 2006 Action and Aries retained McPhar Geosurveys Ltd. to perform ~2600 line kilometres (~1600 miles) of helicopter supported magnetic surveys (MAG), to be flown at a line spacing of 100m over a large portion of the Trident Property, including the Missy Property. The goal of the surveys was to locate mafic dykes spatially associated with the mineralized quartz veins, such as the Magnum Eagle and Missy veins and to identify prospective mineralized bodies, such as Olympic Dam-type IOCG (Iron-oxide/Copper/Gold/Silver/Cobalt) mafic intrusive bodies. In addition, some 820 line kilometres (~500 miles) of frequency electromagnetic surveys (EM) were to be flown over areas known to contain large veins with conductive massive sulphides to determine their geophysical signatures. For increased accuracy, surveys were

conducted at low levels (~30m above ground). By fall, inclement weather and the rugged topography forced the replacement of McPhar with Aeroquest Ltd. which completed the expanded surveys. In total, ~1800 line kilometres of MAG/EM and ~2600 line kilometres of MAG were flown in 2006. The airborne magnetic surveys were successful in mapping the diabase dykes swarms on the Missy property as well as several large buried magnetic intrusive bodies. Significant EM and MAG anomalies were noted at the Churchill Mine, at and above the Keys mine, at the Missy and Goat Matnik. The MAG was successful at delineating basic geological structure at the Missy. A high elevation magnetic survey was flown at about the highest mountain height in 2005 for Archer Cathro associates. The magnetic data was acquired from Archer Cathro Associates.

8.1 Regional Magnetic Survey Results (2007)

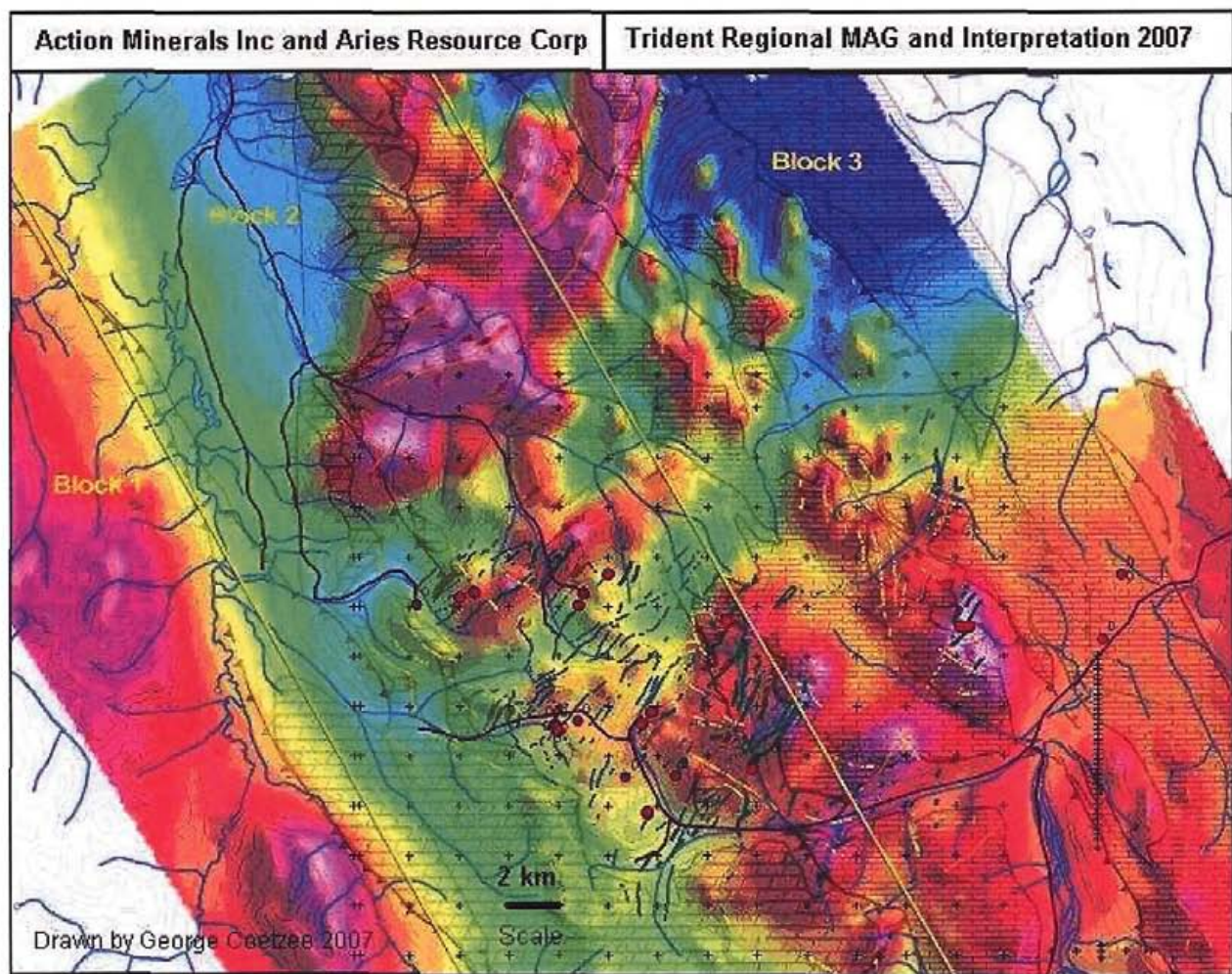
The low level helicopter MAG (for Action minerals Inc) and high level fix wing magnetic survey (for Archer Cathro Associates) were interpreted separately and then combined. Different colors and line thicknesses were utilized for different anomaly intensities strengths as well as directional trends to highlight different dyke trends and faults. Stratigraphy bedding directions and structural lows were also delineated.

The low level helicopter magnetic survey confirms the fact that 95% of the dykes cuts right through the thrust faults without any displacement. This indicates that the folding and thrust faulting are older than > 780 million years (current dating age of the dykes); this confirms that the dykes were emplaced in the fracture system that developed during and after regional folding, thrust faulting and foliation.

The major dykes swarm trends were divided into three main structural zones or blocks. The boundaries of these blocks are controlled by the major thrust faults. The stress pattern for each block was controlled by the different compression and extensional direction forces (pulses) during different geological time periods. This pertains to over thrusting (towards the east) as well as lateral movement on these thrust faults (Left lateral movement?). Normal faults are nearly none existent on the project. (Figure 4).

In block 1 the major dykes trends at about 160° (west of Toad River. [True north = 0°]). In the central area of block 2, the major dykes trend at 35° (Missy and Churchill Mines). The minor dykes trend at 115° and 95° (in the vicinity of Churchill area). In the southern part of block 2, (13km south of Churchill mill site near Toro mine) the directions of dykes are generally about 110° . In block 3 the major trend is 165° , the minor trend is $15-30^{\circ}$ (Figure 5).

Figure 5: High and Low Level MAG Surveys with Structural Blocks Outline



Drawn by George Coetzee, BSc Honours, 2007

9.0 Exploration on the Missy

9.1 The Geology of the Missy Veins

The Missy property is located proximately 4 km southwest of the Churchill mine with known historical resources (Figure 2). The mineralization mirrors the Churchill and Davis Keays mines chalcopyrite veining within structural shear/fault zones paralleling mafic dykes; Genn David; 1991. The outcrops consist mainly of buff grey weathered slatey argillites and calcareous shales of the Aida Formation. The argillite are foliated and folded in places.

The Missy Veins consist of three distinct mineralized vein structures, bearing chalcopyrite and malachite minor bornite and containing anomalous gold and silver values. These veins outcrop only on the north-eastern steep slope of a creek over a 30m strike length within the argillite. The veins are located near the northwest contact of a green-grey medium crystalline mafic dyke that is only exposed on the south-western bank of the creek, for the most part the veins parallels the dyke with an approximately 035 degree strike.

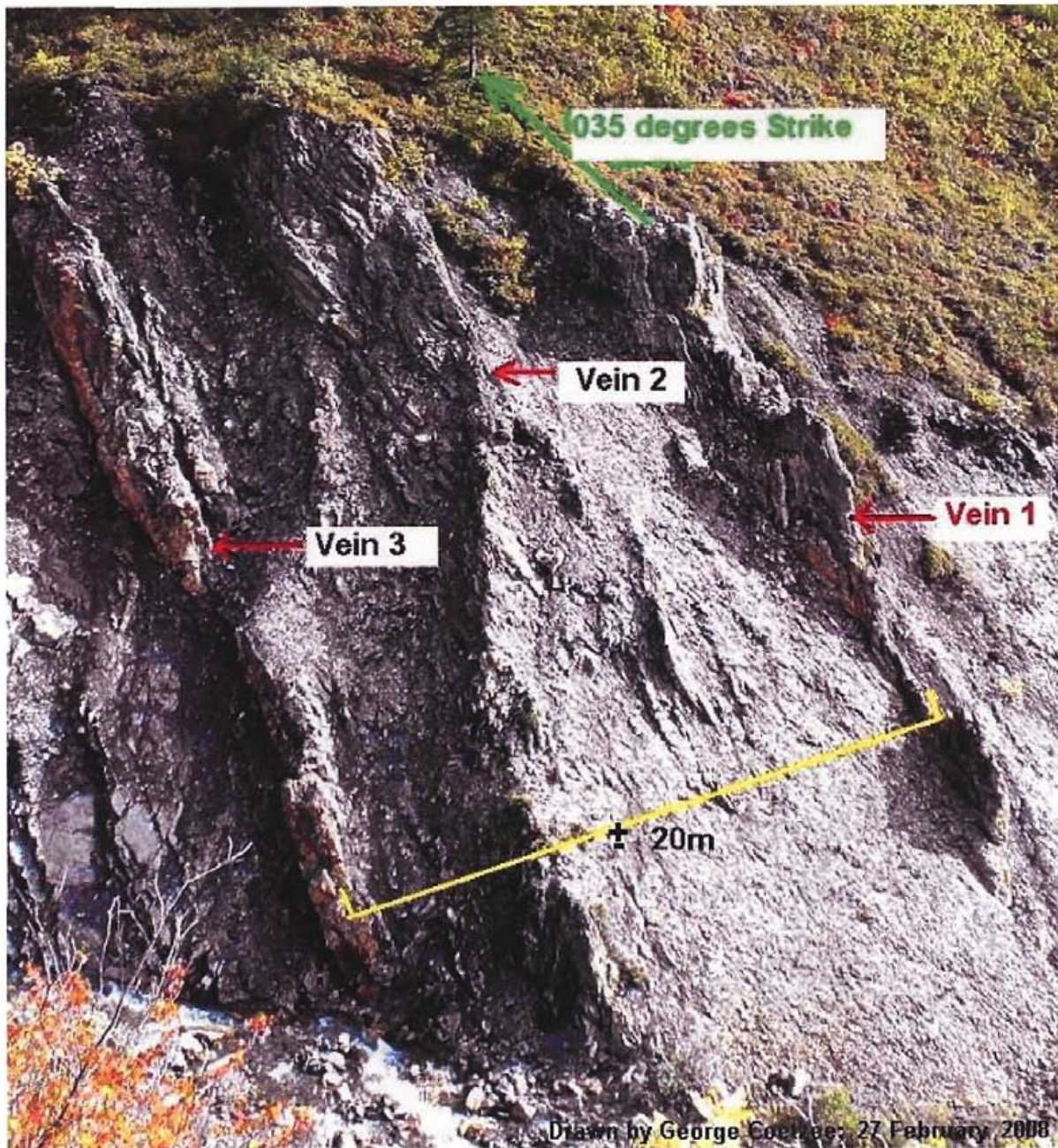
Halfway up the south-western creek slope malachite calcite vein float was sited next to the dyke indicating the mineralization continue towards the southwest. The northeast extensions of the mineralized veins are covered by thick unconsolidated glacial overburden.

The Missy surface mineralization, consists of three approximately 0.5 – 1.5 meter carbonate quartz veins, that contain malachite and lesser azurite veining. The oxidization of the mineralization is generally only pervasive up to 10 to 20 m below surface. The two main semi parallel veins named; vein #1 (southeast side) and vein #3 (northwest side) is spaced approximately 20 meters apart. A horse tail vein #2, originating from vein 1, is positioned about 7 m northwest of vein #1. (Figure 6) The three veins dip vertically or steeply towards the southeast. The argillite host rock contacts are silicified for approximately 0.5 m on both sides of vein #1.

contacts are silicified for approximately 0.5 m on both sides of vein #1.

The below surface the mineralization consists mainly of chalcopyrite with minor pyrite, cobalt, bornite and gold within carbonate - quartz vein material that intruded parallel shear/fault zones in the argillite.

Figure 6: Missy Veins on Surface



Drawn by George Coetzee, 27 February 2008

Drawn by George Coetzee, BSc Honours, 27 February 2008

The three Missy veins were sampled on surface was sampled in September 2008⁷ by David Peake BSc. Geo. under supervision of George Coetzee BSc. Geo Honours and J. Kowalchuk P.Geo. Continuous channel samples were chipped across the mineralized vein structures at five metre intervals along the three veins. (Figure 6) The samples were bagged, securely stored and transported as prescribed by best practice sampling procedures. Seventeen samples were analyzed by Acme Analytical Labs of Vancouver, BC, with multi-element Inductively Coupled Plasma Mass Spectrometer (ICP MS) technique. For the 40 element geochemical analysis as well as for samples returning >10,000ppm values (maximum detection limits), the Group 7TX analytical procedure employing Hot 4-Acid "near total" digestion was used, followed ICP-MS analysis. Values greater than 10,000 ppm copper were re-assayed with a further dilution of the solution to give a more accurate analysis of higher grade samples. Follow up gold fire assays were performed on relevant samples.

The average grade of Vein #1, over a strike length of 30 metres and an average width of 1.0 metres was 4.6% copper.

The average grade of Vein #2 over a strike length of 25 metres and an average width of 0.5 metres was 1.97% copper.

The average grade of Vein #3 over a strike length of 30 metres and an average width of 1.0 metres was 1.14% Cu. The chip sample results are shown as follows:

Table 3: Missy Vein Channel Chip Samples

Sample Number	Type	Width m	Vein #	% Copper
465201	Chip	1	Vein 1	4.6
465202	Chip	1	Vein 1	0.98
465203	Chip	1	Vein 1	16.10
465204	Chip	1	Vein 1	4.52
465205	Chip	1	Vein 1	0.91
465206	Chip	1	Vein 1	0.52
465207	Chip	0.5	Vein 2	0.68
465208	Chip	0.5	Vein 2	3.86
465209	Chip	0.5	Vein 2	0.55
465210	Chip	0.5	Vein 2	1.15
465211	Chip	0.5	Vein 2	3.60
465212	Chip	1	Vein 3	0.38
465213	Chip	1	Vein 3	3.69
465214	Chip	1	Vein 3	0.48
465215	Chip	1	Vein 3	0.13
465216	Chip	1	Vein 3	0.54
465217	Chip	1	Vein 3	1.60

9.3 Geophysical Exploration on the Missy Property

Prior to drilling, mapping, surface a VLF electromagnetic and a magnetic survey were completed over a strike length of about 400m to pinpoint the extended strike length and position of veins under the soil cover. The surface survey was inconclusive in accurately outlining the dyke position and therefore not included in this report.

A VLF electromagnetic instrument was utilized to pinpoint the extended strike length and positions of veins under the soil cover. The VLF electromagnetic survey was carried out using an EM16 unit manufacture by Geonics Limited of Metropolitan Toronto, Ontario. This unit – a sensitive receiver with two orthogonal coils, one axis normally vertical and the other horizontal - makes use of the VLF transmitting stations operating for communication with submarines for its transmitted signal – the vertical antenna currents creates concentric horizontal magnetic fields – and measures the vertical components of the secondary fields created as above.

The signal from the vertical axis coil is first minimized by tilting the instrument – tilt angle calibrated in percentage- and the remaining signal in the coil is finally balanced out by the measured percentage of a signal from the other coil, after being shifted 90 degrees. Thus if the secondary fields are small compared to the primary horizontal field, the mechanical tilt angle is an accurate measure of the vertical real component, and the compensation signal from the horizontal coil is one of the quadrature vertical signal. In all 1.5 kilometres of traverses were done using the above instruments at the station intervals of 5 metres (or 1m near veins) using mainly transmitters of Seattle- NLK 24.8 khz. and Hawaii – NPM 21.4 khz. The field instructions as to how to orientate the instrument during the survey were strictly followed.

The VLF survey lines (every 25m) extended outward from the veins by at least 15 m to 50m depending on the terrain and as to ensure no possible veining could be missed. The location where the VLF instrument emitted the highest pitch signal was marked with surveyor lint. A two man team was predominately used to double check these VLF vein readings.

The VLF signal strengths on both ends of the projected mineralization extensions indicated that the veins extend in both directions. The veins could be more than 1000 metres in length based on the geology and magnetic survey of the regional area. The VLF survey returned weak to moderate signal strengths. The vein positions intersected in the drill holes thus far do not strongly correlate with VLF survey positions. Therefore usefulness of the VLF survey in locating the veins under thick overburden is still to be verified by further drilling. Mineralized float indicate that the veins also extend towards the south west. Based on the geology of the Missy area the veins could be potentially be mineralized for > 1000m in length as was the case at the nearby Churchill mine. For at least 1.8 km, overburden covers the north-east extension of the Missy veins. Future EM work and follow up drilling are required in order to test the mineral potential of the Missy veins under the overburden.

9.4 Mineralization in the Missy Drill Holes

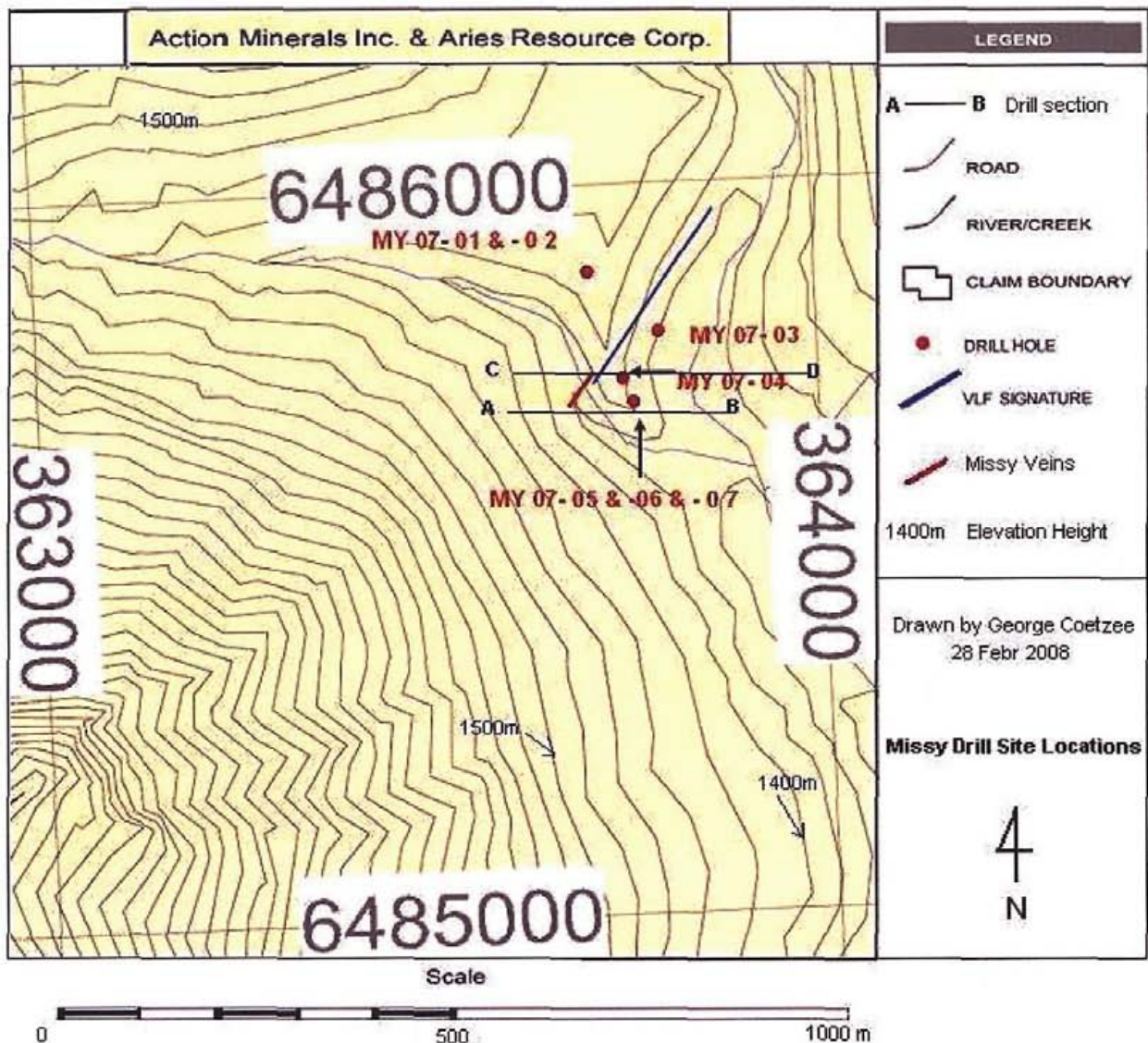
The mineralization consists mainly of chalcopyrite veining and pyrite within quartz-carbonate material that intruded parallel shear zones within and on the margin of the dyke. The chalcopyrite also occurs as patches and disseminations generally in close proximity to the chalcopyrite veins, and predominately in located in a calcite, ankerite and quartz matrix. The white carbonate is generally of a coarse crystalline nature.

No sulphide oxidation (malachite and/or azurite) and or erythrite (hydrated cobalt arsenide) were observed in any of the drill holes.

9.5 Missy Drilling Program

The helicopter-supported drill program was designed to test the down-dip extent of the three Missy veins, which are exposed on the northeast bank of a creek. Diamond drilling at the Missy Prospect was performed in February and between August - October, 2007, and was contracted to Simpson Drilling Ltd. from Stewart, BC. Approximately 274 m of drilling was completed. The drill hole positions and drill hole azimuths were surveyed with a Rhino handheld GPS (5 - 20m accuracy) and a compass. (Figure 7) The drill holes are for the most part less than 62 metres in length. The directional deflections of the short drill holes were negligible.

Figure 7: Missy Diamond Drill Site Locations



http://webmap.em.gov.bc.ca/mapplace/minpot/ex_assist.cfm (MTO Tenure Number 501534)

9.6 Missy Drill Hole Results

Drill hole MY07-01 was drilled at -45° in an east-southeast direction directly towards the three veins. The hole was abandoned in thick glacial till overburden. The final depth of the hole was 37.5 metres (Table 4).

MY07-02 was drilled at -60° in a south-eastern direction. The hole was abandoned in loose till material at a depth of 32 metres.

MY07-03 was drilled in a northwest direction at a dip of -60° . The hole was positioned where the creek cuts into the unconsolidated till overburden. Once again the hole was abandoned at 25.3 metres in the till overburden.

MY07-04 was drilled at -78° in a westerly direction, east of vein #3. It is debatable if the thin carbonate vein with the minor chalcopyrite (at 45.20 to 45.50 m) is vein #3. The hole was completed at 57.61 m in argillite. See table 5 for assay results.

MY07-05 was drilled -52.5° in a westerly direction. A thin carbonate quartz chalcopyrite vein was intersected from 26.96 - 27.16 m (vein #1 [Figure 8]). The final depth of the hole was at 44.81 metres (Table 7).

MY07-06 was drilled at -45° towards the west. The hole was abandoned at 15.85 metres.

MY07-07 was drilled (at -52.5°) towards the west-southwest. Vein #1 was intersected at 31.27 - 31.75 metres (Figure 7). The hole was completed at 61.57 metres in argillite material (For assay results see Table 5).

Photographs were taken of all the drill core boxes. Drill holes 4, 5 and 7 are stored in Vancouver, BC at 1255 West Pender Street. Drill holes 1, 2,3 and 6 that contained only unmineralized till overburden, were discarded.

Table 4 : Missy Drill hole Collar Coordinates

Diamond drill hole	Northing	Easting	Elevation in metres	Dip of hole	Final depth in metres	Casing depth in metres	Azimuth in degrees
MY07-01	6485874	363703	1435	-45°	37.5	18	104
MY07-02	6485874	363703	1435	-60°	32.00	21.34	138
MY07-03	6485797	363752	1409	-60°	25.3	9.75	271
MY07-04	6485733	363732	1413	-78°	57.61	20.12	337
MY07-05	6485727	363745	1408	-52.5°	44.81	15.24	273
MY07-06	6485727	363745	1408	-45°	15.85	15.24	273
MY07-07	6485727	363745	1408	-52.5°	61.57	19.81	256
Total					274.64	119.50	

Table 5: Missy Drilling Intersections

Diamond Drill Hole	Core Type	Assay Numbers	From (m):	To (m):	Apparent Width (m):	Cu (%)	True width (m)
MY07-04	NQ	465307	45.2	45.5	0.24	0.002	0.11
MY07-05	NQ	465309	26.96	27.16	0.20	2.88	0.09
MY07-07	NQ	465317	31.27	31.75	0.48	0.19	0.21

9.6.1 Missy Drill Core Sampling

Under the supervision of George Coetzee the diamond drill core was cut and delivered to Acme Analytical Lab in Vancouver, BC. The samples were analyzed by Acme Analytical Labs of Vancouver (an accredited analytical laboratory), with multi-element Inductively Coupled Plasma Mass Spectrometer (ICP MS) technique. For the 40 element geochemical analysis as well as for samples returning >10,000ppm values (maximum detection limits), the Group 7TX analytical procedure employing Hot 4-Acid "near total" digestion was used, followed ICP-MS analysis. Values greater than 10,000

ppm copper were re-assayed with a further dilution of the solution to give a more accurate analysis of higher grade samples. Gold fire assays was performed on relevant samples with anomalous Ag and As values. Only anomalous values were returned by the Gold fire assaying. See Appendix Drilling assay results of the three drill holes are displaced in table 5:

Figure 8: Missy East - West Drill Section A – B

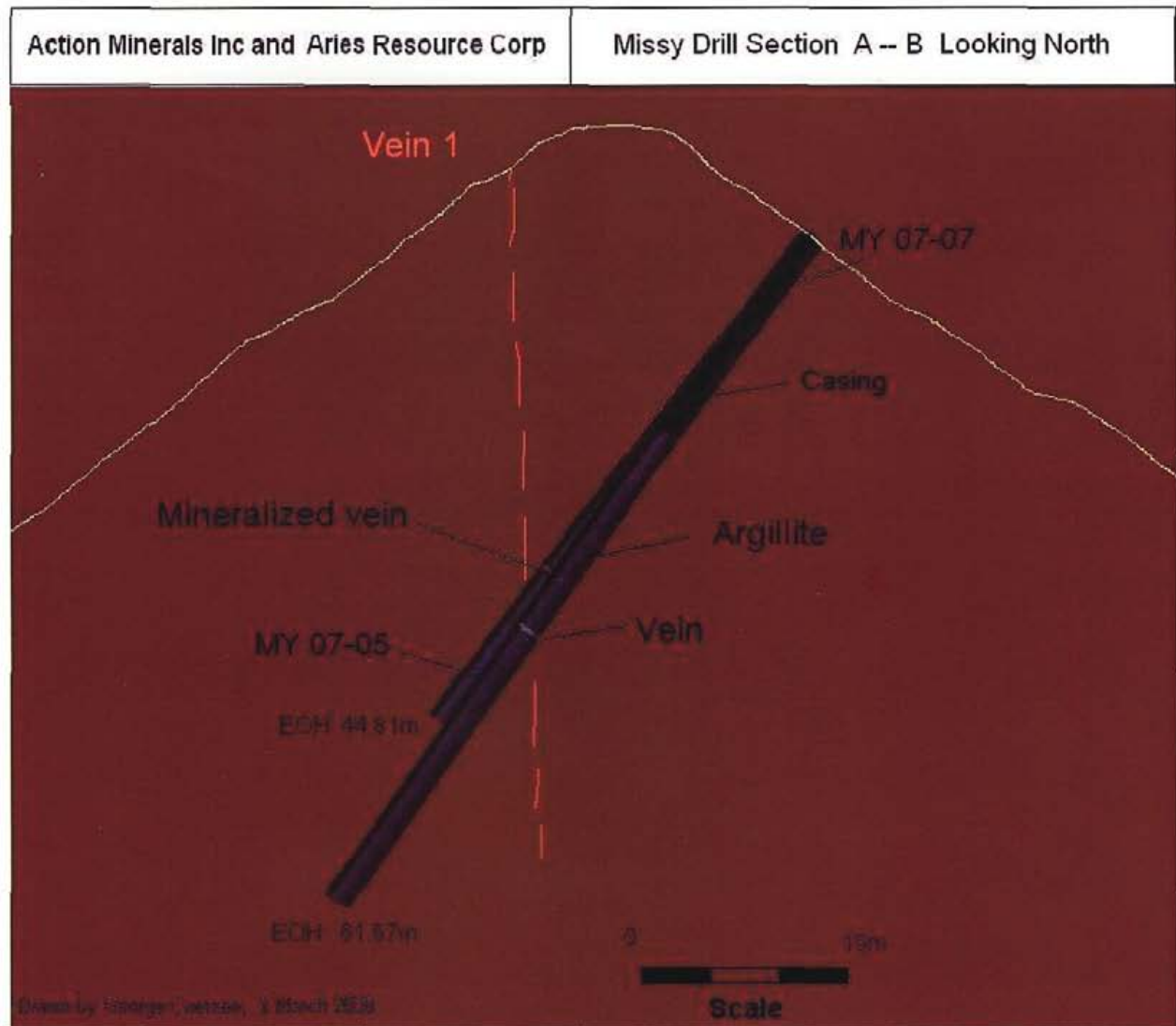
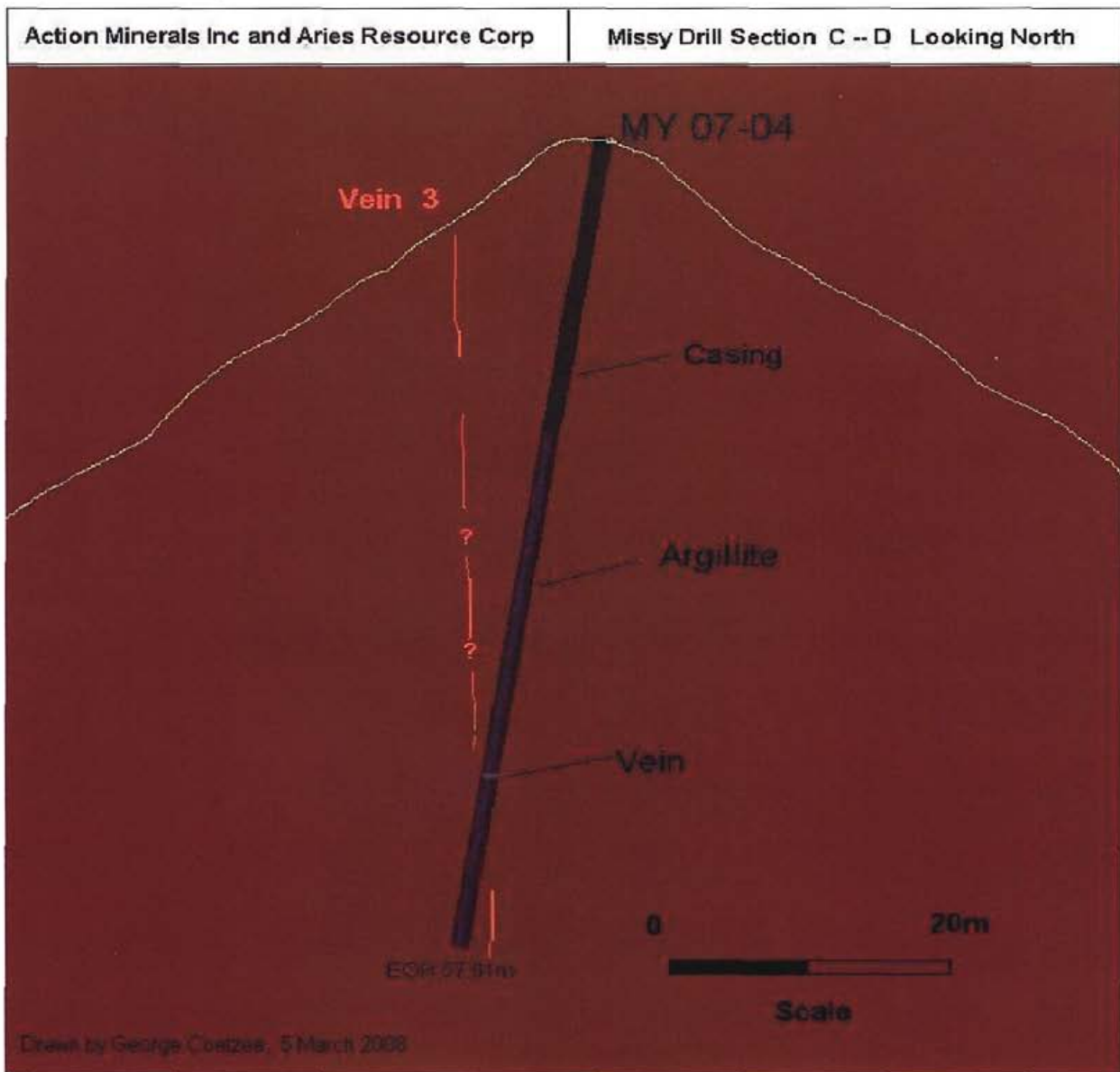


Figure 9: Missy East - West Drill Section C – D



9.7 Missy Drilling Conclusions

Due to the difficult drilling conditions as well as less than ideal placement of drill positions between the two closely spaced creeks (therefore fulfilling the riparian setback regulations), the drill program was not successful in fully intersecting the mineralization identified through chip sampling. The mineralized diamond drill vein intersections assayed from 0 to 3% Cu over ~0.20 m (sub economic grade). The true width of the mineralized carbonate quartz vein #1 in drill holes 5 and 7 were between 9 and

21 cm in thickness (Table 5). The average surface grade of vein #1 over a strike length of 30 metres with an average width of one metre was 4.6 per cent copper. The average grade of vein #2 over a strike length of 25 metres with an average width of 0.5 metre was 1.97 per cent copper. The average grade of vein #3 over a strike length of 30 metres with an average width of one metre was 1.14 per cent Cu. The calcite, quartz and chalcopyrite veins found within the project area, such as at Magnum, Keys and Sox, typically pinch and swell and are a little discontinuous in mineralization and can only be identified through follow up close-spaced drilling.

10.0 Missy NE Extension Mapping

In September 2007, a prospecting and mapping program was initiated on the Missy NE Extension to:

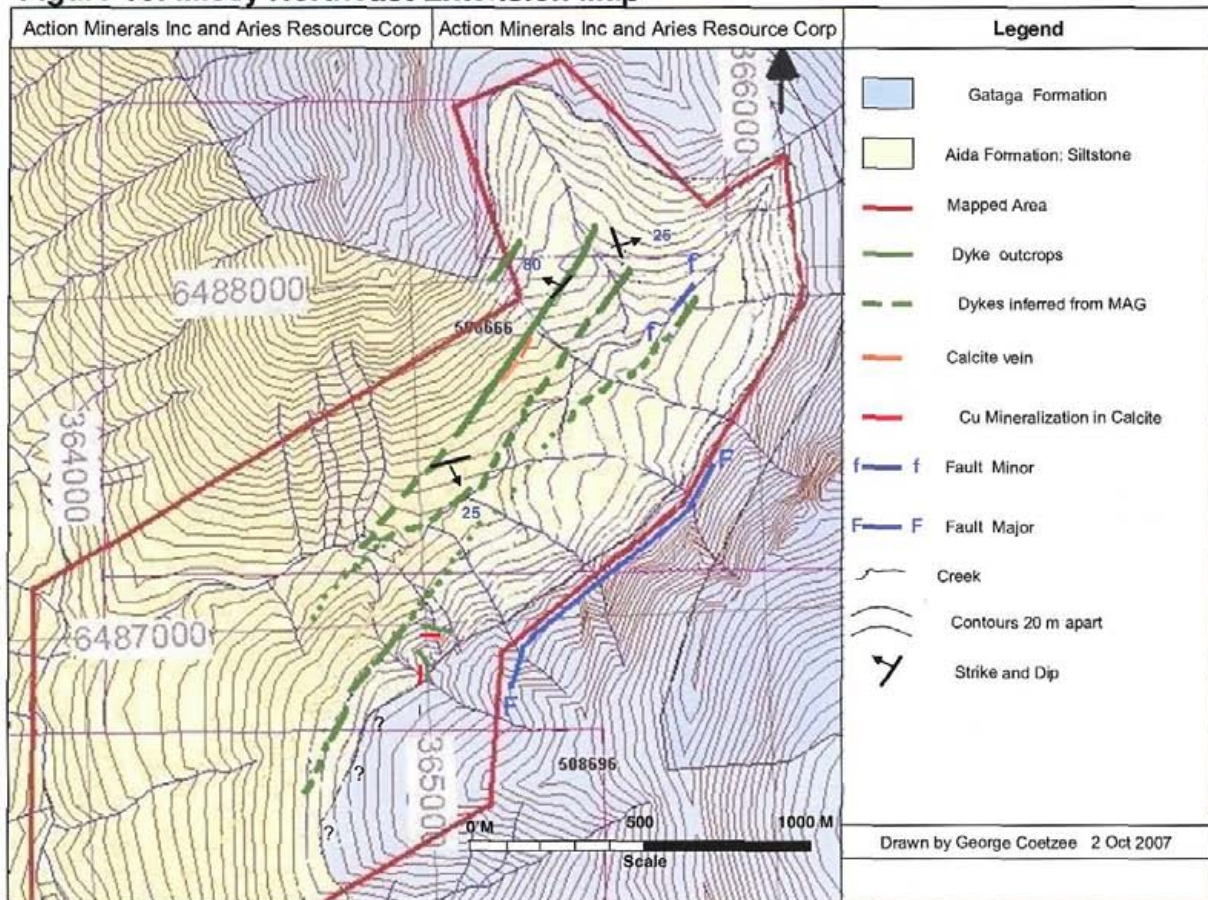
- Identify the mineralized Missy vein extensions towards the north east of the Missy property associated with major dykes or structures.
- Discover new veins paralleling or cross cutting the Missy vein system.
- Map and reconcile the magnetic signatures as located on the airborne magnetic survey and interpreted to be signatures of dykes with the actual dyke locations as well as widths.
- Examine structures or dykes that crosscut the Missy dyke and vein system.

10.1 The Geology of the Missy (NE Extension)

Mapping shows that a large portion of the lower assemblage consists of mainly competent siltstone and sandstone from the Aida Formation (Figure 10; Light yellow in colour), intruded by 5 large mafic dykes (Figure 11). The Aida Formation sandstone and siltstone south-eastern contact with the Gataga Formation is located 400m further south east than outlined by the regional BC Government geology map, on the large blue fault line displayed in Figure 10). This fault could be part of the larger regional thrust fault sequence. The bedding of siltstone and sandstone strikes in a general northeast direction dipping at about 20 to 40° southeast. There is no indication of folding within the

blue colour) consists of mudstone, siltstone and sandstone that are generally well foliated.

Figure 10: Missy Northeast Extension Map



http://webmap.em.gov.bc.ca/mapplace/minpot/ex_assist.cfm (MTO Tenure Number 501534)

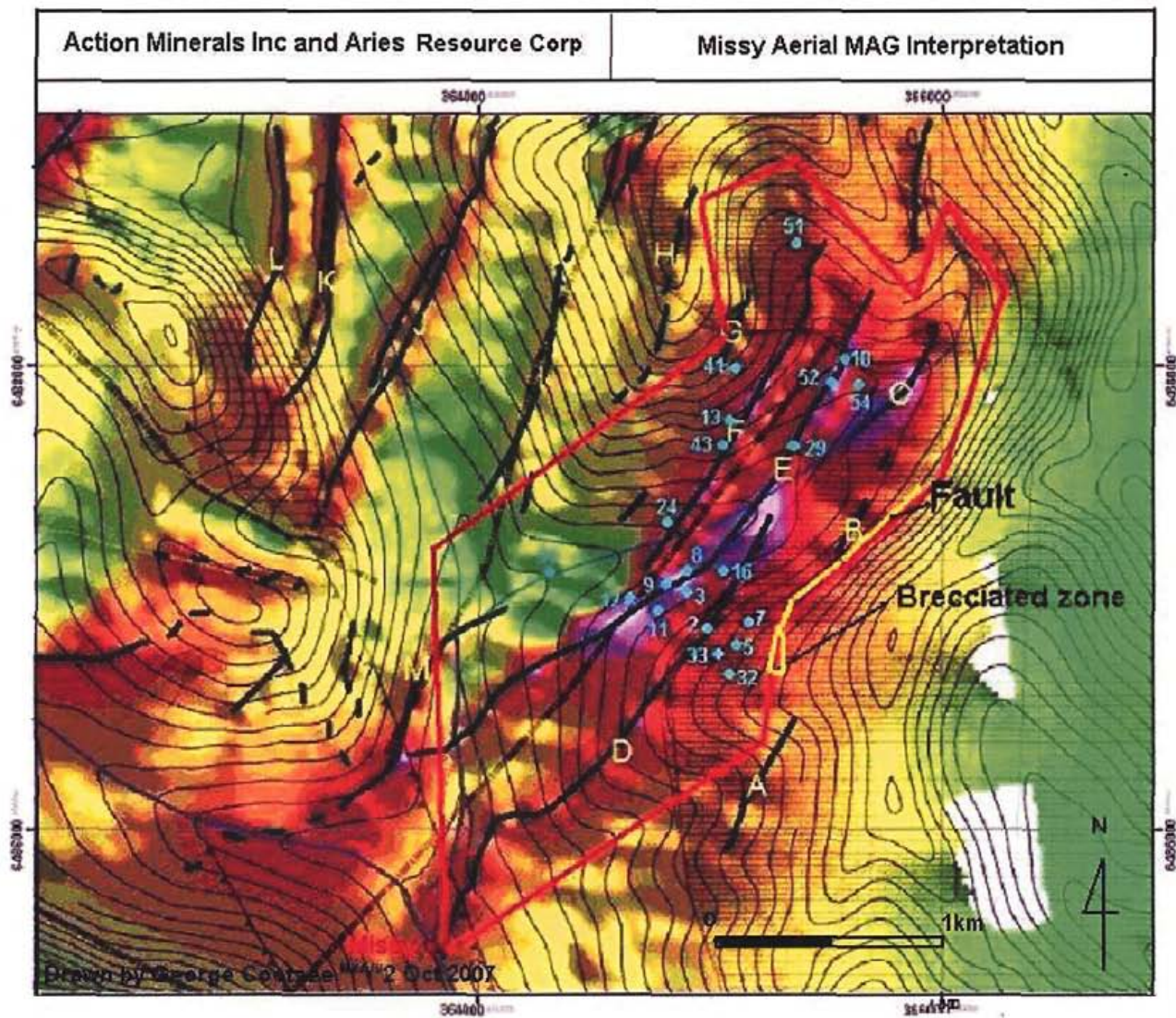
Please also see the enlarged Missy map in Appendix F

10.2 Missy Mapping Correlation with the Aerial Magnetics

The field program has determined that the large, northeast trending magnetic feature outlined in the 2006 airborne geophysical survey consists of four major northeast trending dyke/fault structures B, C, D, E and F (Figure 11). The Major dykes (A, B, C, D, E, F and G) in the corridor average between 20 to 30 metres in width. These dykes are predominantly massive and have a blocky appearance and show minor evidence of shearing or faulting. Minor dykes (approximately one to two kilometres in length) with associated mineralized veins appear to crosscut the larger

northeast trending regional dykes (five to twelve kilometres long). Analysis of the system suggests that the major dyking predates the mineralization events, which are located within fault or shear zones. The surface mapping of dykes and shear zones in the Missy NE Extension was found to correlate closely with the results of the 2006 airborne geophysical survey. All of these structures have the potential of hosting some mineralization.

Figure 11: Missy Aerial MAG Interpretation



Drawn by George Coetzee, BSc. Honours, 2 October 2007

10.3 Missy Faulting (NE Extension)

Approximately two kilometres northeast of the Missy showing, a large iron stained pyritic fault zone with a strike length greater than 700 metres was located. The width of the siliceous fault zone is about one to three metres wide (Figure12).

Figure 12: Missy Fault



Photo by George Coetzee, BSc. Honours, September 2007

A 20 metre wide iron stained/oxidized brecciated area was also located near the southern portion of this fault. Both the fault zone and brecciated area contained pyrite mineralization (Figure 11 and Figure13: in yellow).

Figure 13: Missy Brecciated Zone



Photo by George Coetzee, BSc. Honours, September 2007

Approximately three kilometres northeast of the Missy showing a pyritic fault and shear zone within a sandstone unit is located within the valley, approximately 30 metres northwest of a parallel mafic dyke. (Figure 11: number 54 and Figure 14) The 100m shear fault zone disappears under talus material towards the southwest and northeast.

Figure 14: Missy Pyritic Fault / Shear Zone

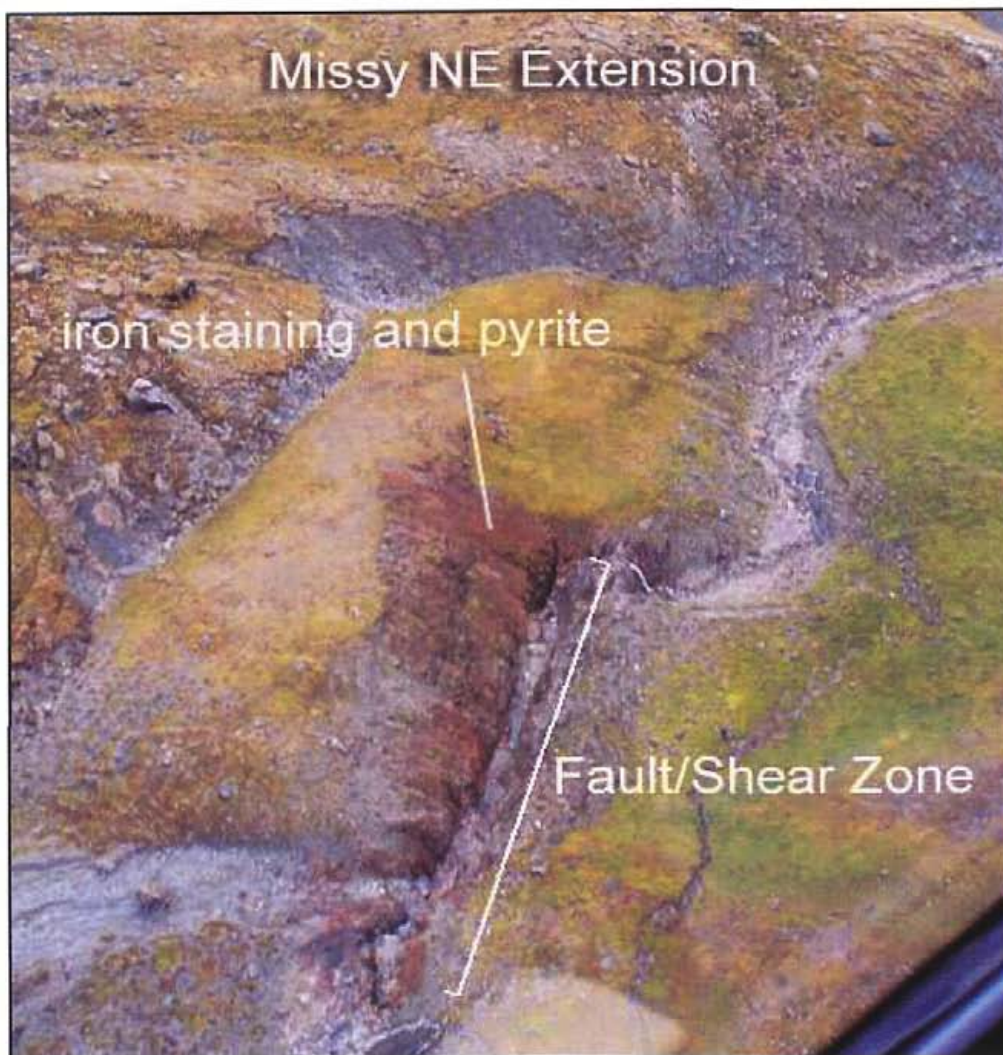


Photo by George Coetzee, BSc. Honours, September 2007.

Drawn by Reza Mohammed BSc.

10.4 Missy Veins (NE Extension)

An approximately one metre wide calcite vein A with some malachite mineralization (Figure 11: Number 7) is sporadically exposed over a 50 metre strike length. This vein parallels a 5 m dyke that is nearly perpendicular to the northeast Missy dyke trend. Mineralized malachite float (vein A) extends for 200 metre towards the southwest.

Figure 15: Missy Vein A (Northeast Extension)



Photo by George Coetzee, September 2007

A thin poorly exposed malachite vein B was located about 250m southwest of vein A on the contact of a mafic dyke(Figure11: Number 5). The length of the mineralized zone is approximately 25m. Both dykes show evidence of faulting and shearing.

A one metre thick calcite vein C, adjacent to mafic dyke F, was observed approximately 300 metres northwest of the main Missy structure (Figure11: Number 43). Vein C has a strike length of approximately 150 metres. Minor oxidization on surface indicates the presence of minor copper mineralization.

One sample was located 300 metres northwest of the Missy dyke D, within a one metre wide calcite vein on the south-eastern contact of a competent mafic dyke. The calcite dyke contains. (Figure11: number 43 and figure 16).

Figure 16: Missy Calcite Vein

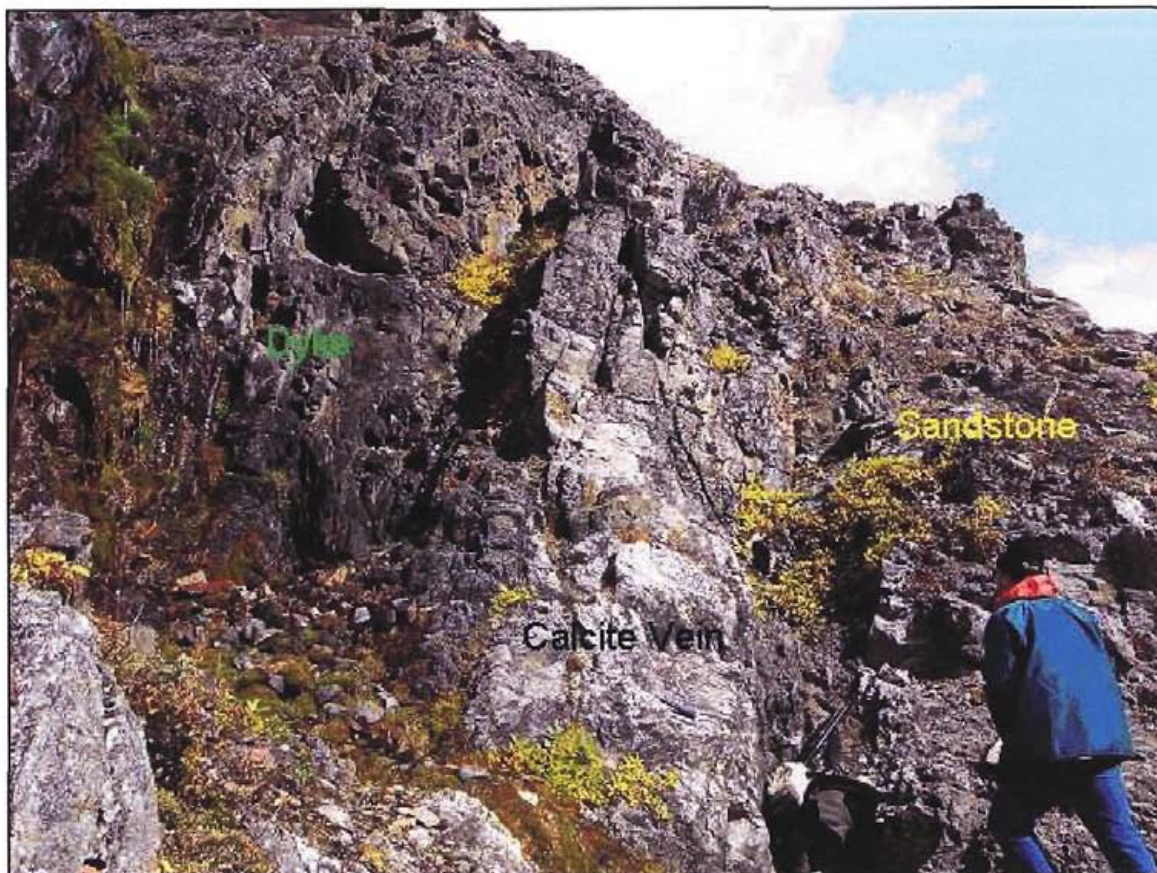


Photo by David Peake, Drawn by George Coetzee BSC Honours, September 2007

10.4.1 Missy Vein Sampling (NE Extension)

Three grab samples were taken from the above mentioned veins A B and C. (Figure 11: Number 5, 7 and 43). Three samples of mineralized float were taken within the valley and one sample on the pyritic fault and shear zone (Figure 11: number 54 and Figure 14). The seven samples were sent to ACME Analytical Laboratories (Vancouver, BC) for multi-element analysis. For anomalous assay results see table 6. Mapping and sampling was performed by George Coetzee BSc. Geo. Honours under the direction of the Company's Qualified Person, John Kowalchuk, and P. Geo.

Table 6: Missy Sample Results (NE Extension)

Sample Number	Location	Sample Type (Width in cm)	% Copper	% Pb	Vein Width in Meter
1	Missy NE Ext	Float	Trace		N/A
2	Missy NE Ext.	Float	1.18		N/A
3	Missy NE Ext.	Grab (20 cm)	0.03	0.08	
4	Missy NE Ext.	Grab (15 cm)	1.44	0.10	1.5
5	Missy NE Ext	Float	40.29	Trace	N/A
6	Missy NE Ext	Grab	0.12	Trace	1.5
7	Missy NE Ext.	Float	0.01		

11.0 General Geology of the Churchill Mine (Magnum vein, Minfile No094K 003)

The showing occurs in the Aida Formation of the Muskwa Assemblage, which comprises shale or slate, dolomitic and calcareous shale, dolostone and minor limestone (Assessment Report 3535; Geology, Exploration and Mining in British Columbia 1971; Geological Survey of Canada Memoir 373). In the area around the Churchill Mine, the formation consists of a lower unit of dark grey thin-bedded calcareous shale and interbedded calcareous shale and limestone, and an upper unit of interbedded buff- to orange-weathering dolomitic shale and dolostone, locally containing beds of algal dolostone. A large number of diabase dykes cut the sedimentary rocks, ranging from a metre to about 100 metres in width and striking from northeast to east. There is minimal contact metamorphism of the sedimentary host rock, although the

adjacent strata are commonly 'bleached' for several metres. The dykes are evenly distributed in the mine area and generally follow the same fracture and alteration zone that contains cupriferous quartz-ankerite veins. In the mine workings and surface showings, dykes are clearly post-mineralization, truncating the veins. Other dykes, locally known as "grey dykes", are known to cut transversely across the zone of mineralization and alteration, and individual veins, striking in a general northwest direction. These dykes are of trachytic composition, contain disseminations and stringers of pyrite, and are generally only a few metres wide.

The sedimentary rocks are deformed into a large number of folds which plunge gently to the south and southeast. These structures range from a metre to several hundred metres in amplitude and are invariably asymmetric, with gently-dipping west limbs and steep east limbs, and axial planes dipping to the west and southwest (Geology, Exploration and Mining in British Columbia 1971, Plate 3). The ubiquitous slaty cleavage in the Aida Formation rocks is parallel to the axial planes of these folds. In the Magnum Creek area, diabase dykes, fracture zones and cupriferous veins all have trends that are at a high angle to these fold structures, and are apparently not deformed by them. It appears that the dykes and veins filled a system of fractures, generally striking northeast, that developed after the folding and transverse to the fold axes.

Faults are not common on the property. A number of small faults and shear zones have been mapped, but none appear to be very large except at the Churchill Mine zone, where there has been considerable faulting. Most of these faults lie parallel to the zone and cut both mineralized veins and dykes, but within the mine workings at least two faults have been mapped which strike across the zone, dipping southwest at approximately 40 degrees, and are thought to displace ore shoots in a reverse manner.

Within the Magnum zone itself, the deformation is much more heterogeneous than that described above, shown by highly variable fold axes. The cleavage, partly curved and wavy, strikes predominantly south-southwest, with a dip of approximately 60 degrees to the east. In general, bedding dips gently to moderately southeast and apparently forms

the southeast limb of a broad anticline, the hinge zone of which approximately follows Magnum Creek. Also within this zone, the originally calcareous succession is conspicuously non-calcareous, the limestone and calcareous argillite having been extensively altered by decalcification to coarsely crystalline Ferro dolomite and ankerite. The same alteration has produced abundant graphite in shale, locally with coarse ankerite crystals. In addition, pyrite was developed in the west part of the zone forming seams and disseminations roughly concordant with bedding.

Mineralization at the Magnum deposit occurs in cupriferous quartz-ankerite veins in the sub vertical north- to northeast-striking shear and fracture zones. The local preservation between the principal veins of septa of schistose country rock or brecciated quartz stock works suggests that the Magnum zone was originally controlled by a narrow shear zone (or a zone) which was subsequently exploited by hydrothermal activity and later by dyke intrusion. In general, this zone of deformation, alteration, mineralization and dyke intrusion trends 035 degrees, dips steeply and is up to 90 metres wide. It has been partly explored for a length of 1375 metres and to a depth of 365 metres. As many as ten veins have been observed, concentrated in the centre of the zone, although some may prove to be extensions of others. They vary in width from less than 1 metre to as much as 7.6 metres and possess continuity, both on strike and in depth, which may measure a hundred metres or more. As many as three parallel principal veins occur within a width of 45 metres or less across the zone. Numerous subsidiary veins are present, some of which are parallel to the principal veins, and others which have an oblique, northerly trend, and are probably branches of the principal veins.

In more detail, the veins consist of varying proportions of ankerite, quartz, chalcopyrite, and locally pyrite, together with partly replaced remnants of the sedimentary host rock. Very minor amounts of bornite have also been observed. Malachite and azurite are common on the surface. Pyrite is locally prominent, but is generally less than about 10 per cent of the total sulphides in the ore. Chalcopyrite is intimately associated with quartz, although in some places the quartz is so sparse that the vein appears to consist of massive chalcopyrite. Chalcopyrite tends to increase noticeably where a vein

changes direction. Such jogs occur over only a metre or so and their shape is such as to displace the northern part of the vein west or, alternatively, the upper part westward by a metre (Canadian Institute of Mining, Transactions, 1971). The latter sense of displacement is affected also by at least one of several minor syn- and post-mineralization faults which occur in the northern part of the mine. These mineralized faults dip approximately 40 degrees southwest, and locally displace the upper parts of two principal veins about 9 metres west along the strike of the fault.

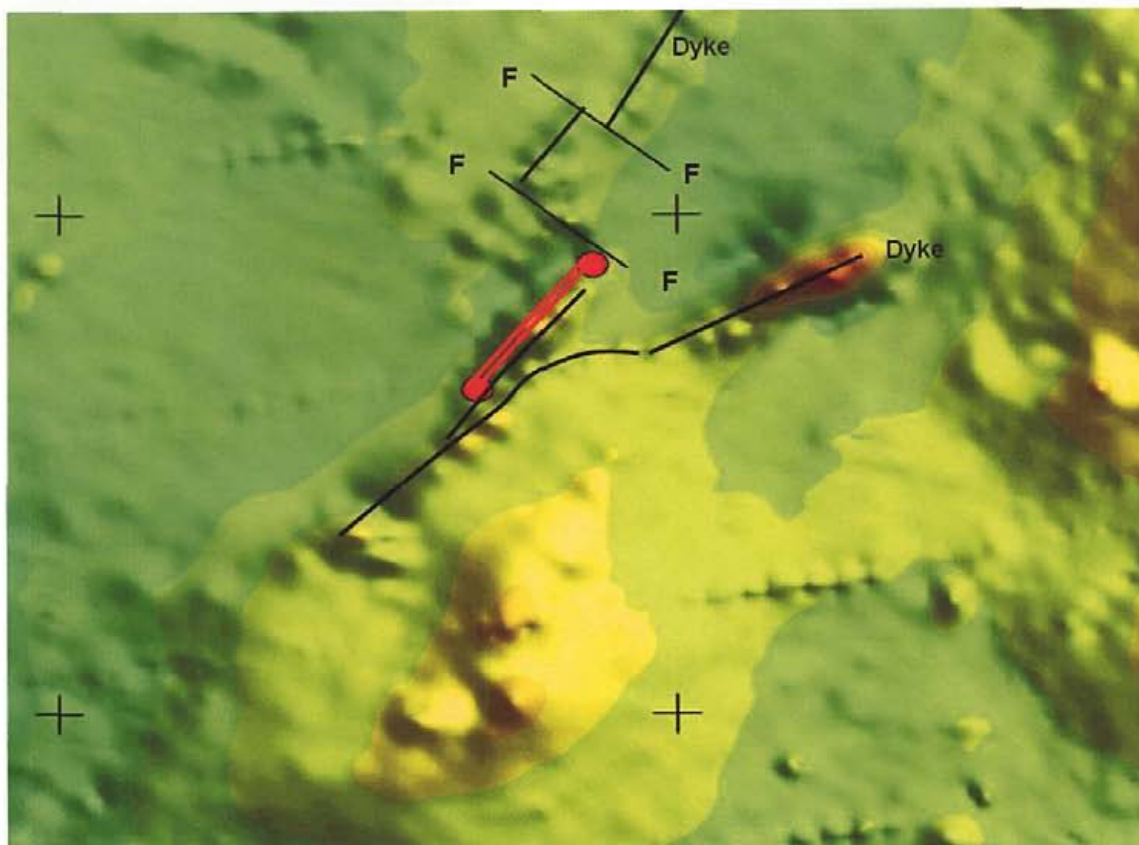
A post-ore diabase dyke of irregular shape and generally steep dip closely follows the southeast side of the vein system and invades it progressively southwards in the zone. The dyke is less than 3 metres wide in the northeast of the zone, but widens southwards and splits locally into two or more parallel branches with an aggregate width which may exceed 45 metres. In places, the dyke becomes sill-like; subsidiary dykes extend west across the vein system. Along part of its length, the main dyke is followed by one or more steep faults, with unknown displacement, near which the diabase is propylitically altered. In the northern part of the mine zone, the dyke adjoins one or more veins, and locally invades and obliterates them; this occurred more extensively in the southern part of the mine zone.

11.1 Sampling of the Magnum Vein and Brecciation (Churchill Mine)

Previous mining and underground diamond drilling indicated that only 850m of the 1400m vein strike length and 370m of the estimated 1000m vein depth were mined. Exploitation of the Magnum veins was terminated when a south east striking fault was intersected on the northern extent of the underground workings. A large folded brecciated and veined mineralized zone approximately 20 to 30 metres wide and over 200m in length, was observed on surface just northeast of this faults .The location of the folded brecciated zone is displayed in Figure 5.

In the MAG survey the southeast trending faults as well as the Magnum dyke is clearly visible (Figure 17).

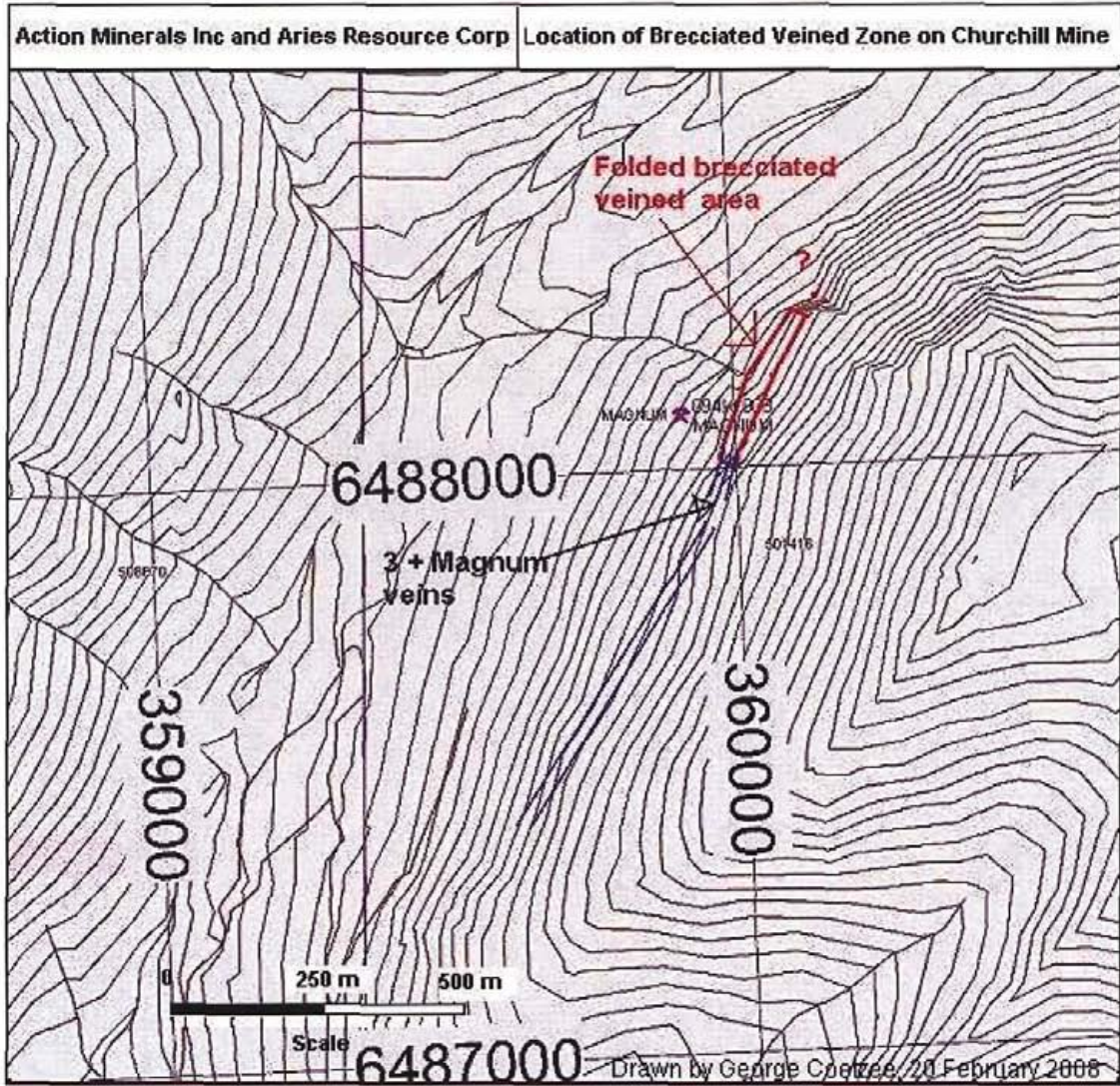
Figure 17: Magnetic interpretation of the Churchill Mine



Updated by George Coetzee, 20 February 2008

The two magnum- as well as the two horse tail-veins striking in a northeast direction cut through a folded and brecciated zone. The veins are the northern extension of the two mined Magnum veins (Figure 18). These types of breccia zones normally carrier relatively large tonnages of copper, as on the Toro property. The northern extension of the Neil vein displays a similar brecciated mineralized structure, after being cut by a northwest striking fault zone. Therefore the brecciated area was sampled in September 2008 by David Peake BSc. Geo. under supervision of George Coetzee BSc. Geo Honours and J. Kowalchuk P.Geo. (Figure 18).

Figure 18: Location of the Magnum Folded, Brecciated and Veined Zone



http://webmap.em.gov.bc.ca/mapplace/minpot/ex_assist.cfm (MTO Tenure Number 501416)

11.1.1 Sampling Results of the of the Magnum Vein and Brecciation

24 grab samples were chipped at 4 to 5 metres intervals over a 100m section on the south western lower section of the folded/brecciated zone). Only the lower contact of the folded, brecciated and veined zone (Zone) as well as vein 2 was sampled (Figure 18). The 24 grab samples were chipped at 4 to 5 metre intervals over a 100 metre of the south western section of the Zone (Figure 19). Nine samples were taken from the folded/brecciated area and 15 samples of vein 2. Vein 2 assayed on average at 4.35 % Cu and the brecciated zone at 0.24% Cu (Table 7).

Figure 19: Magnum Folded, brecciated and Veined Zone (NE Extension)

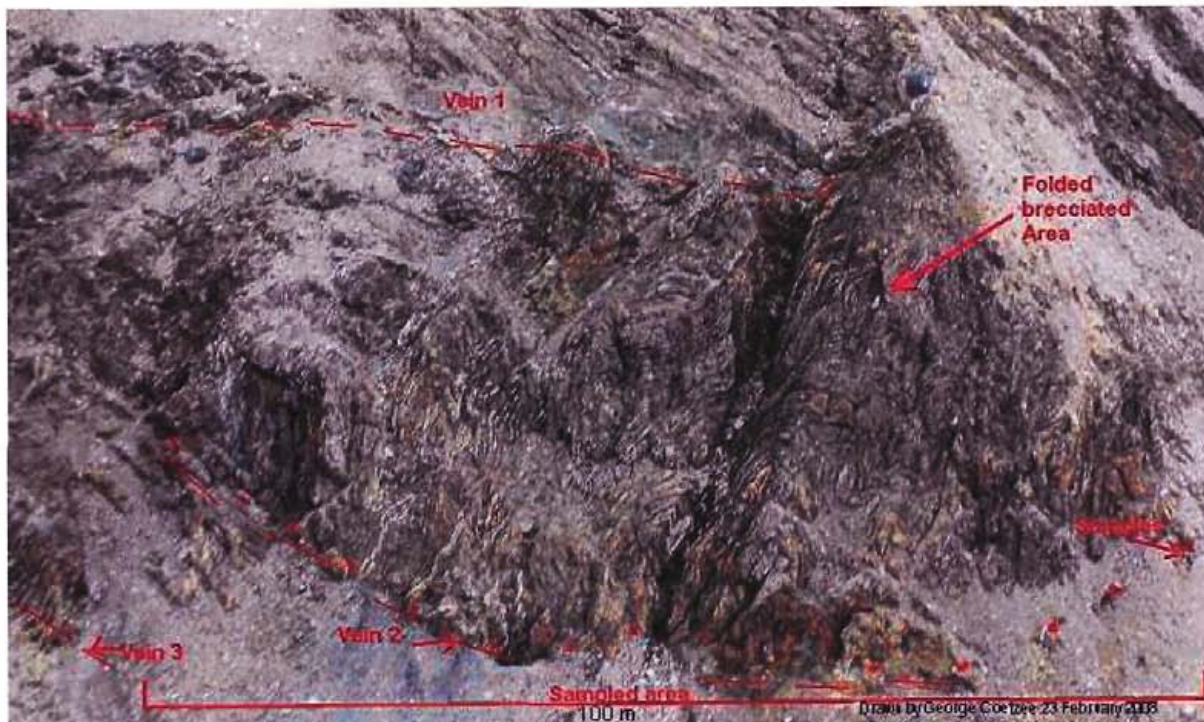


Table 7: Magnum Brecciation Sample Results

Sample Number	Type	Width m	Type:	% Copper
465101	Grab	0.2	Breccia	trace
465102	Grab	0.2	Breccia	0.01
465103	Grab	0.2	Breccia	trace
465104	Grab	0.2	Breccia	0.14
465105	Grab	0.2	Breccia	0.02
465107	Grab	0.2	Breccia	0.01
465108	Grab	0.2	Vein 2	0.14
465109	Grab	0.2	Vein 2	13.54
465110	Grab	0.2	Vein 2	20.36
465111	Grab	0.2	Vein 2	7.11
465112	Grab	0.2	Vein 2	2.17
465113	Grab	0.2	Breccia	0.60
465114	Grab	0.2	Breccia	0.51
465115	Grab	0.2	Breccia	1.40
465116	Grab	0.2	Vein 2	0.08
465117	Grab	0.2	Vein 2	0.08
465118	Grab	0.2	Vein 2	4.00
465119	Grab	0.2	Vein 2	0.74
465120	Grab	0.2	Vein 2	0.63
465121	Grab	0.2	Vein 2	0.99
465122	Grab	0.2	Vein 2	0.03
465123	Grab	0.2	Vein 2	6.53
465124	Grab	0.2	Vein 2	1.51
465125	Grab	0.2	Vein 2	6.86

12.0 Conclusions

12.1 Missy Drilling Conclusions

- Due to the difficult drilling conditions as well as less than ideal placement of drill positions, between the two closely spaced creeks (therefore fulfilling the riparian setback regulations), the drill program was not successful in fully intersecting the mineralization identified through vein chip sampling. The mineralized diamond drill vein intersections assayed from 0 to 3% Cu over ~0.20 m (un-economic grade). The true width of the mineralized carbonate quartz vein #1 in drill holes 5 and 7 were between 9 and 21 cm in thickness (Table 5). The average surface

grade of vein #1 over a strike length of 30 metres with an average width of one metre was 4.6 per cent copper. The average grade of vein #2 over a strike length of 25 metres with an average width of 0.5 metre was 1.97 per cent copper. The average grade of vein #3 over a strike length of 30 metres with an average width of one metre was 1.14 per cent Cu. The calcite, quartz and chalcopyrite veins found within the project area, such as at Magnum, Keys and Sox, typically pinch and swell and are a little discontinuous in mineralization. Therefore, closer spaced follow-up drilling is required to ascertain the grades of the veins

12.2 Missy Mapping Conclusions (NE Extension)

- All the dykes delineated by the 2006 magnetic survey were located and partially mapped within the mapping area.
- The Geology of the Aida Formation and dykes were confirmed and contacts were more accurately plotted. New faults and veins were discovered.
- Iron oxidization associated with faulting, shear zones and a sheared/faulted dyke as well as sporadic Cu mineralization near point 33 indicate that there is a possibility of discovering additional copper mineralization under the talus material, which covers at least 60% of the mapping area.

12.3 Magnum Brecciation Sampling Conclusions

- The grab sampling indicates that there is low copper mineralization within the folded and brecciated zone. Only approximately 5% of the brecciation and vein 2 were sampled due to weather and safety concerns. Consequently the potential still remains of locating copper mineralization within the folded and brecciated zone.

13.0 Recommendations

13.1 Missy Drilling Recommendations

- .Complete a low level EM survey of the area as to identify potential mineralization under the talus material.
- .Complete a VLF survey of the area as to confirm potential mineralization under the talus material.
- Based on the above mentioned surveys decide where to drill if required.

13.2 Missy Mapping Recommendations (NE Extension)

- Complete a low level EM survey of the area as to identify potential mineralization under the talus material.
- Map areas of interest namely:
 1. EM anomalies identified by low level EM survey.
 2. The large mainly inaccessible iron stained pyritic fault zone and the fractured/ brecciated area adjacent to the (thrust) fault zone.
 3. In the vicinity of the two dykes with the associated copper mineralization.
 4. Sheared fault zones paralleling or crosscutting the dyke structures.
- Follow up with a surface MAG and VLF survey where required.

13.3 Magnum Brecciation Sampling Recommendations

- Grid sample the rest of the folded and brecciated zone on Churchill Mine, as to ascertain the mineral content of the total Zone.
- Based on assay results decide if first phase diamond drilling is necessary?

14.0 Sampling Method and approach

14.1 Sample Preparation, Analysis and Security

All the drill core samples were cut; with one half delivered to Acme Analytical Labs of Vancouver, BC, for processing and analysis. The Acme Analytical quality control system complies with requirements of international standards ISO 9001:2000 and ISO

17025:1999. Laboratory procedures employ comprehensive quality control (QC) programs to monitor sample preparation and analysis. QC protocols include the use of barren material to clean sample equipment between sample batches, and size monitoring of crushed material. Analytical accuracy and precision are monitored by the analysis of reagent blanks, reference materials, and replicate samples. Acme Analytical utilizes bar coding and scanning technology providing complete chain of custody records for sample preparation and analytical process.

Each entire sample was passed through a primary crusher to yield a product where greater than 70% is less than 2 mm. A split is then taken using a stainless steel riffle splitter. The crushed sample split of 200 - 300 grams is ground using a ring mill pulverizer with a chrome steel ring set, with the specification for this procedure calling for greater than 85% of the ground material to pass through a 75 micron (Tyler 200 mesh) screen.

Gold was analyzed using the AU-ICP21 fire-assay technique on a 30 gm pulverized rock sample, with atomic absorption finish. For the remaining 47 elements, the ME-MS61 analytical procedure employing four acid "near total" digestion was used, followed by mass spectrographic finish. Samples returning copper values >10,000 ppm were re-analyzed by ore grade CU-AA62 process, where a prepared sample was subjected to four acid "near total" digestion, followed by atomic absorption.

15.0 STATEMENT of COSTS

Statement of Costs

ITEM			COST
Staff			
George Coetzee BSc. Geo. Hon.		10 days @ \$400/day (Office)	\$4,000.00
George Coetzee BSc. Geo. Hon.		13 days @ \$500/day	\$6,500.00
Junior Geologists		67 days @ \$250/day	\$16,750.00
Med-techs		77 days @ \$250/day	\$19,250.00
Employees/Contractors		12 workers 77 days	\$466,724.38
Services			
Helicopter Support		77 days @ \$4550/day	\$350,350.00
Trucking			\$33,734.56
Drill Rental		77 days @ \$3246/day	\$250,000.00
Supplies and Consumables			
Diesel Fuel		316 drums @ \$85/drum	\$26,860.00
Jet-A Helicopter Fuel		137 drums @ \$195 drum	\$26,715.00
Camp rentals			
	computer rentals	7 laptops & accessories 77 days @ \$196/day	\$15,092.00
	GPS rental	77 days @ \$10/day	\$77,0.00
	Skid steer	77 days @ \$294/day	\$22,638.00
	ATV rentals	3 ATVs 77 days @ \$171/day	\$13,167.00
	Core cutter rentals	77 days @ \$ 61/day	\$4,697.00
Communication Equipment	Satellite phones		
	Radios		
	Communication usage		
	wireless internet	77 days @ \$343/day	\$26,411.00
Transportation Airfares		17 flights	\$8,932.20
Drilling Equipment			\$17,917.56
Report			\$7,000.00
		total	\$1,316,738.70

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17.0 Glossary

Conversion Factors

To Convert From	To	Multiply By
Feet	Meters	0.305
Meters	Feet	3.281
Miles	Kilometres ("km")	1.609
Kilometres	Miles	0.6214
Acres	Hectares ("ha")	0.405
Hectares	Acres	2.471
Grams	Ounces (Troy)	0.03215
Grams/Tonnes	Ounces (Troy)/Short Ton	0.02917
Tonnes (metric)	Pounds	2.205
Tonnes (metric)	Short Tons	1.1023

Mineral Elements

Au	Gold	Ce	Cerium	La	Lanthanum
Ag	Silver	Co	Cobalt	P	Phosphorus
Cu	Copper	Ga	Gallium	Se	Selenium
Ba	Barium	Ge	Germanium	U	Uranium

Alteration: Any change in the mineralogical composition of a rock that is brought about by physical or chemical means.

Ankerite: A dolomite group mineral associated with iron ores.

Anomaly: A geochemical or geophysical character which deviates from regularity.

Anticlinorium: A regional scale configuration of many folded, stratified rocks in which rocks dip in two directions away from the crests. Reverse of synclinorium. The crest is called axis.

Arcuate: Curved or bowed.

Argillic: Pertaining to clay or clay minerals. Disseminated precious metal deposits may exhibit "argillic" alteration characterized by the formation of the clay minerals kaolinite and montmorillonite. Epithermal precious metal deposits may exhibit "advanced argillic" alteration characterized by the clays dickite, kaolinite and pyrophyllite.

Basic: An igneous rock having relatively low silica content, such as gabbro and basalt. Basic rocks are relatively rich in iron, magnesium, and/or calcium.

Breccia: A rock composed of highly angular coarse fragments.

Clastic: Consisting of fragments moved from their place of origin.

Conglomerate: Detrital sedimentary rock made up of more or less rounded fragments of such size that an appreciable percentage of volume of rock consists of particles of granule size or larger.

Cratonic: Pertaining to the relatively immobile part of the earth, the generally large central portion of a continent.

Detrital Sedimentary Rock: Rock formed from accumulation of minerals and rocks derived from erosion of previously existing rocks or from weathered products of these rocks.

Diabase: Rock of basaltic composition, essentially labradorite and pyroxene, characterized by ophitic texture.

Dolomitic: Having the characteristics of dolomite, where calcium-magnesium carbonate predominates, rather than calcium carbonate which comprises limestone.

Epigenetic: A mineral deposit formed later than the enclosing rocks. In ore petrology, applied to mineral deposits of later origin than the enclosing rocks or to the formation of secondary minerals by alteration.

Epithermal Deposit: Formed at shallow depths by low-temperature hydrothermal solutions.

Felsic: Composed of light-coloured minerals such as feldspar and quartz.

Ga: Billion years.

Gangue: Accessory minerals associated with ore in a vein.

Hydrothermal: An adjective applied to heated or hot aqueous-rich solutions, to the processes in which they are concerned, and to the rocks, ore deposits and alteration products produced by them.

Ignimbrite: Volcanic glass shards that when cooling wrapped around rock crystals creating a "welded" texture.

Ma: Million years.

Metasomatism: Process whereby rocks are altered when volatiles exchange ions with them and a new mineral may grow inside the body of an old mineral.

Moraine: A mound, ridge, or other distinct accumulation of unsorted, unstratified glacial drift deposited, chiefly by direct action of glacier ice, in a variety of topographic landforms.

Normal Fault: A fault in which the hanging wall is lowered relative to the foot wall.

Ophitic: Rock texture in which lath-shaped plagioclase crystals are enclosed, wholly or in part, in later-formed mineral augite.

Orogeny: Mountain building, particularly by folding and thrusting.

Pluton: Igneous rock formed beneath the surface by consolidation from magma.

Potassic Alteration: The generally high-temperature alteration process where potassium is introduced replacing calcium producing secondary orthoclase (potassium feldspar) and biotite.

Pyroclastic: Volcanic materials explosively or aurally ejected from a volcanic vent.

Reverse/Thrust Fault: A fault in which the hanging wall is raised relative to the foot wall.

Sericitic Alteration: Forming sericite from the decomposition of feldspars.

Skarn: Derived from limestone and dolomite by the addition of silica, iron, magnesium, and aluminium to form a suite of lime-bearing silicate minerals.

Sodic Alteration: The alteration process where sodium is introduced replacing calcium, and sodium-rich minerals such as albite, scapolite, and hornblende predominate.

Stockwork: A rock mass interpenetrated by small veins.

Strike-slip Fault: A fault where displacement is in the strike direction of the fault.

Subduction: Descent of one tectonic unit under another.

Synclorium: A regional scale configuration of many folded, stratified rocks in which rocks dip downward from opposite directions to come together in troughs. Reverse of Anticlinorium.

Talus: Slope established by accumulation of rock fragments at the foot of a cliff or ridge. Rock fragments that form talus may be rock waste, slide rock, or pieces broken by frost action. Widely used to mean the rock debris itself.

Till: unsorted glacial sediment. Glacial drift is a general term for the coarsely graded and extremely heterogeneous sediments of glacial origin. Glacial **till** is that part of glacial drift which was deposited directly by the glacier. It may vary from clays to mixtures of clay, sand, gravel and boulders.

Trachytic: A textural term applied to the ground mass of volcanic rocks in which small crystals of feldspar are arranged in parallel or sub-parallel fashion corresponding to the flow of the lava.

Transverse Fault: A fault with a strike which cuts across the general structure.

18.0 Certificate

Bradford Minerals Explorations Ltd.

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Email: Georgeaction@gmail.com

I, George Coetzee, BSc (Honours) in Geology, hereby certify that I am working for Bradford Minerals Explorations Ltd. (that was contracted By Aries Resources Corp and Action minerals Inc).

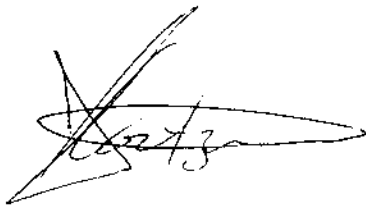
#1- 1255 west Pender St
Vancouver, BC. Canada
V6E 2V1

I graduated with a BSc (Honours) in Geology from University of Pretoria in South Africa in 1981.

I have worked as a geologist for a total of 25 years since my graduation from University.

I was on the property for 10% of the time while the diamond drilling took place.

I am responsible for the preparation of all the sections of the report titled; Assessment Report on the Missy Drilling and Mapping as well as the Sampling of the Missy and Magnum Properties, under the supervision of John Kowalchuk P. Geol.



George Coetzee, BSc. (Honours) in Geology

APPENDIX A
Claim Information

Trident Copper Project Claim Information

<u>Tenure Number</u>	<u>Claim Name</u>	<u>Owner</u>	<u>Map No.</u>	<u>Good To Date</u>	<u>GoodTo Code</u>	<u>Area</u>
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520529	LYNDA5	146886 .	094K	2008/jun/28	20080628	427.616
515811	SOCRATES 56	146886 .	094K	2008/jul/01	20080701	319.277
515813	SOCRATES 57	146886 .	094K	2008/jul/01	20080701	302.597
515816	SOCRATES 58	146886 .	094K	2008/jul/01	20080701	403.095
515817	SOCRATES 59	146886 .	094K	2008/jul/01	20080701	403.34
515818	SOCRATES 60	146886 .	094K	2008/jul/01	20080701	403.333
515819	SOCRATES 61	146886 .	094K	2008/jul/01	20080701	419.939
515820	SOCRATES 62	146886 .	094K	2008/jul/01	20080701	420.678
515821	SOCRATES 63	146886 .	094K	2008/jul/01	20080701	420.988
515822	SOCRATES 64	146886 .	094K	2008/jul/01	20080701	420.979
515823	SOCRATES 65	146886 .	094K	2008/jul/01	20080701	303.142
515824	SOCRATES 66	146886 .	094K	2008/jul/01	20080701	421.259
515825	SOCRATES 67	146886 .	094K	2008/jul/01	20080701	421.248
515826	SOCRATES 68	146886 .	094K	2008/jul/01	20080701	421.499
517407	TOAD 17	146886 .	094K	2008/jul/12	20080712	118.277

517410	TOAD 18	146886 .	094K	2008/jul/12	20080712	118.205
517636	DELANO 27	146886 .	094K	2008/jul/13	20080713	422.181
517637	DELANO 28	146886 .	094K	2008/jul/13	20080713	405.26
517639	DELANO 28	146886 .	094K	2008/jul/13	20080713	405.183
517877	LR2	146886 .	094K	2008/jul/17	20080717	405.195
517878	LR3	146886 .	094K	2008/jul/17	20080717	270.133
517882	LR6	146886 .	094K	2008/jul/17	20080717	422.31
517885	LR7	146886 .	094K	2008/jul/17	20080717	354.947
517886	LR8	146886 .	094K	2008/jul/17	20080717	422.541
517888	LR9	146886 .	094K	2008/jul/17	20080717	422.547
517890	LR10	146886 .	094K	2008/jul/17	20080717	422.555
517891	LR11	146886 .	094K	2008/jul/17	20080717	422.556
517892	LR12	146886 .	094K	2008/jul/17	20080717	422.77
517893	LR5	146886 .	094K	2008/jul/17	20080717	337.844
517894	LR13	146886 .	094K	2008/jul/17	20080717	372.052
517895	LR14	146886 .	094K	2008/jul/17	20080717	405.861
517898	LR15	146886 .	094K	2008/jul/17	20080717	405.854
517899	LR16	146886 .	094K	2008/jul/17	20080717	405.848
517900	LR17	146886 .	094K	2008/jul/17	20080717	405.892
517924	LR41	146886 .	094K	2008/jul/17	20080717	404.979
517925	LR42	146886 .	094K	2008/jul/17	20080717	404.98
517926	LR43	146886 .	094K	2008/jul/17	20080717	404.982
517927	LR44	146886 .	094K	2008/jul/17	20080717	404.982
517928	LR45	146886 .	094K	2008/jul/17	20080717	404.983
517929	LR46	146886 .	094K	2008/jul/17	20080717	404.984
517930	LR49	146886 .	094K	2008/jul/17	20080717	405.191
517931	LR47	146886 .	094K	2008/jul/17	20080717	404.988
517932	LR48	146886 .	094K	2008/jul/17	20080717	421.843
517901	LR18	200740 .	094K	2008/jul/17	20080717	355.343
517902	LR19	200740 .	094K	2008/jul/17	20080717	422.98
517903	LR20	200740 .	094K	2008/jul/17	20080717	422.98
517904	LR21	200740 .	094K	2008/jul/17	20080717	422.978
517905	LR22	200740 .	094K	2008/jul/17	20080717	422.975
517906	LR23	200740 .	094K	2008/jul/17	20080717	422.973
517907	LR24	200740 .	094K	2008/jul/17	20080717	406.126
517908	LR25	200740 .	094K	2008/jul/17	20080717	406.247
517910	LR27	200740 .	094K	2008/jul/17	20080717	406.276
517911	LR28	200740 .	094K	2008/jul/17	20080717	406.276
517912	LR29	200740 .	094K	2008/jul/17	20080717	406.277
517913	LR30	200740 .	094K	2008/jul/17	20080717	406.274
517915	LR32	200740 .	094K	2008/jul/17	20080717	423.429
517916	LR33	200740 .	094K	2008/jul/17	20080717	423.429
517917	LR34	200740 .	094K	2008/jul/17	20080717	423.429
517918	LR35	200740 .	094K	2008/jul/17	20080717	423.425
517919	LR36	200740 .	094K	2008/jul/17	20080717	423.678

517920	LR37	200740 .	094K	2008/jul/17	20080717	423.679
517921	LR38	200740 .	094K	2008/jul/17	20080717	423.678
517922	LR39	200740 .	094K	2008/jul/17	20080717	423.674
517923	LR40	200740 .	094K	2008/jul/17	20080717	406.945
537919	RR1	200740 .	094K	2008/jul/27	20080727	388.153
537920	RR2	200740 .	094K	2008/jul/27	20080727	236.402
537921	RR3	200740 .	094K	2008/jul/27	20080727	388.175
537922	RR4	200740 .	094K	2008/jul/27	20080727	421.937
537923	RR5	200740 .	094K	2008/jul/27	20080727	421.933
537925	RR6	200740 .	094K	2008/jul/27	20080727	421.932
537926	RR7	200740 .	094K	2008/jul/27	20080727	421.934
537927	RR8	200740 .	094K	2008/jul/27	20080727	421.738
537929	RR9	200740 .	094K	2008/jul/27	20080727	421.712
537931	RR10	200740 .	094K	2008/jul/27	20080727	421.721
537932	RR11	200740 .	094K	2008/jul/27	20080727	421.472
537933	RR12	200740 .	094K	2008/jul/27	20080727	421.705
537935	RR3	200740 .	094K	2008/jul/27	20080727	421.7
537936	RR14	200740 .	094K	2008/jul/27	20080727	421.932
537937	RR16	200740 .	094K	2008/jul/27	20080727	421.695
537940	RR18	200740 .	094K	2008/jul/27	20080727	421.695
537942	RR19	200740 .	094K	2008/jul/27	20080727	337.357
537944	RR20	200740 .	094K	2008/jul/27	20080727	404.026
537946	RR21	200740 .	094K	2008/jul/27	20080727	404.332
537949	RR22	200740 .	094K	2008/jul/27	20080727	320.306
537924	AB01	202640 .	094K	2008/jul/27	20080727	421.487
537928	AB02	202640 .	094K	2008/jul/27	20080727	421.463
537930	AB03	202640 .	094K	2008/jul/27	20080727	421.455
537934	AB04	202640 .	094K	2008/jul/27	20080727	303.8
537938	AB05	202640 .	094K	2008/jul/27	20080727	236.16
537941	AB06	202640 .	094K	2008/jul/27	20080727	403.725
537943	GRIZZ 1	202640 .	094K	2008/jul/27	20080727	424.721
537945	GRIZZ 2	202640 .	094K	2008/jul/27	20080727	424.716
537947	GRIZZ 3	202640 .	094K	2008/jul/27	20080727	424.713
537948	GRIZZ 4	202640 .	094K	2008/jul/27	20080727	424.71
537950	GRIZZ 5	202640 .	094K	2008/jul/27	20080727	424.727
537951	GRIZZ 6	202640 .	094K	2008/jul/27	20080727	424.947
537952	GRIZZ 7	202640 .	094K	2008/jul/27	20080727	424.931
537953	GRIZZ 8	202640 .	094K	2008/jul/27	20080727	424.935
537954	GRIZZ 9	202640 .	094K	2008/jul/27	20080727	424.926
537955	GRIZZ 10	202640 .	094K	2008/jul/27	20080727	407.904
538026	PQ01	200740 .	094K	2008/jul/28	20080728	421.236
538029	PQ02	200740 .	094K	2008/jul/28	20080728	421.222
538036	PQ03	200740 .	094K	2008/jul/28	20080728	420.355
538038	PQ04	200740 .	094K	2008/jul/28	20080728	420.354
538045	PQ05	200740 .	094K	2008/jul/28	20080728	386.932

538048	PQ06	200740 .	094K	2008/jul/28	20080728	403.804
538052	PQ07	200740 .	094K	2008/jul/28	20080728	202.02
538055	PQ08	200740 .	094K	2008/jul/28	20080728	420.353
538057	PQ09	200740 .	094K	2008/jul/28	20080728	403.802
538060	PQ10	200740 .	094K	2008/jul/28	20080728	403.329
538062	PQ11	200740 .	094K	2008/jul/28	20080728	403.325
538065	PQ12	200740 .	094K	2008/jul/28	20080728	403.323
538067	PQ13	200740 .	094K	2008/jul/28	20080728	403.323
538070	PQ14	200740 .	094K	2008/jul/28	20080728	352.506
538073	PQ15	200740 .	094K	2008/jul/28	20080728	419.633
538077	PQ16	200740 .	094K	2008/jul/28	20080728	352.329
538079	PQ17	200740 .	094K	2008/jul/28	20080728	385.831
538082	PQ18	200740 .	094K	2008/jul/28	20080728	402.599
538084	PQ19	200740 .	094K	2008/jul/28	20080728	402.937
538085	PQ20	200740 .	094K	2008/jul/28	20080728	402.937
538087	PQ21	200740 .	094K	2008/jul/28	20080728	402.937
538089	PQ22	200740 .	094K	2008/jul/28	20080728	402.936
538091	PQ23	200740 .	094K	2008/jul/28	20080728	386.36
538092	PQ24	200740 .	094K	2008/jul/28	20080728	403.156
538096	PQ25	200740 .	094K	2008/jul/28	20080728	402.601
538025	RR23	202640 .	094K	2008/jul/28	20080728	421.45
538028	RR24	202640 .	094K	2008/jul/28	20080728	421.446
538031	RR25	202640 .	094K	2008/jul/28	20080728	421.441
538033	RR26	202640 .	094K	2008/jul/28	20080728	84.288
538037	RR27	202640 .	094K	2008/jul/28	20080728	421.213
538039	RR28	202640 .	094K	2008/jul/28	20080728	421.205
538042	RR29	202640 .	094K	2008/jul/28	20080728	421.201
538043	RR30	202640 .	094K	2008/jul/28	20080728	421.196
538046	RR31	202640 .	094K	2008/jul/28	20080728	303.237
538047	RR32	202640 .	094K	2008/jul/28	20080728	420.957
538050	RR33	202640 .	094K	2008/jul/28	20080728	420.95
538053	RR34	202640 .	094K	2008/jul/28	20080728	420.944
538054	RR35	202640 .	094K	2008/jul/28	20080728	420.941
538056	RR36	202640 .	094K	2008/jul/28	20080728	336.75
538058	RR37	202640 .	094K	2008/jul/28	20080728	403.802
538061	RR38	202640 .	094K	2008/jul/28	20080728	403.802
538063	RR39	202640 .	094K	2008/jul/28	20080728	403.803
538064	RR40	202640 .	094K	2008/jul/28	20080728	403.805
538066	RR41	202640 .	094K	2008/jul/28	20080728	269.267
538069	RR42	202640 .	094K	2008/jul/28	20080728	336.465
538071	RR43	202640 .	094K	2008/jul/28	20080728	420.353
538072	RR44	202640 .	094K	2008/jul/28	20080728	420.354
538075	RR45	202640 .	094K	2008/jul/28	20080728	420.355
538076	RR46	202640 .	094K	2008/jul/28	20080728	420.358
538078	RR47	202640 .	094K	2008/jul/28	20080728	269.018

538080	RR48	202640 .	094K	2008/jul/28	20080728	403.325
538081	RR49	202640 .	094K	2008/jul/28	20080728	403.325
538083	RR50	202640 .	094K	2008/jul/28	20080728	403.329
538086	RR51	202640 .	094K	2008/jul/28	20080728	419.911
538088	RR52	202640 .	094K	2008/jul/28	20080728	419.908
538090	RR53	202640 .	094K	2008/jul/28	20080728	419.907
538093	RR54	202640 .	094K	2008/jul/28	20080728	419.905
538095	RR55	202640 .	094K	2008/jul/28	20080728	419.902
538097	RR56	202640 .	094K	2008/jul/28	20080728	402.991
538098	RR57	202640 .	094K	2008/jul/28	20080728	402.89
538099	RR58	202640 .	094K	2008/jul/28	20080728	402.603
538100	RR59	202640 .	094K	2008/jul/28	20080728	402.604
518973	GRIZZLY 65	146886 .	094K	2008/aug/12	20080812	406.601
518974	GRIZZLY 66	146886 .	094K	2008/aug/12	20080812	406.412
518975	GRIZZLY 67	146886 .	094K	2008/aug/12	20080812	423.337
518976	GRIZZLY 68	146886 .	094K	2008/aug/12	20080812	406.604
518977	GRIZZLY 69	146886 .	094K	2008/aug/12	20080812	406.7
518978	GRIZZLY 70	146886 .	094K	2008/aug/12	20080812	406.983
518979	GRIZZLY 71	146886 .	094K	2008/aug/12	20080812	407.268
518980	GRIZZLY 72	146886 .	094K	2008/aug/12	20080812	424.502
519444	Y01	200103 .	094K	2008/aug/28	20080828	337.272
519445	Y02	200103 .	094K	2008/aug/28	20080828	303.66
519446	Y03	200103 .	094K	2008/aug/28	20080828	404.991
519447	Y04	200103 .	094K	2008/aug/28	20080828	202.528
519448	Y05	200103 .	094K	2008/aug/28	20080828	405.054
519449	Y06	200103 .	094K	2008/aug/28	20080828	303.903
519450	Y07	200103 .	094K	2008/aug/28	20080828	405.42
519451	Y08	200103 .	094K	2008/aug/28	20080828	422.192
519452	Y09	200103 .	094K	2008/aug/28	20080828	253.436
519453	Y10	200103 .	094K	2008/aug/28	20080828	202.751
519454	Y11	200103 .	094K	2008/aug/28	20080828	405.715
519455	Y12	200103 .	094K	2008/aug/28	20080828	202.962
519456	Y13	200103 .	094K	2008/aug/28	20080828	304.289
519457	Y14	200103 .	094K	2008/aug/28	20080828	422.642
519458	Y15	200103 .	094K	2008/aug/28	20080828	304.354
539991	ANVIL01	202640 .	094K	2008/aug/28	20080828	408.128
539993	ANVIL02	202640 .	094K	2008/aug/28	20080828	408.121
539994	ANVIL03	202640 .	094K	2008/aug/28	20080828	204.058
539996	ANVIL04	202640 .	094K	2008/aug/28	20080828	408.094
539997	ANVIL05	202640 .	094K	2008/aug/28	20080828	408.284
539998	ANVIL06	202640 .	094K	2008/aug/28	20080828	408.282
539999	ANVIL07	202640 .	094K	2008/aug/28	20080828	408.281
540000	ANVIL08	202640 .	094K	2008/aug/28	20080828	408.423
540001	ANVIL09	202640 .	094K	2008/aug/28	20080828	136.141
540002	ANVIL10	202640 .	094K	2008/aug/28	20080828	306.251

508545	Grizzly 1	146886 .	094K	2008/sep/09	20080909	220.665
511143	GRIZZLY 6	146886 .	094K	2008/sep/09	20080909	407.61
511145	GRIZZLY 8	146886 .	094K	2008/sep/09	20080909	407.633
511146	GRIZZLY 9	146886 .	094K	2008/sep/09	20080909	424.838
511148	GRIZZLY 11	146886 .	094K	2008/sep/09	20080909	407.779
511150	GRIZZLY 12	146886 .	094K	2008/sep/09	20080909	407.873
520483	TOWER1	200103 .	094K	2008/sep/27	20080927	355.197
520485	TOWER2	200103 .	094K	2008/sep/27	20080927	423.104
520486	TOWER3	200103 .	094K	2008/sep/27	20080927	423.291
520487	TOWER4	200103 .	094K	2008/sep/27	20080927	406.523
520650	TOWER5	200103 .	094K	2008/sep/30	20080930	338.278
520651	TOWER6	200103 .	094K	2008/sep/30	20080930	338.437
520652	TOWER7	200103 .	094K	2008/sep/30	20080930	338.596
520653	TOWER8	200103 .	094K	2008/sep/30	20080930	338.755
520701	GS1	146887 .	094K	2008/oct/02	20081002	389.013
520702	GS2	146887 .	094K	2008/oct/02	20081002	338.414
520703	GS3	146887 .	094K	2008/oct/02	20081002	355.456
520704	GS4	146887 .	094K	2008/oct/02	20081002	355.58
520707	GS5	146887 .	094K	2008/oct/02	20081002	372.642
509549	Ed	146887 .	094K	2008/nov/23	20081123	425.068
501179		146886 .	094K	2009/jan/12	20090112	153.498
525256	GODOT01	200740 .	094K	2009/jan/13	20090113	101.87
525267	GODOT02	200740 .	094K	2009/jan/13	20090113	67.862
525433	TORO_SOUTH	200740 .	094K	2009/jan/14	20090114	407.638
525439	TORO_NORTH	200740 .	094K	2009/jan/14	20090114	203.591
504054	Talus	146887 .	094K	2009/jan/17	20090117	423.475
511144	GRIZZLY 7	146886 .	094K	2009/jan/20	20090120	339.543
511147	GRIZZLY 10	146886 .	094K	2009/jan/20	20090120	339.697
510811	MEDS 1	124708 .	094K	2009/jan/31	20090131	253.999
508444	Gataga 1	146886 .	094K	2009/mar/09	20090309	341.22
508445	Gataga 2	146886 .	094K	2009/mar/09	20090309	392.393
508447	Gataga 3	146886 .	094K	2009/mar/09	20090309	409.33
508449	Gataga 4	146886 .	094K	2009/mar/09	20090309	238.775
508450	Gataga 5	146886 .	094K	2009/mar/09	20090309	375.484
508451	Gataga 6	146886 .	094K	2009/mar/09	20090309	392.551
508452	Gataga 7	146886 .	094K	2009/mar/09	20090309	409.757
508454	Gataga 8	146886 .	094K	2009/mar/09	20090309	409.753
508455	Gataga 9	146886 .	094K	2009/mar/09	20090309	409.894
508456	Gataga 10	146886 .	094K	2009/mar/09	20090309	410.035
508457	Gataga 11	146886 .	094K	2009/mar/09	20090309	341.667
508459	Gataga 12	146886 .	094K	2009/mar/09	20090309	410.178
508460	Gataga 13	146886 .	094K	2009/mar/09	20090309	273.447
508462	Gataga 14	146886 .	094K	2009/mar/09	20090309	341.914
508464	Gataga 15	146886 .	094K	2009/mar/09	20090309	205.205
508467	Gataga 16	146886 .	094K	2009/mar/09	20090309	323.945

508469	Gataga 17	146886	094K	2009/mar/09	20090309	409.189
508470	Gataga 18	146886	094K	2009/mar/09	20090309	255.651
508471	Gataga 19	146886	094K	2009/mar/09	20090309	409.02
508479	Socrates 1	146886	094K	2009/mar/09	20090309	420.076
508482	Socrates 2	146886	094K	2009/mar/09	20090309	403.3
508483	Socrates 2	146886	094K	2009/mar/09	20090309	353.034
508484	Socrates 4	146886	094K	2009/mar/09	20090309	403.374
508485	Socrates 5	146886	094K	2009/mar/09	20090309	336.284
508486	Socrates 6	146886	094K	2009/mar/09	20090309	403.539
508487	Socrates 7	146886	094K	2009/mar/09	20090309	420.576
508488	Socrates 8	146886	094K	2009/mar/09	20090309	420.577
508489	Socrates 9	146886	094K	2009/mar/09	20090309	420.573
508490	Socrates 10	146886	094K	2009/mar/09	20090309	420.569
508492	Socrates 11	146886	094K	2009/mar/09	20090309	336.57
508494	Socrates 12	146886	094K	2009/mar/09	20090309	420.856
508497	Socrates 13	146886	094K	2009/mar/09	20090309	420.861
508504	Socrates 14	146886	094K	2009/mar/09	20090309	420.861
508506	Socrates 15	146886	094K	2009/mar/09	20090309	420.86
508507	Socrates 16	146886	094K	2009/mar/09	20090309	404.242
508508	Socrates 17	146886	094K	2009/mar/09	20090309	336.876
508509	Socrates 18	146886	094K	2009/mar/09	20090309	404.371
508510	Socrates 19	146886	094K	2009/mar/09	20090309	404.518
508511	Delano 1	146886	094K	2009/mar/09	20090309	406.178
508512	Delano 2	146886	094K	2009/mar/09	20090309	338.339
508515	Delano 3	146886	094K	2009/mar/09	20090309	406.042
508521	Delano 4	146886	094K	2009/mar/09	20090309	406.165
508527	Delano 5	146886	094K	2009/mar/09	20090309	406.021
508535	Delano 6	146886	094K	2009/mar/09	20090309	405.873
508537	Delano 7	146886	094K	2009/mar/09	20090309	405.729
508540	Delano 8	146886	094K	2009/mar/09	20090309	405.654
508550	Grizzly 2	146886	094K	2009/mar/09	20090309	424.21
508554	Delano 3	146886	094K	2009/mar/09	20090309	423.961
508557	Grizzly 4	146886	094K	2009/mar/09	20090309	406.982
508560	Grizzly 5	146886	094K	2009/mar/09	20090309	423.724
508597	Dieppe 1	146886	094K	2009/mar/10	20090310	337.139
508598	Dieppe 2	146886	094K	2009/mar/10	20090310	337.143
508599	Dieppe 3	146886	094K	2009/mar/10	20090310	337.147
508600	Dieppe 4	146886	094K	2009/mar/10	20090310	421.65
508602	Dieppe 6	146886	094K	2009/mar/10	20090310	421.656
508603	Dieppe 7	146886	094K	2009/mar/10	20090310	421.66
508605	Dieppe 8	146886	094K	2009/mar/10	20090310	269.851
508606	Dieppe 9	146886	094K	2009/mar/10	20090310	405.02
508607	Dieppe 10	146886	094K	2009/mar/10	20090310	405.021
508609	Dieppe 11	146886	094K	2009/mar/10	20090310	405.021
508617	Dieppe 12	146886	094K	2009/mar/10	20090310	421.892

508621	Dieppe 13	146886 .	094K	2009/mar/10	20090310	404.948
508623	Dieppe 14	146886 .	094K	2009/mar/10	20090310	405.051
508627	Dieppe 15	146886 .	094K	2009/mar/10	20090310	405.052
508629	Dieppe 16	146886 .	094K	2009/mar/10	20090310	422.263
508633	Dieppe 17	146886 .	094K	2009/mar/10	20090310	422.097
508634	Dieppe 17	146886 .	094K	2009/mar/10	20090310	422.561
508636	Dieppe 18	146886 .	094K	2009/mar/10	20090310	422.63
508639	Dieppe 18	146886 .	094K	2009/mar/10	20090310	422.629
508642	Dieppe 20	146886 .	094K	2009/mar/10	20090310	405.27
508644	Dieppe 21	146886 .	094K	2009/mar/10	20090310	388.452
508645	Dieppe 22	146886 .	094K	2009/mar/10	20090310	422.467
508647	Dieppe 23	146886 .	094K	2009/mar/10	20090310	405.56
508651	Dieppe 24	146886 .	094K	2009/mar/10	20090310	422.486
508656	Dieppe 25	146886 .	094K	2009/mar/10	20090310	338.186
508659	Dieppe 26	146886 .	094K	2009/mar/10	20090310	422.736
508666	Dieppe 27	146886 .	094K	2009/mar/10	20090310	422.665
508670	Dieppe 28	146886 .	094K	2009/mar/10	20090310	304.394
508671	Dieppe 29	146886 .	094K	2009/mar/10	20090310	355.231
508675	Dieppe 30	146886 .	094K	2009/mar/10	20090310	405.998
508685	Dieppe 31	146886 .	094K	2009/mar/10	20090310	372.18
508686	Dieppe 32	146886 .	094K	2009/mar/10	20090310	423.009
508687	Dieppe 33	146886 .	094K	2009/mar/10	20090310	406.271
508688	Dieppe 34	146886 .	094K	2009/mar/10	20090310	355.674
508689	Dieppe 35	146886 .	094K	2009/mar/10	20090310	338.66
508690	Dieppe 36	146886 .	094K	2009/mar/10	20090310	338.523
508691	Dieppe 36	146886 .	094K	2009/mar/10	20090310	406.415
508692	Dieppe 38	146886 .	094K	2009/mar/10	20090310	406.672
508693	Dieppe 39	146886 .	094K	2009/mar/10	20090310	305.023
508694	Dieppe 40	146886 .	094K	2009/mar/10	20090310	372.987
508696	Dieppe 41	146886 .	094K	2009/mar/10	20090310	372.206
508697	Dieppe 42	146886 .	094K	2009/mar/10	20090310	406.241
508699	Dieppe 43	146886 .	094K	2009/mar/10	20090310	406.385
508704	Dieppe 44	146886 .	094K	2009/mar/10	20090310	406.124
508771	Delano 9	146886 .	094K	2009/mar/11	20090311	405.508
509141	Gataga 20	146886 .	094K	2009/mar/17	20090317	410.227
509544	Goat	146887 .	094K	2009/mar/23	20090323	422.436
511151	GRIZZLY 13	146886 .	094K	2009/apr/20	20090420	424.864
511153	GRIZZLY 13	146886 .	094K	2009/apr/20	20090420	425.069
511155	GRIZZLY 14	146886 .	094K	2009/apr/20	20090420	425.065
511157	GRIZZLY 15	146886 .	094K	2009/apr/20	20090420	425.078
511159	GRIZZLY 16	146886 .	094K	2009/apr/20	20090420	425.074
511160	GRIZZLY 16	146886 .	094K	2009/apr/20	20090420	425.224
511162	GRIZZLY 17	146886 .	094K	2009/apr/20	20090420	425.323
511165	GRIZZLY 18	146886 .	094K	2009/apr/20	20090420	425.323
511188	GRIZZLY 19	146886 .	094K	2009/apr/20	20090420	425.324

511189	GRIZZLY 20	146886 .	094K	2009/apr/20	20090420	425.319
511191	GRIZZLY 21	146886 .	094K	2009/apr/20	20090420	425.282
511192	GRIZZLY 22	146886 .	094K	2009/apr/20	20090420	425.573
511193	GRIZZLY 23	146886 .	094K	2009/apr/20	20090420	425.575
511195	GRIZZLY 24	146886 .	094K	2009/apr/20	20090420	425.579
511198	GRIZZLY 25	146886 .	094K	2009/apr/20	20090420	425.58
511200	GRIZZLY 26	146886 .	094K	2009/apr/20	20090420	357.475
511201	GRIZZLY 27	146886 .	094K	2009/apr/20	20090420	425.54
511203	GRIZZLY 28	146886 .	094K	2009/apr/20	20090420	425.576
511205	GRIZZLY 29	146886 .	094K	2009/apr/20	20090420	340.464
517875	LR1	146886 .	094K	2009/jul/17	20090717	405.186
517879	LR4	146886 .	094K	2009/jul/17	20090717	422.298
517876	TR1	200740 .	094K	2009/jul/17	20090717	406.942
517880	TR2	200740 .	094K	2009/jul/17	20090717	406.943
517881	TR3	200740 .	094K	2009/jul/17	20090717	406.945
517909	LR26	200740 .	094K	2009/jul/17	20090717	406.298
517914	LR31	200740 .	094K	2009/jul/17	20090717	372.664
510008		124708 .	094K	2009/jul/23	20090723	591.197
510739	KEY1	124708 .	094K	2009/jul/23	20090723	84.474
510740	KEY2	124708 .	094K	2009/jul/23	20090723	84.476
510741	KEY3	124708 .	094K	2009/jul/23	20090723	152.056
510808	KEY X	124708 .	094K	2009/jul/23	20090723	16.897
510809	KEY Y	124708 .	094K	2009/jul/23	20090723	16.891
510810	NUCO 1	124708 .	094K	2009/jul/23	20090723	16.881
510255		124708 .	094K	2009/aug/30	20090830	270.179
519544	KEY	124708 .	094K	2009/aug/31	20090831	422.374
519545	KEY 1	124708 .	094K	2009/aug/31	20090831	422.15
519546	KEY 3	124708 .	094K	2009/aug/31	20090831	219.48
504085	Carmen	146887 .	094K	2009/sep/17	20090917	405.558
501321	Lana	124708 .	094K	2009/dec/31	20091231	101.627
501446	Meg	124708 .	094K	2009/dec/31	20091231	236.91
501482	Hunter	124708 .	094K	2009/dec/31	20091231	406.726
501523	Sara	124708 .	094K	2009/dec/31	20091231	287.368
501534	Missy	124708 .	094K	2009/dec/31	20091231	406.025
501416	Angel	124708 .	094K	2010/jan/12	20100112	338.184
504049	Lucky Lady	146887 .	094K	2010/jan/17	20100117	406.228
504060	Peak	146887 .	094K	2010/jan/17	20100117	422.084
504064	Peak South	146887 .	094K	2010/jan/17	20100117	422.362
504869		146886 .	094K	2010/may/1 2	20100512	746.834
501462	Sox	124708 .	094K	2010/dec/31	20101231	253.727
501497	Taya	124708 .	094K	2010/dec/31	20101231	202.688
501161		146886 .	094K	2011/jan/12	20110112	153.57
501201		146886 .	094K	2016/jan/12	20160112	153.709

APPENDIX B
MUSKWA-KECHIKA SMZ

**LINKS TO INFORMATION ON THE
MUSKWA-KECHIKA SPECIAL MANAGEMENT ZONE**

Government and separate advisory board

<http://srmwww.gov.bc.ca/rmd/lrmp/mk>

http://www.qp.gov.bc.ca/statreg/stat/M/98038_01.htm

<http://www.em.gov.bc.ca/subwebs/oilandgas/ptp/MKMA.htm>

http://www.qp.gov.bc.ca/statreg/reg/M/53_2002.htm

<http://www.dir.gov.bc.ca/gtds.cgi?show=Branch&organizationCode=SRM&organizationUnitCode=MK>

Canadian Parks and Wilderness Society

<http://www.cpaws.org/northernrockies>

The Muskwa-Kechika Management Area

http://www.wilderness.net/library/documents/IJWDec03_ShultisRutledge.pdf

APPENDIX C

ASSAYS

Magnum Brecciated Veined Zone and Missy Veins Assays (Churchill Mine)



ACME ANALYTICAL LABORATORIES LTD.
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Phone (604) 253-3158 Fax (604) 253-1716

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Client: Action Mineral
1255 W. Pender St.
Vancouver BC V6E 2V1 Canada

Submitted By: George Coetsee
Receiving Lab: Acme Analytical Laboratories (Vancouver) Ltd.
Received: October 04, 2007
Report Date: February 05, 2008
Page: 1 of 3

CERTIFICATE OF ANALYSIS

VAN07001993.2

CLIENT JOB INFORMATION

Project: Trident
Shipment ID:
P.O. Number:
Number of Samples: 99

SAMPLE DISPOSAL

SAMPLE PREPARATION AND ANALYTICAL PROCEDURES

Method Code	Number of Samples	Code Description	Test Wgt (g)	Report Status
R150	53	Crush, split and pulverize rock to 150 mesh		
7TX	99	4 Acid digestion ICP-ES/ICP-MS analysis	0.5	Completed
3A	18	Ignite samples, acid digest, Au by ICP-MS	15	Completed

ADDITIONAL COMMENTS

Version 2 to include Au by 3A analysis

Acme does not accept responsibility for samples left at the laboratory after 90 days without prior written instructions for sample storage or return.

Invoice To: Action Mineral
1255 W. Pender St.
Vancouver BC V6E 2V1
Canada

CC:



This report is prepared as a preliminary report and final report will be issued and dated prior to the date of this certificate. Significant findings that appear on preliminary reports are finalized and should be used for reference only. All reports are considered the confidential property of the client. Acme assumes the liability for analytical results only.



ACME ANALYTICAL LABORATORIES LTD.
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Client: **Action Mineral**
1255 W. Pender St.
Vancouver BC V6E 2V1 Canada

Project: **Trident**
Report Date: **February 05, 2008**

Page: 2 of 3 Part 1

CERTIFICATE OF ANALYSIS

VAN07001993.2

Method	Analyte	Unit	MDL	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
				Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La
				ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
465101	Rock			0.6	19.3	1.4	<5	<0.5	5.5	7	512	3.10	9	1.7	4.0	111	<0.5	0.8	0.8	28	10.33	0.05	13.4
465102	Rock			<0.5	52.7	1.4	<5	<0.5	4.4	15	303	1.00	21	1.1	2.7	87	<0.5	1.1	0.6	26	5.97	0.03	10.5
465103	Rock			0.5	36.9	1.4	<5	<0.5	4.1	4	404	1.98	<5	3.1	8.8	72	<0.5	0.7	<0.5	62	9.10	0.03	15.2
465104	Rock			<0.5	1376	0.8	<5	<0.5	5.9	5	374	2.10	7	1.3	3.6	62	<0.5	1.3	<0.5	23	5.12	0.02	6.8
465105	Rock			0.7	216.3	1.2	<5	<0.5	6.9	7	473	2.38	12	2.4	6.5	88	<0.5	1.4	<0.5	48	7.18	0.04	10.9
465107	Rock			0.6	115.8	0.7	<5	<0.5	3.8	4	514	2.41	<5	1.1	3.0	80	<0.5	1.0	<0.5	28	8.03	0.02	3.9
465108	Rock			0.8	1494	7.7	<5	<0.5	24.8	9	283	1.41	17	1.5	2.8	64	<0.5	3.3	<0.5	24	3.99	0.04	4.9
465109	Rock			<0.5	135397	5.4	36	6.4	45.2	20	1066	13.63	116	<0.5	<0.5	55	<0.5	13.4	13.5	13	6.07	<0.01	2.0
465110	Rock			0.6	203639	13.3	94	11.7	129.3	84	497	19.73	254	<0.5	<0.5	60	1.2	32.1	96.6	<10	4.25	0.02	1.0
465111	Rock			<0.5	71123	4.7	34	4.6	34.2	26	1375	8.83	111	0.9	1.7	131	<0.5	4.3	24.0	<10	10.44	<0.01	2.4
465112	Rock			0.9	21667	1.8	18	1.8	22.1	16	1751	4.65	16	0.6	1.2	72	<0.5	2.6	5.0	<10	11.01	0.03	9.1
465113	Rock			<0.5	5967	<0.5	9	<0.5	9.4	6	559	2.47	9	0.6	1.2	59	<0.5	0.9	2.2	<10	5.84	0.02	2.9
465114	Rock			0.6	5066	1.1	6	<0.5	15.5	8	346	1.77	<5	<0.5	1.0	34	<0.5	1.1	1.3	<10	2.70	0.04	4.2
465115	Rock			<0.5	13958	0.6	<5	<0.5	25.0	18	632	1.66	5	<0.5	1.1	63	<0.5	1.2	0.6	<10	4.30	0.01	1.4
465116	Rock			0.5	785.0	0.8	<5	<0.5	1.5	2	1482	2.32	<5	0.7	1.7	150	<0.5	<0.5	<0.5	<10	10.72	0.04	8.9
465117	Rock			<0.5	775.1	<0.5	<5	<0.5	2.3	2	782	1.81	<5	1.7	3.8	111	<0.5	0.8	<0.5	27	7.49	0.03	3.9
465118	Rock			<0.5	40004	6.3	21	2.7	18.5	17	358	6.05	28	<0.5	<0.5	26	<0.5	4.8	5.5	<10	0.77	<0.01	2.1
465119	Rock			0.6	7397	2.2	9	<0.5	15.6	10	309	2.39	7	<0.5	0.5	103	<0.5	3.5	1.0	<10	4.31	0.11	2.3
465119A	Rock Pulp			71.5	4414	1916	6709	24.0	50.7	61	882	20.08	575	3.5	6.2	138	30.7	116.4	21.5	60	1.74	0.04	14.0
465119B	Rock Pulp			5.1	46.9	3.5	55	<0.5	30.4	11	772	3.98	<5	0.9	2.3	262	<0.5	1.1	<0.5	112	2.50	0.06	10.4
465120	Rock			0.6	6278	1.9	7	<0.5	6.3	4	177	1.74	42	1.2	2.3	29	<0.5	1.1	1.0	23	1.94	0.06	5.3
465121	Rock			0.7	9928	2.6	8	2.0	6.3	3	55	3.27	7	<0.5	<0.5	15	<0.5	1.6	1.1	<10	0.32	0.04	1.3
465122	Rock			<0.5	311.8	0.7	<5	<0.5	3.2	2	922	2.39	<5	1.4	3.8	141	<0.5	0.7	<0.5	31	10.29	0.04	8.2
465123	Rock			0.8	65347	8.5	47	6.0	24.8	26	193	8.58	<5	1.4	2.2	37	0.7	2.4	1.1	12	2.24	0.32	4.2
465124	Rock			0.5	15101	11.0	15	2.9	27.8	5	354	2.87	7	1.6	3.4	48	<0.5	2.5	<0.5	22	3.70	0.15	8.7
465125	Rock			<0.5	68591	7.9	14	2.9	10.6	7	65	6.53	9	1.2	2.5	17	<0.5	3.1	0.7	35	0.80	0.09	2.8
465126	Rock			2.0	5508	14.6	21	<0.5	10.2	10	167	1.72	14	0.9	2.4	20	<0.5	3.0	0.8	40	0.76	0.03	12.4
465127	Rock			2.4	23362	26.9	58	<0.5	16.7	24	383	3.40	78	1.8	4.7	58	<0.5	11.2	3.3	55	2.62	0.04	15.0
465128	Rock			1.1	1203	7.6	63	<0.5	11.6	11	386	1.45	16	1.3	3.7	84	<0.5	1.1	1.0	54	3.18	0.03	15.0
465129	Rock			1.8	6750	8.4	25	<0.5	53.8	34	308	1.53	132	1.1	2.3	26	<0.5	6.6	3.2	29	1.11	0.03	8.4

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval, preliminary reports are unsigned and should be used for reference only.



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Vancouver BC V6E 2V1 Canada

Project: Trident
Report Date: February 05, 2008

Page: 2 of 3 Part 2

CERTIFICATE OF ANALYSIS

VAN07001993.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	U	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
465 101	Rock	9	6.31	59	0.095	2.36	0.05	1.18	<0.5	40.5	31	15	11.8	3.4	<0.5	<5	5	21.1	<0.5	39.3	1.3
465 102	Rock	11	3.32	50	0.062	1.72	0.09	0.85	<0.5	21.9	23	10	8.2	2.9	<0.5	<5	4	23.4	<0.5	25.6	0.7
465 103	Rock	22	4.89	100	0.199	4.42	0.02	1.62	0.8	76.6	33	23	17.5	6.7	<0.5	<5	6	8.5	<0.5	47.2	2.3
465 104	Rock	11	2.78	47	0.073	2.01	0.03	1.10	<0.5	31.3	17	23	7.9	3.0	<0.5	<5	4	17.7	<0.5	27.2	1.1
465 105	Rock	15	3.81	61	0.140	3.63	0.02	1.75	<0.5	64.9	27	29	12.5	5.4	<0.5	<5	6	15.2	<0.5	41.3	2.1
465 107	Rock	10	4.47	32	0.067	1.66	0.03	0.93	<0.5	31.5	9	14	11.4	2.7	<0.5	<5	5	17.0	<0.5	18.8	1.0
465 108	Rock	10	2.09	36	0.068	1.71	0.03	0.92	<0.5	26.4	12	57	6.7	2.4	<0.5	<5	7	29.0	<0.5	24.0	1.0
465 109	Rock	4	3.11	14	0.007	0.24	0.02	0.11	<0.5	4.9	<5	445	6.4	<0.5	<0.5	<5	8	8.0	6.7	5.1	<0.5
465 110	Rock	4	2.38	8	0.005	0.19	0.01	0.06	<0.5	3.4	<5	72.0	4.4	<0.5	<0.5	<5	2	5.2	10.7	<0.5	<0.5
465 111	Rock	9	6.17	15	0.033	0.70	0.01	0.31	<0.5	14.0	6	262	9.4	1.4	<0.5	<5	6	8.1	4.2	4.8	0.5
465 112	Rock	10	6.21	19	0.016	0.81	0.02	0.18	<0.5	8.6	25	95	8.0	0.7	<0.5	<5	9	19.8	1.3	4.1	<0.5
465 113	Rock	10	3.80	29	0.018	0.84	0.03	0.12	<0.5	11.7	8	28	6.5	0.6	<0.5	<5	11	31.5	<0.5	1.5	1.4
465 114	Rock	15	2.13	29	0.016	0.91	0.04	0.16	<0.5	10.4	11	36	3.6	<0.5	<0.5	<5	3	37.4	<0.5	2.6	<0.5
465 115	Rock	13	2.56	130	0.016	0.72	0.04	0.26	<0.5	8.3	<5	19	5.4	0.7	<0.5	<5	12	25.8	<0.5	7.0	<0.5
465 116	Rock	10	6.08	70	0.031	1.23	0.01	0.50	<0.5	14.9	20	11	8.6	1.1	<0.5	<5	2	22.4	<0.5	14.9	0.5
465 117	Rock	16	3.88	335	0.086	2.27	0.03	1.25	<0.5	37.6	9	13	8.8	2.2	<0.5	<5	5	15.6	<0.5	26.2	1.3
465 118	Rock	8	0.45	571	0.008	0.43	0.04	0.19	<0.5	3.5	7	56.1	2.9	<0.5	<0.5	<5	5	28.8	2.0	2.9	<0.5
465 119	Rock	8	2.08	1662	0.010	0.72	0.04	0.33	<0.5	4.9	6	53	7.9	<0.5	<0.5	<5	12	24.4	0.6	8.2	<0.5
465 119A	Rock Pulp	96	0.86	557	0.114	3.93	0.57	1.59	7.0	15.9	24	63	7.3	1.5	<0.5	<5	4	7.4	16.6	61.7	0.6
465 119B	Rock Pulp	39	1.23	504	0.361	7.12	2.78	0.94	<0.5	38.2	21	20	14.5	4.4	<0.5	<5	13	10.0	<0.5	24.6	1.4
465 120	Rock	12	1.07	163	0.047	1.73	0.04	0.92	<0.5	26.4	14	103	5.3	1.6	<0.5	<5	7	41.5	<0.5	30.7	0.7
465 121	Rock	18	0.26	108	0.007	0.45	0.05	0.17	<0.5	3.4	<5	20.6	1.8	<0.5	<0.5	<5	1	45.1	0.8	4.6	<0.5
465 122	Rock	17	5.19	85	0.080	2.39	0.02	1.32	<0.5	40.3	23	28	12.5	3.6	<0.5	<5	8	18.4	<0.5	32.5	1.2
465 123	Rock	10	1.04	120	0.049	1.58	0.03	0.84	<0.5	24.2	12	71.6	8.4	2.5	<0.5	<5	3	24.5	5.1	22.8	0.9
465 124	Rock	17	1.94	83	0.060	1.87	0.03	0.99	<0.5	26.3	18	18.0	8.8	3.9	<0.5	<5	6	13.9	1.8	25.5	1.0
465 125	Rock	19	0.53	133	0.048	1.53	0.03	0.80	<0.5	29.8	7	42.4	5.9	2.4	<0.5	<5	3	26.5	4.6	28.7	0.7
465 126	Rock	16	0.44	138	0.047	1.63	0.06	0.80	<0.5	23.6	29	28	4.0	2.4	<0.5	<5	3	16.4	<0.5	29.6	0.7
465 127	Rock	21	0.86	219	0.083	2.70	0.03	1.31	0.5	47.5	41	5.6	8.7	4.1	<0.5	<5	6	24.5	0.7	56.9	1.6
465 128	Rock	14	0.71	173	0.077	2.43	0.03	1.20	<0.5	35.4	40	20	9.5	3.8	<0.5	<5	4	22.0	<0.5	49.2	1.6
465 129	Rock	16	0.42	143	0.053	1.72	0.04	0.89	<0.5	35.0	23	10.4	6.5	2.3	<0.5	<5	5	16.3	<0.5	36.2	0.7

This report represents all analytical preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval; preliminary reports are undated and should be used for reference only.

CERTIFICATE OF ANALYSIS

VAN07001993.2

	Method	3A
	Analyte	Au
	Unit	ppb
	MDL	0.5
465 101	Rock	N.A.
465 102	Rock	N.A.
465 103	Rock	N.A.
465 104	Rock	N.A.
465 105	Rock	N.A.
465 107	Rock	N.A.
465 108	Rock	N.A.
465 109	Rock	77.0
465 110	Rock	391.2
465 111	Rock	27.2
465 112	Rock	N.A.
465 113	Rock	N.A.
465 114	Rock	N.A.
465 115	Rock	N.A.
465 116	Rock	N.A.
465 117	Rock	N.A.
465 118	Rock	N.A.
465 119	Rock	N.A.
465 119A	Rock Pulp	N.A.
465 119B	Rock Pulp	N.A.
465 120	Rock	15.7
465 121	Rock	N.A.
465 122	Rock	N.A.
465 123	Rock	N.A.
465 124	Rock	N.A.
465 125	Rock	N.A.
465 126	Rock	N.A.
465 127	Rock	46.9
465 128	Rock	N.A.
465 129	Rock	18.9



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Project: Trident
Report Date: February 06, 2008

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CERTIFICATE OF ANALYSIS

VAN07001993.2

Method	SA	
Analyte	µg	
Unit	ppb	
MDL	0.5	
465130	Rock	N.A.
465131	Rock	46.2
465132	Rock	N.A.
465133	Rock	N.A.
465134	Rock	N.A.
465135	Rock	N.A.
465136	Rock	16.5
465137	Rock	N.A.
465200	Rock Pulp	N.A.
465200B	Rock Pulp	N.A.
465201	Rock	19.8
465202	Rock	36.6
465203	Rock	63.4
465204	Rock	34.1
465205	Rock	13.0
465206	Rock	9.3
465207	Rock	16.9
465208	Rock	13.1
465209	Rock	4.2
465210	Rock	N.A.
465210A	Rock Pulp	N.A.
465210B	Rock Pulp	N.A.
465211	Rock	N.A.
465212	Rock	N.A.
465213	Rock	27.3
465214	Rock	N.A.
465215	Rock	N.A.
465216	Rock	N.A.
465217	Rock	N.A.

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates that approval; preliminary reports are unsigned and should be used for reference only.

QUALITY CONTROL REPORT

VAN07001993.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
Analyte	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	
Unit	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
MDL	0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5	
Pulp Duplicates																					
465130	Rock	1.0	1205	7.0	29	<0.5	70.3	47	154	1.59	62	1.6	5.1	29	<0.5	2.4	2.1	57	1.09	0.04	11.4
REP 465130	QC	1.6	1233	11.4	44	<0.5	68.4	49	161	1.60	63	1.6	4.9	30	<0.5	2.9	2.2	56	1.11	0.03	11.9
465201	Rock	<0.5	3298	4.6	<5	<0.5	26.2	22	543	1.75	50	1.2	2.5	82	<0.5	0.8	1.1	15	7.56	0.03	10.8
REP 465201	QC																				
465211	Rock	<0.5	35961	6.5	25	1.3	21.8	6	753	5.16	6	1.4	3.7	82	<0.5	1.3	1.0	27	9.70	0.02	9.7
REP 465211	QC	<0.5	37416	6.9	22	1.3	22.4	6	768	5.18	6	1.4	3.8	82	<0.5	1.4	1.2	22	9.94	0.02	9.6
Reference Materials																					
STD OXD57	Standard																				
STD OXD57	Standard																				
STD SF-3T	Standard	314.6	7762	9363	11120	52.2	3518	183	4226	8.25	41	3.9	4.5	445	49.0	9.6	4.7	118	4.07	0.06	18.1
STD SF-3T	Standard	315.1	7795	9333	11055	52.1	3517	182	4203	8.23	45	4.0	4.7	444	48.8	11.0	4.8	119	4.05	0.06	17.9
STD SF-3T	Standard	316.0	7646	8723	10885	50.9	3483	191	4202	8.04	41	3.4	3.9	415	47.2	8.6	3.7	135	4.01	0.06	17.0
STD SF-3T	Standard	310.3	7542	8394	10665	46.3	3511	190	3940	7.57	41	3.5	4.0	377	47.8	8.1	3.8	135	3.92	0.06	17.7
STD SF-3T	Standard	326.1	7820	9610	11923	59.4	3681	198	4377	8.46	48	4.0	4.7	437	53.4	11.3	4.7	139	4.13	0.06	17.9
STD SF-3T	Standard	329.7	7883	9449	11967	52.5	3627	195	4266	8.34	45	3.9	4.6	430	51.0	10.9	4.6	139	4.13	0.06	17.9
STD SF-3T	Standard	319.0	8108	9394	11068	53.6	3530	183	4247	8.35	40	4.0	4.9	438	49.8	10.8	4.7	135	4.10	0.06	17.9
STD SF-3T	Standard	314.8	8013	9378	11185	53.6	3520	184	4287	8.40	45	3.9	4.5	435	50.8	9.4	4.6	135	4.14	0.06	17.5
STD SF-3T	Standard	323.6	7713	9397	10896	53.7	3541	183	4290	8.10	43	4.1	4.7	406	49.8	10.9	4.7	137	4.08	0.06	17.8
STD SF-3T	Standard	318.6	7719	9960	10763	53.2	3534	183	4320	8.08	42	3.9	4.7	434	50.4	10.5	4.7	137	4.06	0.06	17.8
STD SF-3T Expected		320	7723	9610	10672	52	3500	181	4320	8.33	40	4	4.7	440	47.5	11.1	4.8	143	4.1	0.06	17
STD OXD57 Expected																					
BLK	Blank	<0.5	<0.5	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank	<0.5	<0.5	11.4	<5	<0.5	4.9	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank	<0.5	<0.5	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank	<0.5	5.9	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank	<0.5	6.7	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank																				
Prep Wash																					

QUALITY CONTROL REPORT

VAN07001993.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	So	U	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
Pulp Duplicates																					
465130	Rock	16	0.75	359	0.098	2.78	0.04	1.56	<0.5	46.9	29	5.9	7.7	3.4	<0.5	<5	7	28.1	<0.5	57.0	1.6
REP 465130	QC	20	0.76	356	0.102	2.85	0.03	1.63	<0.5	46.2	30	5.3	7.8	4.3	<0.5	<5	7	25.5	<0.5	58.9	1.6
465201	Rock	105	4.09	46	0.051	1.51	0.02	0.86	<0.5	24.8	24	2.6	10.3	1.9	<0.5	<5	5	8.6	0.5	34.4	1.0
REP 465201	QC																				
465211	Rock	28	5.45	69	0.087	2.36	0.04	1.25	<0.5	36.3	23	19.7	13.4	3.7	<0.5	<5	3	12.5	3.3	53.4	1.1
REP 465211	QC	28	5.44	68	0.088	2.37	0.03	1.28	<0.5	37.6	23	20.9	13.4	3.8	<0.5	<5	4	12.0	3.2	51.7	1.3
Reference Materials																					
STD OXD57	Standard																				
STD OXD57	Standard																				
STD SF-3T	Standard	168	4.63	608	0.191	5.44	2.09	2.51	3.8	11.5	40	5.9	10.4	12.3	0.7	<5	7	20.3	4.2	89.8	<0.5
STD SF-3T	Standard	161	4.64	547	0.191	5.46	2.09	2.50	4.2	11.2	41	5.6	10.5	11.0	0.9	<5	7	20.5	4.2	88.3	<0.5
STD SF-3T	Standard	199	4.59	483	0.191	5.41	2.07	2.45	4.1	11.2	40	5.7	10.7	12.4	<0.5	<5	7	22.7	3.7	89.3	0.6
STD SF-3T	Standard	197	4.51	483	0.186	5.25	2.01	2.41	4.1	10.8	39	5.8	10.3	12.5	<0.5	<5	7	20.3	3.6	86.0	0.6
STD SF-3T	Standard	212	4.76	491	0.199	5.52	2.18	2.59	4.5	11.9	48	6.0	10.9	14.2	<0.5	<5	7	23.1	4.4	92.1	0.7
STD SF-3T	Standard	206	4.70	513	0.197	5.51	2.14	2.59	4.6	13.1	45	7.3	10.7	15.2	0.5	<5	7	20.8	4.2	91.2	0.6
STD SF-3T	Standard	206	4.73	481	0.195	5.50	2.09	2.50	4.3	13.5	42	5.9	10.9	14.4	<0.5	<5	7	20.5	4.1	92.2	0.6
STD SF-3T	Standard	199	4.76	495	0.195	5.56	2.11	2.47	4.5	13.8	40	6.0	10.8	14.4	<0.5	<5	7	20.6	4.2	92.6	<0.5
STD SF-3T	Standard	169	4.64	549	0.202	5.48	2.10	2.52	4.0	13.5	40	6.1	10.4	15.1	<0.5	<5	7	22.6	3.6	89.6	0.5
STD SF-3T	Standard	169	4.66	545	0.195	5.49	2.08	2.48	4.4	14.4	40	7.3	11.0	12.9	0.5	<5	6	22.7	3.6	88.3	0.6
STD SF-3T Expected		207.4	4.67	508	0.19	5.43	2.05	2.47	4.3	14	38	5.8	11.5	15.1	0.9	0	7	19.1	3.5	90.8	0.6
STD OXD57 Expected																					
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
Prep/Wash																					

This report supersedes all our previous preliminary and final reports with file number dated prior to the date on this report. Signature indicates that approval; preliminary reports are unsigned and should be used for reference only.

QUALITY CONTROL REPORT

VAN07001993.2

	Method	3A
	Analyte	Au
	Unit	ppb
	MDL	0.5
Pulp Duplicates		
465130	Rock	N.A.
REP 465130	QC	
465201	Rock	19.6
REP 465201	QC	17.0
465211	Rock	N.A.
REP 465211	QC	
Reference Materials		
STD OXD57	Standard	346.1
STD OXD57	Standard	368.9
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T Expected		
STD OXD57 Expected		413
BLK	Blank	
BLK	Blank	
BLK	Blank	
BLK	Blank	
BLK	Blank	
BLK	Blank	<0.5
Prep/Wash		

QUALITY CONTROL REPORT **VAN07001993.2**

		7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
		Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La
		ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm
		0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5
G1	Prep Blank	1.6	4.1	21.0	51	<0.5	4.2	5	732	2.22	<5	3.1	6.9	717	<0.5	<0.5	<0.5	48	2.38	0.08	19.6
G1	Prep Blank	0.5	1.8	21.2	51	<0.5	4.4	4	727	2.21	<5	4.6	7.9	668	<0.5	<0.5	<0.5	48	2.30	0.08	22.7

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval; preliminary reports are unsigned and should be used for reference only.

QUALITY CONTROL REPORT

VAN07001993.2

		7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
		Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	Li	S	Rb	Hf
		ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm
		1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5
G1	Prep Blank	7	0.61	005	0.218	7.98	2.70	2.22	<0.5	8.3	40	1.3	12.7	20.4	1.0	<5	4	34.4	<0.5	96.1	0.6
G1	Prep Blank	9	0.60	822	0.232	7.61	2.68	2.11	<0.5	7.4	46	1.6	12.5	20.5	1.0	<5	4	33.7	<0.5	85.8	0.8

QUALITY CONTROL REPORT

VAN07001993.2

		3A Au ppb 0.5
G1	Prep Blank	N.A.
G1	Prep Blank	N.A.



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Project: Trident
 Report Date: February 05, 2008

Page: 3 of 3 Part 1

CERTIFICATE OF ANALYSIS

VAN07001993.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
Analyte	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	
Unit	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
M DL	0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5	
465130	Rock	1.0	1205	7.0	29	<0.5	70.3	47	154	1.59	62	1.8	5.1	29	<0.5	2.4	2.1	57	1.09	0.04	11.4
465131	Rock	2.3	5294	13.3	37	0.7	139.4	72	229	1.59	273	1.4	4.0	25	<0.5	7.8	4.9	66	1.02	0.09	23.5
465132	Rock	0.6	1053	4.7	19	<0.5	79.0	44	288	1.13	30	1.3	4.0	57	<0.5	1.3	2.5	40	2.40	0.05	18.4
465133	Rock	0.5	1159	3.7	79	<0.5	26.0	35	780	6.48	9	0.7	2.3	247	<0.5	0.6	<0.5	254	7.27	0.11	16.4
465134	Rock	2.4	181.5	5.3	68	<0.5	12.6	7	155	1.48	6	2.4	8.4	88	<0.5	1.0	<0.5	106	0.58	0.04	23.3
465135	Rock	2.2	3592	21.3	65	<0.5	24.0	22	418	4.92	7	1.9	5.3	78	<0.5	2.2	0.9	153	3.38	0.07	19.9
465136	Rock	1.7	43161	10.7	62	1.4	53.2	140	506	6.10	393	1.8	5.0	47	<0.5	9.4	2.5	98	3.06	0.06	16.5
465137	Rock	1.4	21479	3.1	47	<0.5	20.3	35	642	2.03	20	1.6	4.4	55	<0.5	0.9	1.0	31	5.46	0.03	17.4
465200	Rock Pulp	4.5	38.0	4.0	54	<0.5	22.1	10	839	4.11	7	1.0	2.3	282	<0.5	1.0	<0.5	102	2.66	0.07	9.3
465200B	Rock Pulp	67.9	4254	1832	6718	26.9	54.9	62	922	19.36	580	3.1	5.5	149	30.7	114.3	22.5	42	1.72	0.04	11.9
465201	Rock	<0.5	3298	4.6	<5	<0.5	26.2	22	543	1.75	50	1.2	2.5	82	<0.5	0.8	1.1	15	7.56	0.03	10.8
465202	Rock	<0.5	9822	6.8	13	<0.5	18.4	22	351	1.86	103	1.3	2.7	54	<0.5	0.9	1.3	17	4.85	0.08	15.4
465203	Rock	0.9	160959	19.8	72	4.0	55.9	32	202	15.96	66	0.7	1.3	28	<0.5	3.7	3.3	10	2.39	0.03	10.1
465204	Rock	<0.5	45199	6.5	20	1.3	42.2	15	206	5.28	32	0.8	1.5	29	<0.5	1.4	1.7	14	2.98	0.03	9.7
465205	Rock	<0.5	9095	3.8	<5	<0.5	4.9	9	180	1.65	28	<0.5	1.1	33	<0.5	0.6	0.6	<10	2.67	0.04	9.6
465206	Rock	<0.5	5225	3.1	<5	<0.5	11.8	14	393	1.99	34	0.7	1.6	49	<0.5	0.9	0.5	22	6.59	0.06	13.7
465207	Rock	0.5	6814	7.5	6	<0.5	20.6	24	382	1.99	70	1.3	3.4	61	<0.5	0.8	1.1	24	6.88	0.04	26.9
465208	Rock	<0.5	36511	3.1	19	2.0	26.5	13	182	4.77	24	0.8	1.1	25	<0.5	1.2	0.7	<10	3.04	0.06	4.9
465209	Rock	<0.5	5451	4.9	42	<0.5	33.0	38	303	1.42	102	1.0	2.4	62	<0.5	0.9	0.7	20	3.94	0.06	13.6
465210	Rock	<0.5	11460	4.8	<5	0.7	6.9	3	801	2.85	6	1.1	2.2	95	<0.5	0.6	0.7	25	12.65	0.01	8.6
465210A	Rock Pulp	4.0	42.7	3.5	43	<0.5	26.7	11	785	4.00	7	0.7	2.0	256	<0.5	0.9	<0.5	107	2.51	0.07	10.1
465210B	Rock Pulp	75.2	4639	1835	6995	28.3	55.3	66	822	19.77	538	2.7	4.6	123	31.6	96.5	17.9	57	1.78	0.04	11.9
465211	Rock	<0.5	35981	6.5	25	1.3	21.8	6	753	5.16	6	1.4	3.7	82	<0.5	1.3	1.0	27	9.70	0.02	9.7
465212	Rock	0.9	3808	2.3	5	<0.5	5.5	1	103	0.78	<5	<0.5	<0.5	13	<0.5	0.9	<0.5	<10	1.01	0.09	2.0
465213	Rock	<0.5	36925	18.3	47	1.5	38.5	15	340	5.17	14	<0.5	0.5	26	<0.5	2.6	0.8	<10	3.29	<0.01	3.7
465214	Rock	0.8	4812	7.6	12	<0.5	11.2	3	780	1.75	<5	<0.5	<0.5	35	<0.5	1.3	<0.5	11	6.71	<0.01	3.0
465215	Rock	<0.5	1343	10.5	9	0.9	3.7	2	401	1.41	<5	0.6	0.7	39	<0.5	2.2	<0.5	<10	5.60	0.03	8.7
465216	Rock	0.6	5441	14.4	6	2.3	5.8	2	207	1.16	6	<0.5	<0.5	19	<0.5	4.7	<0.5	<10	2.66	0.05	4.8
465217	Rock	<0.5	15965	4.9	17	0.7	10.1	4	556	3.13	<5	<0.5	<0.5	52	<0.5	1.1	<0.5	<10	8.16	<0.01	4.0

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Client: **Action Mineral**
1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project: Trident
Report Date: February 05, 2008

Page: 3 of 3 Part 2

CERTIFICATE OF ANALYSIS

VAN07001993.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
Analyte	Cr	Mg	Ba	Ti	Al	Nb	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	U	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
465130	Rock	16	0.75	359	0.098	2.78	0.04	1.66	<0.5	45.9	29	59	7.7	3.4	<0.5	<5	7	26.1	<0.5	57.0	1.6
465131	Rock	17	0.42	212	0.086	2.56	0.04	1.43	<0.5	38.8	60	26.1	7.2	3.9	<0.5	<5	8	20.0	<0.5	52.5	1.2
465132	Rock	19	0.64	239	0.086	2.63	0.03	1.40	<0.5	41.9	41	43	8.9	4.7	<0.5	<5	8	22.1	<0.5	52.5	1.1
465133	Rock	21	3.01	127	1.207	4.75	1.12	0.51	<0.5	57.8	43	19	26.5	15.7	<0.5	<5	18	40.8	<0.5	11.2	2.9
465134	Rock	27	0.82	2787	0.174	4.30	0.05	3.31	0.8	72.4	49	2.3	13.4	7.3	<0.5	<5	6	31.2	<0.5	99.1	2.7
465135	Rock	33	1.77	1258	0.632	4.47	0.07	2.52	1.2	77.8	44	6.9	16.5	14.7	<0.5	<5	11	34.7	1.3	68.9	2.8
465136	Rock	23	1.74	494	0.495	4.19	0.03	2.03	0.6	81.7	42	16.2	16.7	8.4	<0.5	<5	9	42.9	3.4	66.5	2.7
465137	Rock	17	1.22	137	0.098	2.69	0.02	1.23	<0.5	43.6	43	5.9	13.1	3.9	<0.5	<5	6	31.4	0.6	46.4	1.5
465200	Rock Pulp	44	1.22	465	0.335	7.01	2.76	0.99	0.6	38.5	20	2.5	15.0	3.9	<0.5	<5	13	9.1	<0.5	23.1	1.5
465200B	Rock Pulp	69	0.91	619	0.106	3.89	0.58	1.67	7.3	12.7	21	6.9	7.0	1.5	<0.5	<5	6	7.5	16.2	66.2	<0.5
465201	Rock	105	4.09	46	0.051	1.51	0.02	0.85	<0.5	24.8	24	2.6	10.3	1.9	<0.5	<5	5	8.6	0.5	34.4	1.0
465202	Rock	38	2.51	59	0.051	1.52	0.03	0.87	<0.5	23.4	41	8.1	7.5	1.8	<0.5	<5	4	13.7	0.9	31.0	0.6
465203	Rock	90	1.25	21	0.025	0.83	0.01	0.49	<0.5	11.9	24	131.3	4.8	1.0	<0.5	<5	3	9.7	11.5	17.8	<0.5
465204	Rock	39	1.83	31	0.031	1.08	<0.01	0.57	<0.5	14.8	22	25.2	5.3	1.1	<0.5	<5	3	20.4	3.7	23.5	<0.5
465205	Rock	51	1.39	25	0.023	0.87	0.02	0.46	<0.5	13.0	25	8.5	5.7	1.1	<0.5	<5	1	10.7	0.8	16.7	<0.5
465206	Rock	154	3.46	47	0.032	1.13	0.02	0.59	<0.5	21.2	32	3.5	9.7	1.3	<0.5	<5	3	10.1	0.8	22.1	0.6
465207	Rock	39	3.82	80	0.066	2.09	0.02	1.18	<0.5	34.0	62	3.4	11.5	2.8	<0.5	<5	6	13.4	0.7	46.7	1.4
465208	Rock	149	1.68	23	0.022	0.90	0.02	0.46	<0.5	10.6	16	29.9	5.5	1.4	<0.5	<5	2	20.2	3.2	17.2	<0.5
465209	Rock	46	2.18	44	0.049	1.53	0.02	0.80	<0.5	23.9	34	2.6	7.2	1.9	<0.5	<5	4	16.2	<0.5	29.3	0.7
465210	Rock	67	6.82	54	0.051	1.46	0.02	0.81	<0.5	22.1	20	3.6	17.5	1.9	<0.5	<5	3	9.8	0.9	34.1	0.8
465210A	Rock Pulp	42	1.25	548	0.328	6.92	2.77	0.97	<0.5	35.3	21	1.8	17.6	4.6	<0.5	<5	13	8.8	<0.5	23.4	1.1
465210B	Rock Pulp	68	0.95	605	0.106	4.00	0.60	1.67	8.8	11.7	23	6.3	6.9	1.4	<0.5	<5	6	6.5	17.5	66.7	<0.5
465211	Rock	28	5.45	69	0.087	2.36	0.04	1.25	<0.5	36.3	23	19.7	13.4	3.7	<0.5	<5	3	12.5	3.3	53.4	1.1
465212	Rock	235	0.50	16	0.008	0.44	0.05	0.23	<0.5	2.2	<5	1.3	3.0	<0.5	<0.5	<5	<1	13.5	<0.5	8.0	<0.5
465213	Rock	42	1.79	18	0.012	0.47	0.03	0.25	<0.5	6.0	10	16.1	3.8	0.6	<0.5	<5	2	10.3	3.8	10.1	<0.5
465214	Rock	206	3.57	11	0.005	0.29	0.04	0.14	<0.5	2.3	6	1.5	5.1	<0.5	<0.5	<5	3	13.4	0.5	5.3	<0.5
465215	Rock	39	3.10	21	0.017	0.78	0.04	0.41	<0.5	8.1	22	10	8.1	0.9	<0.5	<5	1	33.2	<0.5	14.9	<0.5
465216	Rock	237	1.40	20	0.013	0.61	0.04	0.32	<0.5	4.4	11	2.3	3.3	0.9	<0.5	<5	1	23.4	<0.5	11.0	<0.5
465217	Rock	56	4.34	9	0.007	0.31	0.04	0.17	<0.5	3.6	10	6.6	9.6	0.6	<0.5	<5	2	9.1	1.6	6.4	<0.5

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Missy Diamond drill Core Assays



ACME ANALYTICAL LABORATORIES LTD.
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Client: Bradford Minerals
 1255 W. Pender St
 Vancouver BC V6E 2V1 Canada

Submitted By: George Cootzee
 Receiving Lab: Acme Analytical Laboratories (Vancouver) Ltd.
 Received: November 29, 2007
 Report Date: January 08, 2008
 Page: 1 of 2

CERTIFICATE OF ANALYSIS VAN07002749.1

CLIENT JOB INFORMATION

Project: None Given
 Shipment ID:
 P.O. Number:
 Number of Samples: 7

SAMPLE PREPARATION AND ANALYTICAL PROCEDURES

Method Code	Number of Samples	Code Description	Test Wgt (g)	Report Status
R150	7	Crush split and pulverize drill core to 100mesh		
7TX	7	4 Acid Digestion Analysis by ICP-ES/ICP-MS	0.5	Completed

SAMPLE DISPOSAL

DISP-PLP Dispose of Pulp After 90 days

ADDITIONAL COMMENTS

Acme does not accept responsibility for samples left at the laboratory after 90 days without prior written instructions for sample storage or return.

Invoice To: Bradford Minerals
 1255 W. Pender St.
 Vancouver BC V6E 2V1
 Canada

CC:



This report is for your use only. It is preliminary and final reports will be issued prior to the date on this certificate. Signature indicates final approval, preliminary reports are for your use only and should be used for reference only. All test results are considered the confidential property of the client. Acme assumes the liability for analytical results only.



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Client:

Bradford Minerals

1255 W. Pender St.
Vancouver BC V6E 2V1 Canada

Project:

None Given

Report Date:

January 08, 2008

Page:

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Part 1

CERTIFICATE OF ANALYSIS

VAN07002749.1

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	
Unit	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
MDL	0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5	
465307	Drill Core	0.8	22.2	40.4	11	<0.5	11.2	6	959	4.62	22	17	4.1	81	<0.5	2.1	<0.5	22	14.07	0.03	16.5
465308	Drill Core	1.3	44.9	10.3	<5	<0.5	10.7	10	630	3.34	12	2.5	6.5	112	<0.5	2.2	0.9	26	11.86	0.02	21.6
465309	Drill Core	<0.5	28929	6.0	32	1.8	18.1	5	931	5.12	<5	0.5	1.2	106	<0.5	1.3	1.1	<10	12.72	<0.01	7.4
465312	Drill Core	2.6	21.4	23.2	8	<0.5	18.9	21	492	2.13	21	3.5	7.3	101	<0.5	2.9	0.8	62	9.71	0.03	26.8
465317	Drill Core	1.1	1890	5.5	<5	<0.5	4.1	3	601	2.72	<5	0.9	1.9	106	<0.5	1.0	0.7	14	12.78	0.01	16.9
1	Drill Core	4.3	47.8	7.0	64	<0.5	28.4	10	912	4.22	<5	1.0	2.3	290	<0.5	1.0	<0.5	115	2.60	0.06	11.1
2	Drill Core	79.0	4394	1938	6724	26.9	51.7	63	988	20.83	537	3.4	5.8	148	36.9	130.6	23.5	66	1.83	0.04	12.8

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Client: **Bradford Minerals**

1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project: None Given

Report Date: January 08, 2008

Page: 2 of 2 Part 2

CERTIFICATE OF ANALYSIS

VAN07002749.1

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	Li	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
466307	Drill Core	12	7.66	74	0.095	2.21	0.01	1.30	<0.5	42.8	33	1.1	13.5	3.6	<0.5	<5	3	16.5	3.4	48.8	1.2
466308	Drill Core	13	8.33	61	0.117	3.11	0.03	1.12	<0.5	55.6	46	1.4	15.2	5.1	<0.5	<5	4	43.9	1.2	41.8	1.9
466309	Drill Core	5	6.96	21	0.019	0.70	0.02	0.36	<0.5	12.2	17	8.3	14.1	1.2	<0.5	<5	3	6.8	2.9	11.9	<0.5
466312	Drill Core	21	6.78	125	0.172	4.10	0.03	2.49	0.5	75.2	60	2.8	17.2	5.6	<0.5	<5	5	14.6	1.2	94.5	2.3
466317	Drill Core	11	7.02	36	0.033	1.06	0.03	0.60	<0.5	20.0	38	1.7	16.3	2.0	<0.5	<5	3	8.5	0.8	20.0	<0.5
1	Drill Core	48	1.27	627	0.339	7.03	2.68	1.00	0.7	34.3	21	2.1	147	4.8	<0.5	<5	12	14.2	<0.5	21.8	1.3
2	Drill Core	61	0.98	285	0.110	3.99	0.61	1.80	7.3	14.1	24	7.6	7.2	2.0	<0.5	<5	6	7.7	19.1	66.2	<0.5

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Client: **Bradford Minerals**

1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project: *None Given*

Report Date: January 09, 2008

Page: 1 of 1 Part 1

QUALITY CONTROL REPORT

VAN07002749.1

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
Analyte	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	La
Unit	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	ppm
MDL	0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5	0.5
Reference Materials																					
STD SF-3T	Standard	321.0	7870	9410	10047	52.3	3567	186	4327	8.21	40	4.0	4.8	446	63.9	11.2	5.0	136	4.16	0.06	18.1
STD SF-3T	Standard	316.4	7795	9417	10092	52.4	3623	187	4394	8.22	42	4.0	4.7	440	59.8	11.0	4.9	135	4.14	0.06	19.4
STD SF-3T Expected		320	7723	9610	10672	52	3500	181	4320	8.33	40	4	4.7	440	47.5	11.1	4.8	143	4.1	0.06	17
BLK	Blank	<0.5	<0.5	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
Prep Wash																					
G-1	Prep Blank	0.5	11.0	32.4	54	<0.5	6.9	5	799	2.37	<5	4.2	8.0	728	<0.5	<0.5	<0.5	53	2.49	0.08	25.9
G-1	Prep Blank	0.8	9.5	21.6	51	<0.5	6.8	5	794	2.30	6	3.6	7.7	698	<0.5	<0.5	<0.5	51	2.49	0.08	23.9

This report is preliminary and should not be used for legal purposes. All data are preliminary and should be used for reference only.

QUALITY CONTROL REPORT

VAN07002749.1

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
Analyte	Cr	Mg	Ba	Ti	Kl	Ns	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	Li	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
Reference Material:																					
STD SF-3T	Standard	178	4.70	513	0.197	5.51	2.08	2.56	4.2	13.6	40	6.3	10.7	15.0	0.5	<5	7	25.6	4.2	90.7	<0.5
STD SF-3T	Standard	168	4.71	514	0.197	5.43	2.07	2.60	4.3	13.9	40	6.1	10.7	14.8	0.5	<5	7	24.9	4.2	91.8	0.6
STD SF-3T Expected		207.4	4.67	508	0.19	5.43	2.08	2.47	4.3	14	38	5.8	11.5	15.1	0.9	0	7	19.1	3.5	90.8	0.6
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
Prep/Wash																					
G1	Prep Blank	10	0.71	1681	0.231	8.16	2.77	3.14	<0.5	9.9	49	1.3	14.6	26.9	1.0	<5	5	42.4	<0.5	135.4	0.6
G1	Prep Blank	7	0.72	1007	0.220	7.89	2.69	2.95	<0.5	8.2	44	1.4	13.9	23.9	1.0	<5	4	32.2	<0.5	126.9	0.6

Missy NE Extension assays (Samples 8 and 9 were taken on the Taya Claim)



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Client: Bradford Minerals
1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Submitted By: George Coetzee
Receiving Lab: Acme Analytical Laboratories (Vancouver) Ltd.
Received: September 24, 2007
Report Date: February 01, 2008
Page: 1 of 2

CERTIFICATE OF ANALYSIS

VAN07001699.2

CLIENT JOB INFORMATION

Project: None Given
Shipment ID:
P.O. Number:
Number of Samples: 11

SAMPLE DISPOSAL

DISP-PLP Dispose of Pulp After 90 days

Acme does not accept responsibility for samples left at the laboratory after 90 days without prior written instructions for sample storage or return.

Invoice To: Bradford Minerals
1255 W. Pender St
Vancouver BC V6E 2V1
Canada

CC:

SAMPLE PREPARATION AND ANALYTICAL PROCEDURES

Method Code	Number of Samples	Code Description	Test Wgt (g)	Report Status
R150	9	Crush, split and pulverize rock to 150 mesh		
7TX	11	-4 Acid Digestion Analysis by ICP-ES/ICP-MS	0.5	Completed
3A	6	Ignite samples, acid digest, Au by ICP-MS	15	Completed

ADDITIONAL COMMENTS

Version 2 to include Au by 3A analysis



This report represents all previous preliminary and final reports with this file number dated prior to the date of this certificate. Signatures indicate final approval, preliminary reports are unsigned and should be filed for reference only. All test results are considered the confidential property of the client. Acme assumes the liability for analytical results only.

CERTIFICATE OF ANALYSIS

VAN07001699.2

Method	Analyte	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX
		Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La
Unit		ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
MDL		0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	10	0.01	0.01	0.5	
1	Rock	0.8	82.9	46.0	<5	<0.5	71.4	29	178	7.57	7	1.8	4.8	21	<0.5	2.0	<0.5	<10	0.79	0.06	4.8
2	Rock	0.8	11831	1.0	9	3.6	4.8	6	116	1.46	13	<0.5	<0.5	16	<0.5	0.8	<0.5	<10	1.10	0.11	2.1
3	Rock	12.0	296.0	842.2	5	5.5	147.9	46	15	28.91	127	0.9	0.7	<5	<0.5	23.5	<0.5	<10	0.02	<0.01	1.0
4	Rock	0.9	14487	12.3	27	7.8	14.8	38	256	3.44	82	0.8	1.5	10	<0.5	1.4	1.6	10	0.64	0.15	4.0
5	Rock	102.9	402960	964.2	811	<0.5	331.3	44	44	29.26	101	0.8	<0.5	<5	19.9	81.0	19.9	22	0.03	<0.01	2.1
6	Rock	0.7	1248	7.0	6	<0.5	27.5	29	533	4.58	<5	0.9	0.8	36	<0.5	0.9	<0.5	163	6.91	0.20	11.3
7	Rock	1.1	1009	6.6	18	<0.5	44.5	35	165	20.94	6	<0.5	<0.5	20	<0.5	1.6	<0.5	249	0.56	0.06	5.0
8	Rock	0.6	72722	12.6	16	2.4	155.5	599	193	19.92	290	<0.5	<0.5	6	<0.5	2.8	3.1	<10	1.20	<0.01	0.8
9	Rock	13.9	1151	384.0	14	3.3	91.9	39	196	13.20	56	3.4	2.7	39	<0.5	13.3	<0.5	19	4.03	0.07	3.5
10	Rock Pulp	73.1	4362	1978	6418	26.7	57.9	67	844	20.07	667	3.4	5.7	137	34.5	116.7	21.7	51	1.78	0.04	12.0
11	Rock Pulp	4.0	42.8	6.1	54	<0.5	24.7	12	811	4.08	6	1.0	2.6	287	<0.5	1.1	<0.5	108	2.67	0.07	10.7



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Client:

Bradford Minerals

1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project:

None Given

Report Date:

February 01, 2008

Page:

2 of 2

Part 2

CERTIFICATE OF ANALYSIS

VAN07001699.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	Li	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
1	Rock	115	0.51	177	0.100	2.50	0.04	2.44	0.5	41.9	12	20	6.9	4.0	<0.5	<5	3	16.9	6.9	46.5	1.5
2	Rock	61	0.57	25	0.010	0.53	0.06	0.22	<0.5	3.2	6	16.2	2.9	<0.5	<0.5	<5	3	54.5	0.7	4.7	<0.5
3	Rock	125	0.03	23	0.032	0.23	0.01	0.16	<0.5	15.5	<5	1.1	1.5	1.7	<0.5	<5	<1	8.4	300	4.5	<0.5
4	Rock	74	0.42	46	0.046	1.07	0.03	0.54	<0.5	16.9	9	10.5	5.4	1.9	<0.5	<5	3	32.3	1.0	13.5	0.5
5	Rock	23	0.03	59	0.047	0.30	<0.01	0.27	<0.5	13.3	<5	0.6	3.1	<0.5	<0.5	<5	3	1.0	147	3.9	<0.5
6	Rock	62	2.93	60	0.211	2.66	0.02	1.27	<0.5	28.2	27	3.3	13.9	1.8	<0.5	<5	23	21.8	<0.5	24.5	0.7
7	Rock	166	4.44	112	0.366	3.64	0.01	0.32	<0.5	13.2	13	0.9	11.1	2.9	<0.5	<5	18	73.0	<0.5	7.6	<0.5
8	Rock	34	0.64	6	0.006	0.20	<0.01	0.09	<0.5	1.1	<5	4.3	3.7	<0.5	<0.5	<5	4	15.7	148	1.0	<0.5
9	Rock	47	1.97	61	0.079	0.72	0.02	0.44	<0.5	85.3	8	0.6	3.7	3.1	<0.5	<5	<1	7.4	110	10.0	2.1
10	Rock Pulp	91	0.95	629	0.114	3.05	0.59	1.68	8.3	14.3	24	6.5	6.7	1.7	<0.5	<5	5	7.5	170	67.7	0.5
11	Rock Pulp	49	1.29	538	0.347	7.49	2.88	1.09	0.7	31.3	22	2.3	15.8	4.6	<0.5	<5	13	10.3	<0.5	24.2	1.4

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval; preliminary reports are unsigned and should be used for reference only.



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Client: **Bradford Minerals**
1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project: None Given
Report Date: February 01, 2008

Page: 2 of 2 **Part:** 3

CERTIFICATE OF ANALYSIS

VAN07001699.2

	Method	3A
	Analyte	Au
	Unit	ppb
	MDL	0.5
1	Rock	N.A.
2	Rock	N.A.
3	Rock	2.0
4	Rock	16.5
5	Rock	3.2
6	Rock	N.A.
7	Rock	N.A.
8	Rock	46.0
9	Rock	N.A.
10	Rock Pulp	517.0
11	Rock Pulp	42.2

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QUALITY CONTROL REPORT

VAN07001699.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	
Unit	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	
MDL	0.5	0.5	0.5	5	0.5	0.5	1	5	0.01	5	0.5	0.5	5	0.5	0.5	0.5	10	0.01	0.01	0.5	
Pulp Duplicates																					
11	Rock Pulp	4.0	42.8	6.1	54	<0.5	247	12	811	4.08	6	1.0	2.6	287	<0.5	1.1	<0.5	108	2.67	0.07	10.7
REP 11	QC	4.7	55.7	5.3	58	<0.5	310	13	819	4.12	6	0.9	2.4	292	<0.5	0.9	<0.5	112	2.82	0.07	10.7
Reference Materials																					
STD DxD57	Standard																				
STD DxD57	Standard																				
STD SF-3T	Standard	317.3	7738	9383	10673	52.8	3562	183	4297	8.14	42	4.1	4.6	428	50.3	9.5	4.7	136	4.10	0.06	17.0
STD SF-3T	Standard	316.0	7719	9298	10639	52.9	3550	188	4246	8.17	41	3.8	4.4	430	49.6	10.6	4.6	136	4.07	0.06	18.0
STD SF-3T	Standard	327.6	7968	9466	11241	52.5	3696	189	4397	8.28	40	4.7	4.5	418	48.3	9.4	4.6	135	4.16	0.06	17.4
STD SF-3T	Standard	324.6	7896	9418	10991	52.9	3588	183	4276	8.22	40	3.9	4.4	422	47.3	9.1	4.6	135	4.13	0.06	17.3
STD SF-3T Expected		320	7723	9610	10672	52	3600	181	4320	8.33	40	4	4.7	440	47.5	11.1	4.8	143	4.1	0.06	17
STD DxD57 Expected																					
BLK	Blank	<0.5	<0.5	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank	<0.5	5.1	<0.5	<5	<0.5	<0.5	<1	<5	<0.01	<5	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<10	<0.01	<0.01	<0.5
BLK	Blank																				
Prep Wash																					
G-1	Prep Blank	<0.5	8.1	2.12	58	<0.5	4.2	5	742	2.23	<5	4.8	8.7	664	<0.5	<0.5	<0.5	50	2.40	0.09	27.9
G-1	Prep Blank	<0.5	8.3	20.1	54	<0.5	5.3	5	754	2.30	<5	4.1	8.5	647	<0.5	<0.5	<0.5	53	2.37	0.09	28.4



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1255 W. Pender St
Vancouver BC V6E 2V1 Canada

Project

None Given

Report Date:

February 01, 2008

Page:

1 of 1

Part 2

QUALITY CONTROL REPORT

VAN07001699.2

Method	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	7TX	
Analyte	Cr	Mg	Ba	Ti	Al	Na	K	W	Zr	Ce	Sn	Y	Nb	Ta	Be	Sc	Li	S	Rb	Hf	
Unit	ppm	%	ppm	%	%	%	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	
MDL	1	0.01	5	0.001	0.01	0.01	0.01	0.5	0.5	5	0.5	0.5	0.5	0.5	5	1	0.5	0.5	0.5	0.5	
Pulp Duplicates																					
11	Rock Pulp	49	1.29	538	0.347	7.49	2.88	1.09	0.7	31.3	22	2.3	15.8	4.6	<0.5	<5	13	10.3	<0.5	24.2	1.4
REP 11	QC	54	1.31	540	0.337	7.46	2.82	1.06	0.7	33.8	22	2.2	16.6	5.1	<0.5	<5	13	10.6	<0.5	25.3	1.3
Reference Materials																					
STD OXD57	Standard																				
STD OXD57	Standard																				
STD SF-3T	Standard	170	4.67	531	0.189	5.50	2.10	2.44	4.1	11.9	40	6.0	10.8	12.4	0.7	<5	6	25.4	3.8	87.7	<0.5
STD SF-3T	Standard	195	4.66	545	0.188	5.44	2.07	2.50	4.5	13.5	41	5.8	10.9	12.4	0.7	<5	7	24.3	3.7	90.7	0.6
STD SF-3T	Standard	209	4.70	610	0.193	5.53	2.12	2.36	4.3	13.1	40	5.7	10.9	12.6	<0.5	<5	7	23.6	3.7	92.4	0.6
STD SF-3T	Standard	206	4.67	528	0.193	5.49	2.09	2.38	4.1	11.8	40	5.7	10.6	14.2	<0.5	<5	6	23.3	3.7	88.8	0.6
STD SF-3T Expected		207.4	4.67	508	0.19	5.43	2.06	2.47	4.3	14	38	5.8	11.5	15.1	0.9	0	7	19.1	3.5	90.8	0.6
STD OXD57 Expected																					
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank	<1	<0.01	<5	<0.001	<0.01	<0.01	<0.01	<0.5	<0.5	<5	<0.5	<0.5	<0.5	<0.5	<5	<1	<0.5	<0.5	<0.5	<0.5
BLK	Blank																				
Prep Wash																					
G1	Prep Blank	74	0.69	1001	0.219	8.16	2.60	3.15	<0.5	7.8	53	1.3	13.9	19.5	1.2	<5	5	41.7	<0.5	131.0	0.7
G1	Prep Blank	28	0.71	960	0.238	7.90	2.64	3.10	<0.5	9.0	54	1.4	15.6	21.5	1.4	<5	6	39.8	<0.5	129.1	0.6

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval, preliminary reports are unsigned and should be used for reference only.



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Client: **Bradford Minerals**
1255 W. Pender St.
Vancouver BC V6E 2V1 Canada

Project: None Given
Report Date: February 01, 2008

Page: 1 of 1 **Part:** 3

QUALITY CONTROL REPORT

VAN07001699.2

	Method	3A
	Analyte	Au
	Unit	ppb
	MDL	0.5
Pulp Duplicates		
11	Rock Pulp	42.2
REP 11	QC	
Reference Materials		
STD OXD57	Standard	346.1
STD OXD57	Standard	368.9
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T	Standard	
STD SF-3T Expected		
STD OXD57 Expected		413
BLK	Blank	
BLK	Blank	
BLK	Blank	<0.5
Prep/Wash		
G1	Prep Blank	N.A.
G1	Prep Blank	N.A.

The report is prepared as preliminary and the report with the number does prior to be done on this or blank. Signature indicates that approval, preliminary reports are unsigned and should be used for reference only.

APPENDIX D
Racing River Claims

TWENTY-SEVEN CAPITAL CORP.- CLAIM LIST SEPTEMBER 6, 2006

Claim Name	Grant Number	District	Registered	Mining
------------	--------------	----------	------------	--------

Bronson and Toro Properties (1)

Bronson	501161	094K		
428 North	501179	094K		
Book	501201	094K		
Toro	504869	094K		

Muskwa Property

Delano 1-2	508511-508512	094K		
3	508515	094K		
3	508554	094K		
4	508521	094K		
5	508527	094K		
6	508535	094K		
7	508537	094K		
8	508540	094K		
9	508771	094K		
10-11	511472-511473	094K		
12-13	511475-511476	094K		
14	511478	094K		
15	511480	094K		
16-17	511482-511483	094K		
18	511485	094K		
19	511488	094K		
20	511490	094K		
21-22	511619-511620	094K		
23	515490	094K		
24	515495	094K		
25	515505	094K		
26	515516	094K		
27-28	517636-517637	094K		
28	517639	094K		

Dieppe 1-4	508597-508600	094K		
6-7	508602-508603	094K		
8-10	508605-508607	094K		
11	508609	094K		
12	508617	094K		
13	508621	094K		
14	508623	094K		
15	508627	094K		

(1) optioned from Horst Klassen

Dieppe 16	508629	094K		
17	508633	094K		
17	508634	094K		
18	508636	094K		
18	508639	094K		
20	508642	094K		
21-22	508644-508645	094K		

23 508647 094K
 24 508651 094K
 25 508656 094K
 26 508659 094K
 27 508666 094K
 28-29 508670-508671 094K
 30 508675 094K
 31-36 508685-508690 094K
 36 508691 094K
 38-40 508692-508694 094K
 41-42 508696-508697 094K
 43 508699 094K
 44 508704 094K
 45 511492 094K
 46 511494 094K
 46 511496 094K
 47 511498 094K
 48 511500 094K
 49 511600 094K
 50-52 511602-511604 094K
 53 511614 094K
 54-55 525822-525823 094K
 Gataga 1-2 508444-508445 094K
 508447 094K
 508449 094K
 508450-508452 094K
 8-11 508454-508457 094K
 12-13 508459-508460 094K
 Gataga 14 508462 094K
 15 508464 094K
 16 508467 094K
 17-19 508469-508471 094K
 20 509141 094K
 21 511520 094K
 22-23 511522-511523 094K
 24-25 511525-511526 094K
 26-32 511528-511534 094K
 33-36 511536-511539 094K
 37-38 511615-511616 094K
 39 511618 094K
 Grizzly 1 508545 094K
 2 508550 094K
 4 508557 094K
 5 508560 094K
 6-11 511143-511148 094K
 12-13 511150-511151 094K
 13 511153 094K
 14 511155 094K
 15 511157 094K
 16 511159 094K
 16 511160 094K
 17 511162 094K
 18 511165 094K
 19-20 511188-511189 094K
 21-23 511191-511193 094K
 24 511195 094K
 25 511198 094K
 26-27 511200-511201 094K

28	511203 094K
29	511205 094K
30	511212 094K
31	511215 094K
32	511217 094K
33	511219 094K
34	511220 094K
35-36	511222-511223 094K
37	511225 094K
38	511228 094K
39	511232 094K
40-41	511235-511236 094K
42	511242 094K
43	511245 094K
Grizzly 44-45	511247-511248 094K
46	511250 094K
47-49	511252-511254 094K
50	511256 094K
51	511258 094K
52	511260 094K
53-54	511262-511263 094K
55	511265 094K
56-58	511267-511269 094K
59-64	511271-511276 094K
65-72	518973-518980 094K
73-76	525771-525774 094K
77	525780 094K
78-80	525783-525785 094K
81-83	525787-525789 094K
84-85	525791-525792 094K
86-87	525794-525795 094K
88-90	525797-525799 094K
91-95	525801-525805 094K
96-97	525808-525809 094K
98	525811 094K
99-101	525814-525816 094K
102	525818 094K
103-104	525820-525821 094K
Socrates 1	508479 094K
2	508482 094K
2	508483 094K
4-10	508484-508490 094K
11	508492 094K
12	508494 094K
13	508497 094K
14	508504 094K
15-19	508506-508510 094K
20	511436 094K
21	511439 094K
22	511441 094K
23	511443 094K
24-27	511446-511449 094K
28-38	511451-511461 094K
39	511463 094K
40-41	511465-511466 094K
42-44	511595-511597 094K
Socrates 45	511599 094K
46	515464 094K

47-49 515466-515468 094K
50-52 515470-515472 094K
53 515476 094K
54 515482 094K
55 515485 094K
56 515811 094K
57 515813 094K
58-68 515816-15826 094K

Toad 1 508707 094K
2-3 508709-508710 094K
4 511502 094K
5 511505 094K
6 511507 094K
7 511509 094K
8-10 511511-511513 094K
11 511515 094K
12-13 511607-511608 094K
14-15 511610-511611 094K
16 511613 094K
17 517407 094K
18 517410 094K

Bronson and Toro Properties (1)

Bronson 501161 094K
428 North 501179 094K
Book 501201 094K
Toro 504869 094K

*Muska Property

Delano 1-2 508511-508512 094K
3 508515 094K
3 508554 094K
4 508521 094K
5 508527 094K
6 508535 094K
7 508537 094K
8 508540 094K
9 508771 094K
10-11 511472-511473 094K
12-13 511475-511476 094K
14 511478 094K
15 511480 094K
16-17 511482-511483 094K
18 511485 094K
19 511488 094K
20 511490 094K
21-22 511619-511620 094K
23 515490 094K
24 515495 094K
25 515505 094K
26 515516 094K
27-28 517636-517637 094K
28 517639 094K
D'Hope 1-4 508597-508600 094K
6-7 508602-508603 094K
8-10 508605-508607 094K

11	508609 094K
12	508617 094K
13	508621 094K
14	508623 094K
15	508627 094K

(1) optioned from Horst Klassen

Claim Name	Grant Number	District	Registered	Mining
Dieppe 16	508629 094K			
17	508633 094K			
17	508634 094K			
18	508636 094K			
18	508639 094K			
20	508642 094K			
21-22	508644-508645 094K			
23	508647 094K			
24	508651 094K			
25	508656 094K			
26	508659 094K			
27	508666 094K			
28-29	508670-508671 094K			
30	508675 094K			
31-36	508685-508690 094K			
36	508691 094K			
38-40	508692-508694 094K			
41-42	508696-508697 094K			
43	508699 094K			
44	508704 094K			
45	511492 094K			
46	511494 094K			
46	511496 094K			

TWENTY-SEVEN CAPITAL CORP.- CLAIM LIST SEPTEMBER 6, 2006

Muskwa Property (cont'd)

Claim Name	Grant Number	District	Registered	Mining
47	511498 094K			
48	511500 094K			
49	511600 094K			
50-52	511602-511604 094K			
53	511614 094K			
54-55	525822-525823 094K			
Gataga 1-2	508444-508445 094K			
508447 094K				
508449 094K				
508450-508452 094K				
8-11	508454-508457 094K			
12-13	508459-508460 094K			
Gataga 14	508462 094K			
15	508464 094K			
16	508467	094K		
17-19	508469-508471 094K			
20	509141 094K			
21	511520 094K			
-23	511522-511523 094K			
24-25	511525-511526 094K			

26-32	511528-511534 094K
33-36	511536-511539 094K
37-38	511615-511616 094K
39	511618 094K
Grizzly M 1	508545 094K
2	508550 094K
4	508557 094K
5	508560 094K
6-11	511143-511148 094K
12-13	511150-511151 094K
13	511153 094K
14	511155 094K
15	511157 094K
16	511159 094K
16	511160 094K
17	511162 094K

TWENTY-SEVEN CAPITAL CORP. - CLAIM LIST SEPTEMBER 6, 2006

	Registered	Mining
Claim Name	Grant Number	District

Muskwa Property (cont'd)

18	511165 094K
19-20	511188-511189 094K
21-23	511191-511193 094K
24	511195 094K
25	511198 094K
26-27	511200-511201 094K
28	511203 094K
29	511205 094K
30	511212 094K
31	511215 094K
32	511217 094K
33	511219 094K
34	511220 094K
35-36	511222-511223 094K
37	511225 094K
38	511228 094K
39	511232 094K
40-41	511235-511236 094K
42	511242 094K
43	511245 094K
Grizzly 44-45	511247-511248 094K
46	511250 094K
47-49	511252-511254 094K
50	511256 094K
51	511258 094K
52	511260 094K
53-54	511262-511263 094K
55	511265 094K
56-58	511267-511269 094K
59-64	511271-511276 094K
65-72	518973-518980 094K
73-76	525771-525774 094K
77	525780 094K
78-80	525783-525785 094K

11 511515 094K
12-13 511607-511608 094K
14-15 511610-511611 094K
16 511613 094K
17 517407 094K
18 517410 094K

APPENDIX E

DRILL AND GEOTECHNICAL LOGS

Drill Hole: MY-07-01 Claim: Missy N: 6485874 E: 363703 Final depth: 37.5m

AZ: 104 DIP: -45 EL: 1435m DHS: 17 Febr 2007 DHF: 22 Febr 2007

Logged by: David Peake Teched by: George Coetzee

Notes: To intersect the three exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-01

From: To:
0 18m

Notes: **Casing**

Drill Hole: MY-07-01

From: To:
0 37.5m

Notes: About 60% of the rocks are carbonates. All the larger rocks greater than 3cm consist of dolomite except for a 0.18m long dyke material at 31.2m depth. There are two more 3cm dyke rocks at about 8m and 12m depth. A shale segment begins after around 35m. There are two prominent clay seems at around 26m and 32m.

Lithology: Assorted glacial rock rubble and two clay seems.

Structure: Glacial rocks, pebbles and gravel to larger boulders (largest 19cm). Mud and sand most likely also was incorporated but was washed out during drilling.

Alteration: Glacial weathering and rounding

Veining: Two rocks in the 32m clay seem have small calcite veins.

Mineralization: No apparent.

Final Depth: Abandoned at 37.5m

Drill Hole: MY-07-02 Claim: Missy N: 6485874 E: 363703 Final depth: 32m
AZ:138 DIP: -60 EL:1435m DHS: Aug. 27th, 2007 DHF: September 2nd, 2007
Logged by: David Peake Teched by: George Coetzee

Notes: To intersect the three veins exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-02

From: To:
0 21.34m

Notes: **Casing**

Drill Hole: MY-07-02

From: To:
0 32.00m

Notes: About 65-75% of the rocks are a siliceous sandstone. All the larger rocks greater than 3cm consist of this sandstone except for core segments deeper than about 30m which then some of the larger rocks are shale float. Shale mostly shows up at these deeper depths and consists of about 5-10% of the core. There are a few fragments 2% that are of the green and red conglomerate. There is some dyke material also within the pebbles, the largest 3cm wide, about 1% of the core. There is also a clay section with mostly shale fragments, but there may have been more clay but it was washed out.

Lithology: Assorted glacial rock rubble

Structure: Glacial rocks, pebbles and gravel to larger boulders (largest 19cm). Mud and sand most likely also was incorporated but was washed out during drilling.

Alteration: Glacial weathering and rounding

Veining: Four larger rocks contain calcite veining varying from fracture points to pockets but none are wider than 0.5cm except for one segment that is about 4 cm thick but only protrudes through have the core width. There is about 1% of the float that is calcite pebbles.

Mineralization: No apparent other than some iron oxidation on about 5-10% of the siliceous fine grained sandstones.

Final Depth: Abandoned at 32.00m

Drill Hole: MY-07-03 Claim: Missy N: 6485797 E: 363752 Final depth: 25.3m
AZ: 271 DIP:- 60 EL: 1409m DHS: Sept. 8th, 2007 DHF: September 12th, 2007
Logged by: David Peake Teched by: George Coetzee

Notes: To intersect the three exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-03

From: To:
0 9.75m
Notes: **Casing**

Drill Hole: MY-07-03

From: To:
0.61 3.05m

Notes: Fine silts with various sizes of pebbles incorporated.

Lithology: River rock, organic matter, and a clay that has calcite within (fizzes from HCl)

Structure: Mud most likely from decayed organic matter .

Alteration: Decomposed matter and runoff

Veining: None

Mineralization: No apparent mineralization

Drill Hole: MY-07-03

From: To:

3.05m 15.85m

Notes: About 70-75% of the material is a fine grained sandstone, ranging in size from sands to 7cm rocks. Five larger pieces of dyke material are intermingled all of which are about 2-3cm cubed. Drywall mud is also incorporated by the drillers to reduce collapsing of the hole.

Lithology: Assorted glacial rock rubble

Structure: Glacial rocks, pebbles and gravel to larger boulders (largest 7cm). Mud and sand most likely also was incorporated but was washed out during drilling.

Alteration: Glacial weathering and rounding

Veining: No apparent other than a few calcitic pebbles

Mineralization: One segment of mostly pyrite incorporated into a shale (0.5cm squared)

Drill Hole: MY-07-03

From: 15.85 To: 25.3

Notes:

About 70% of the material is a fine and coarse grained sandstone, ranging in size from sands to 27cm rocks. One larger piece of dyke material with a length of 23cm at 25.30m depth. Contains one exposed surface also on the core side giving the appearance of a halved core section suggesting cored from a boulder. Also intermingled there are a few pieces of dyke material 2-3cm in diameter. About 20% is shale or varying sizes, largest being 20cm. One larger piece of conglomerate 14cm long with veining.

Lithology:

Assorted glacial rock rubble

Structure:

Glacial and possibly glacial rocks, pebbles and gravel to larger boulders (largest 27cm). Mud and sand most likely also was incorporated but was washed out during drilling.

Alteration:

Glacial weathering and rounding, one piece of dyke material 0.22m long.

Veining:

One conglomerate boulder (13cm) has veining across the length about 3mm wide. No other apparent veining other than a few calcitic pebbles.

Mineralization:

No apparent mineralization

Final Depth: Abandoned at 25.3m

Drill Hole: MY-07-04 Claim: Missy N: 6485733 E: 363732 Final depth: 57.61m
AZ: 337 DIP: -78 EL: 1413m DHS: Sept. 15, 2007 DHF: Sept. 21, 2007
Logged by: David Peake Teched by: George Coetzee

Notes: To intersect vein three exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-04

From: To:

0 20.12

Notes: **Casing**

Drill Hole: MY-07-04

From: To:
0 20.14

- Notes: There is a 0.23m section, biggest of all is a silica/calcite matrix with a combination of chalcopyrite and pyrite. Judging from the amount of malachite versus iron oxides, there appears to be more pyrite, with a ratio of 75:25 pyrite:chalcopyrite. The other rocks consist of grey/black shale, silica rich bedrock (some with dark chlorite stringers, with malachite on them, one piece is about 1cm in diameter), red iron rich bedrock, a calcite rich pebble, one pebble has heavy folding
- Lithology: Assorted till rock rubble
- Structure: Glacial till rocks, pebbles to larger boulders (of 23cm). Mud and sand most likely also was incorporated but was washed out during drilling.
- Alteration: Glacial weathering and rounding, within the silica 23cm section there is minor amounts of Fe oxidation and minor malachite secondary mineralization.
- Veining: Some pieces have minor veining (less than 1%) incorporated or are part of a larger structure but no piece appears to be attached to a structure.
- Mineralization: Three rocks contain sacrificial malachite and/or iron oxidation. They have about 2-4% chalcopyrite /pyrite. Most of the mineralization appears to be pyrite 70-80% and the chalcopyrite being 30-20% mineralization.

Drill Hole: MY-07-04

From: To:
20.14m 21.26m

Notes: Black/grey shale with varying dip changes. Some areas in longer drill sections and others are brecciated, one from faulting at 21.20m.

Lithology: Black to Grey Shale

Structure: Some bedding 0 (20.14-20.42m), 30 (20.71-21.11m), and 50 (21.01-21.26 m) degrees off the drill angle. Shale is brecciated from 20.42-20.74m, and 21.19-31m

Alteration: None

Veining: Minor amounts of calcite veining most less than 1mm, with one 5mm veining is less than 1% of the section.

Mineralization: Found two spots of chalcopryite both less than 1mm squared. Mineralization minor.

Drill Hole: MY-07-04

From: To:
21.26 22.77

Notes: More fluvial rubble. Most are less than 3cm diameter.

Lithology: Weathered shale segments, fine grained silica sandstones, some iron oxidized segments.

Structure: Fluvial rounded segments, most less than 0.03m diameters with the largest 0.05m. One shale segment 0.08m long, with calcite in filled fracture point less than 0.002m.

Alteration: Fluvial weathering

Veining: One shale segment 0.08m long, with calcite in filled fracture point less than 0.002m. Other pebbles have veining that is 1mm and less cutting across them.

Mineralization: Only one visible area with a small chalcopryite/pyrite in a 1mm square area.

Drill Hole: MY-07-04

From: To:
22.77 24.30

Notes: Some micro-faulting, with the shale mostly

Lithology: Black and grey shale bedding, shale bedding has some calcite composition as fizzes with acid.

Structure: Most of the bedding is 75-70 degrees TCA. Parting at bed angles.

Alteration: Some of the veining has some silica content

Veining: Quartz carbonate veining in fracture points (less than 1%), most are less than 0.5mm with a few about 1mm.

Mineralization: Minor amounts of mineralization of chalcopryrite within calcite veining. Most are small pockets less than 1mm square, one at 22.92m.

Drill Hole: MY-07-04

From: To:
24.30 24.85

Notes: A section of heavy quartz carbonate veining, about 35% veining. Shale within section very brecciated suggesting fault zone. Veining varies in angle with no specific trend for TCA. A milled zone occurs at 22.56m.

Lithology: Black and grey shale and quartz carbonate veining.

Structure: Fault zone with brecciated shale with veining in fractures, with milled fault at 24.56m.

Alteration: Quartz carbonate infill.

Veining: About 35% quartz carbonate veined zone.

Mineralization: No apparent.

Drill Hole: MY-07-04

From: To:

24.85 35.56

Notes: Mostly grey shale with black sections. Some faulting and folding but on a minor scale. About 5-10% veining. Between two larger veins (each about 5-6cm wide) there is more mineralization of pyrite and chalcopyrite.

Lithology: Black and grey shale with some large veins cross cutting.

Structure: Most bedding at a 40-50 TCA. At 32.55m the bedding becomes more brecciated until 33.21m. The shale in this section has a higher calcite content and fizzes.

Alteration: Veining infill

Veining: Stringers within most of the section with two larger sections of about 5-6cm, one at 31.85m and the other 31.61m and another at 32.89m that is 1.25cm wide.

Mineralization: At 31.75 in a more brecciated zone between the two larger veined areas there is a 15cm section with a predominantly pyrite zone that has mineralization scattered within. Then at 32.07m there is a small stringer of chalcopyrite/pyrite only 5mm by 0.5mm. At 34.42m also larger pyrite mineralized bleb within the veining.

Drill Hole: MY-07-04

From: To:
35.56 38.2

Notes: A large brecciated zone of shale/fine silt with 40% veining. Little apparent mineralization through the area.

Lithology: Brecciated black and grey shale and quartz carbonate veining.

Structure: Brecciated zone with pieces avg about 3cm in diameter.

Alteration: Brecciation.

Veining: Veining has no apparent trend other than infill. About 40% of the zone. One section of 25cm has mostly veining with small pieces of included shale at 36.32m.

Mineralization: A small stringer of chalcopyrite at 37.05m. No other apparent mineralization.

Drill Hole: MY-07-04

From: To:
38.2 57.61

Notes: Most of the shale is bedded with areas of brecciation. The areas of brecciation tend to carry more of the mineralization. Clay seams are also present with one likely between 47.16 to 47.34m and the other 49.33 to 49.72m.

Lithology: Black and grey shale with quartz carbonate veining with bedding of 45-60 TCA. The lighter grey shale has a carbonate composition. Clay seam at 47.16m to 47.34m and between 49.33 and 49.72.

Structure: Shale bedding trends 45-60 TCA. There are two main clay seams where milling could have taken place as a cause of faulting.

Alteration: Grey bedding has higher carbonate composition. No major alterations other than so brecciation in 20-30cm sections.

Veining: Quartz carbonate veining. Veining varies from less than a mm stringers to 5cm. The larger veining tends to be parallel to the bedding. One main quartz carbonate bedding at 53.4m to 53.55m.

Mineralization: Two larger mineralized zones with one being about **20cm at 45.26m** consisting of pyrite/ chalcopyrite? blend spotty within the shale/veinin. The other zone is at 43.58m for 18cm. Other stringers at 46.33, 48.08, 50.31, 56.46, and 57.58m.

Assay Number	From (m)	To (m)	Cu PPM
465307	45.26	45.5	22.2

Final Depth: 57.61m

Drill Hole: MY-07-05 Claim: Missy N: 6485727 E: 363745 Final depth: 44.81m

AZ: 273 DIP: 52.5 EL: 1408m DHS: Oct. 4, 2007 DHF: Oct. 11, 2007

Logged by: David Peake Teched by: George Coetzee

Notes: To intersect vein one exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-05

From: To:
0 13.95
Notes: **Casing**

Drill Hole: MY-07-05

From: To:
0 16.27

Notes: Within the weathered glacial till there is quite a bit of calcite/silica blend (calcite prominent), ~10% calcitic silica. The high majority of the till rock consists of broken and fractured shale possibly from the contact between the till and the bedrock.

Lithology: Assorted glacial till rubble

Structure: Glacial rocks, pebbles to rocks (of 5-6cm). Mud and sand most likely also was incorporated but was washed out during drilling. Most of the rocks are fractured possibly from the drilling too.

Alteration: Glacial weathering and rounding, within the silica rich rock there is visible amounts of Fe oxidation.

Veining: Some pieces are from a larger vein structure most likely broken from.

Mineralization: A small section of iron oxidation.

Drill Hole: MY-07-05

From: 16.27
To: 22.36

Notes: Black/grey shale with varying dip changes 40-60 TCA variation. Variation from bedding to brecciated.

Lithology: Black to Grey Shale with some veining

Structure: At 22.1m milling until 22.36m. Also some brecciation cemented at 19.99m. Most of the bedding 60 TCA until after 20.20.63m when the bedding changes to 40 TCA.

Alteration: Milling at 22.1m.

Veining: Minor amounts of calcite veining most less than 1mm, with one 1cm.

Mineralization: Found two spots of chalcopyrite with some pyrite both less than 1mm squared at 20.09m and 20.37m. Mineralization minor.

Drill Hole: MY-07-05

From: To:
22.36 27.85

Notes: Most of the shale is bedded with large areas of brecciation. Brecciation carries mineralization.

Lithology: Light grey shale brecciated with quartz carbonate infilling. The lighter grey shale has a carbonate composition.

Structure: Brecciated zone

Alteration: Some potential milling throughout.

Veining: **Carbonate quartz veining** at about 25-35% of section from 23.47m to about 27.85m.

Mineralization: Quite a bit of **chalcopyrite/pyrite mineralization**, probably <4% where present. One section very prominent primary chalcopyrite mineralized zone about 1cm thick at 24.00m and another 19cm chalcopyrite/pyrite blend zone cutting long ways across the core/vein at 26.16m to 27.17m and then another 1cm bleb at 22.60m. Primary mineralization also quite prominent throughout in blebs and incorporated within the brecciation at 27.80m, 24.16m, 23.12m, 24.87m. The hole crosscuts vein no 1

Drill Hole: MY-07-05

From: 27.85 To: 32.36
Notes: Most of the shale is bedded with large areas of brecciation. Brecciation carries minor mineralization.
Lithology: Light grey shale brecciated with quartz carbonate infilling. The lighter grey shale has a carbonate composition.
Structure: Brecciated zone
Alteration: Some potential milling throughout. Major milling at 32.25m and 32.01m. Clays within milled areas and surround regions.
Veining: Carbonate quartz veining at about 25-35% of section towards upper regions and then at about 29.45m to 29.70m fracturing.
Mineralization: One heavily chalcopyrite/pyrite mineralized area from **30.95m to 31.09m**. Minor areas of mineralization in a few spots but in small blebs and less than 1mm stringers, like at 30.05m.

Assay Number	From (m)	To (m)	Cu PPM
4653012	30.95	31.09	21.4

Drill Hole: MY-07-05

From: To:

32.36 44.81

Notes: Black/grey shale with varying dip changes 15-40 TCA variation. Mostly bedded with a few clay spots.

Lithology: Black to Grey Shale with minor veining and minor mineralization. A 5cm long piece of dyke material at 35.66m.

Structure: Most of the area bedded 15-40 TCA. One piece of 5cm green dyke material at 35.66 m, seems rounded and has a minor amount of Fe oxidation. A few faulted zones a major cemented one at 34.69m.

Alteration: Faulting at 34.69 other than that no major alterations.

Veining: Minor amounts of calcite stringers most less than 1mm. One veins less than 1cm cuts across the core horizontally for about 30cm.

Mineralization: Found minor amounts of chalcopyrite both less than 1mm squared or incorporated in shale like in 32.8m to 33.30m and 23.73m.
Mineralization minor.

Final Depth: 44.81m

Drill Hole: MY-07-06 Claim: Missy N: 6485727 E: 363745 Final depth: 15.85m

AZ: 273 DIP: 45 EL: 1408m DHS: Oct. 12, 2007 DHF: Oct. 13, 2007

Logged by: David Peake Tched by: George Coetzee

Notes: To intersect vein one exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-06

From: To:

0 15.24

Notes: **Casing**

Drill Hole: MY-07-06

From: To:
0 15.85

Notes: Assorted river rocks of shale (a couple with veining) and sand.

Lithology: Assorted River Rock rubble

Structure: Fluvial rocks, pebbles to larger boulders (of 8cm). Sand in the second section but believed to be added to make space and to demonstrate the material that came out of the wash..

Alteration: Fluvial weathering and rounding.

Veining: Some pieces have minor veining (less than 1%) incorporated or are part of a larger structure but no piece appears to be attached to a structure.

Mineralizat No apparent Cu but minor Fe oxidation.

Final Depth: 15.85m

Drill Hole: MY-07-07 Claim: Missy N: 6485727 E: 363745 Final depth: 61.57
AZ: 256 DIP: 52.5 EL: 1408 DHS: Oct. 15, 2007 DHF: Oct. 18, 2007
Logged by: David Peake Teched by: George Coetzee

Notes: To intersect the three veins exposed on the south-east slope of the Missy Knoll.

Drill Hole: MY-07-07

From: To:
0 19.81

Notes: **Casing**

Drill Hole: MY-07-07

From: To:
0 17.5

Notes: There is a 0.17m section, with the rock types variable from shale, dolomitic limestone, and then some of the reddish and greenish rocks possibly from the conglomerate from higher elevations. Some minor veining in a shale breccia but no major vein systems.

Lithology: Assorted River Rock rubble

Structure: Fluvial rocks, pebbles to larger boulders (of 23cm). Mud and sand most likely also was incorporated but was washed out during drilling.

Alteration: Fluvial weathering and rounding with some minor Fe oxidation.

Veining: Some pieces have minor veining (less than 1%) incorporated or are part of a larger structure

Mineralization: No apparent major mineralization other than Fe oxidation.

Drill Hole: MY-07-07

From:

To:

17.5

27.62

Notes:

Black/grey shale with varying dip changes 20-50 TCA. Some veining that does carry some pyrite/chalco mineralization.

Lithology:

Black to Grey Shale

Structure:

Thrust bedding that varies 20-50 TCA. Some vein stringers and cross cut and follow bedding planes. At the 23.19m to 23.39 sections the vein occurred in a brecciated region.

Alteration:

No major

Veining:

Minor amounts of calcitic silica veining most less than 1mm, with one 0.5cm veining cutting along the core angle from 23.19m to 23.39m. Another vein follows the bedding and is blebbed at 1cm wide at 50 TCA.

Mineralization:

Some minor and then more concentrated areas of chalco and pyrite mineralization. Two smaller more minor chalco stringers at 19.45m and 20.39m. The more concentrated areas at 23.39m being 0.5cm by 4cm long appearing to be primarily pyrite with some chalcopyrite. The other segment consists of two blebs cutting across the core at 25.64m being mostly pyrite, and 2cm bleb at 23.99m and then another chalco/pyrite blends at 24.91m.

Drill Hole: MY-07-07

From: To:
27.62 37.75

Notes: A shale region of brecciation and veining. The heaviest mineralization in the drill hole occurs in this region .

Lithology: Brecciated shale with veining altering from brecciation and bedding about every 0.33m to 0.5m. Where the lithology is not brecciated the bedding varies from 50-60 TCA. A segment of large vein is 0.48m long. One small region at about 32.05m appears to be more weathered, with it including a very small piece of green dyke material but only about 2cm squared. There are some other rocks accompanying it that could be more of a limestone/ankerite.

Structure: Brecciated shale with bedding altering every 0.33m to 0.5m. Some milling apparent at 29.72m, 33.39m (>10cm), 34.34m (>15cm long) and 36.65m (>20cm).

Alteration: Infilling of calcite/silica blend with some carrying chalco/pyrite blend.

Veining: Major veining occurs in region with many stringers throughout ≤ 1 mm. Some of the more major veining occurs at 31.27m-31.75m and 36.72m-36.88m. Other sections of veining +/-5cm occurs at 29.47m, 30.03m, 32.41. There is a long stringer about 1cm wide spanning from 37.32m-37.75m.

Mineralization: A more heavily Cu/Fe mineralized zone, although no apparent secondary mineralization. Some minor apparent mineralized areas in blebs and stringers less than 0.5cm squared at 29.19m (pyrite), 29.57m (pyrite), 30.42m (chalco), 32.41m (chalco), 36.75m (chalco), 37.37m (chalco), 37.49m (chalco/pyrite) and 37.62m (chalco/pyrite). The largest area of mineralization takes place between/within the vein of **31.27m and 31.75m** with two sections about 2cm squared the mineralization appears to be primarily chalcopyrite.

Assay Number	From (m)	To (m)	Cu PPM
4653012	31.27	31.75	1890

Drill Hole: MY-07-07

From: To:
37.75 61.57

Notes: A predominately monotonous shale area with minor alterations or veining.

Lithology: Black and grey shale bedding with minor amounts of veining (mostly stringers).

Structure: Most of the bedding is 20-30 degrees TCA.

Alteration: No major alterations.

Veining: Minor stringers <1mm throughout but not as common as the upper regions of the drill hole. One 3cm vein cuts about at 40TCA at 44.56m. Some of the larger stringers about 0.5cm to 1cm wide occur at 38.91m and then 48.05m.

Mineralization: Some minor amounts of mineralization mostly carried within the stringers. Most appear to have a more higher pyrite content then chalcopryite ratio. There stringers with mineralization mostly of pyrite with some chalco occur at 42.15, 42.6, 42.93, 45.72, 45.79, 45.83, and 47.71m depths.

Final Depth: 61.57m

Missy Drill Hole Core Recoveries and RQD

MY-07-01

Recovery and RQD

From:	To:	REC	RQD	%REC	Length
0	7.01	0.44	0	6	7.01
7.01	10.06	0.82	0.49	27	3.05
10.06	13.11	0.24	0	8	3.05
13.11	16.15	0.5	0.21	16	3.04
16.15	19.2	0.26	0	9	3.05
19.2	22.25	0.16	0	5	3.05
22.25	25.3	0.98	0.39	32	3.05
25.3	28.35	0.44	0	14	3.05
28.35	31.4	0.26	0.19	9	3.05
31.4	34.44	0.99	0	33	3.04
34.44	37.5	0.12	0	4	3.06

MY-07-02

Core and RQD recovery

From:	To:	REC	RQD	%REC	Length
0	10.06	0.63	0.21	6	10.06
10.06	16.15	0.75	0.17	12	6.09
16.15	19.2	0.48	0	16	3.05
19.2	22.25	0.66	0.26	22	3.05
22.25	25.3	0.17	0	6	3.05
25.3	28.35	0.19	0	6	3.05
28.35	31.39	0.98	0.21	32	3.04
31.39	32	0.23	0.13	38	0.61

MY-07-03

Core and RQD recovery

From:	To:	REC	RQD	%REC	Distance
0	3.05	0		0	3.05
3.05	11.58	0.39	0	5	8.53
11.58	13.72	0.25	0	12	2.14
13.72	15.85	0.36	0	17	2.13
15.85	16.76	0.82	0.51	90	0.91
16.76	17.98	0.72	0.43	59	1.22
17.98	18.9	0.15	0.1	16	0.92
18.9	19.81	0.17	0.13	19	0.91
19.81	20.73	0.08	0	9	0.92
20.73	23.17	0.18	0	7	2.44
23.17	25.3	0.71	0.54	33	2.13

MY-07-04

Core and RQD recovery

From:	To:	REC	RQD	%REC	Distance
0	20.42	0.74	0.39	4	20.42
20.42	23.47	1.88	0.93	62	3.05
23.47	29.57	2.23	1.77	37	6.1
29.57	32.61	2.61	2.1	86	3.04
32.61	35.66	2.53	2.18	83	3.05
35.66	38.71	2.41	2.25	79	3.05
38.71	41.76	2.83	2.83	93	3.05
41.76	44.81	2.59	2.32	85	3.05
44.81	47.85	2.76	2.68	91	3.04
47.85	50.90	2.45	2.21	80	3.05
50.90	53.95	2.28	2.1	75	3.05
53.95	57.00	1.94	1.16	64	3.05
57.00	57.61	0.61	0.36	100	0.61

MY-07-05

Core and RQD recovery

From:	To:	REC	RQD	%REC	Distance
0	14.33	0.16	0	1	14.33
14.33	17.37	1.9	0.48	63	3.04
17.37	20.42	2.77	2.47	91	3.05
20.42	23.47	2.45	2.03	80	3.05
23.47	26.52	3.03	1.77	99	3.05
26.52	29.57	2.44	1.68	80	3.05
29.57	32.61	2.66	1.69	88	3.04
32.61	35.66	2.77	2.65	91	3.05
35.66	38.71	2.94	2.84	96	3.05
38.71	41.76	2.9	2.71	95	3.05
41.76	44.81	2.74	2.18	90	3.05

MY-07-06

Core and RQD recovery

From:	To:	REC	RQD	%REC	Distance
0	14.33	0.24	0	2	14.33
14.33	15.85	0.4	0	26	1.52

MY-07-07

Core and RQD recovery

From:	To:	REC	RQD	%REC	Distance
0	14.33	0.34	0	2	14.33
14.33	17.37	0.76	0.28	25	3.04
17.37	20.42	2.48	1.45	81	3.05
20.42	23.47	2.62	2.24	86	3.05
23.47	26.52	2.63	2.43	86	3.05
26.52	29.57	2.88	2.82	94	3.05
29.57	32.61	2.7	2.08	89	3.04
32.61	35.66	2.36	0.99	77	3.05
35.66	38.71	2.6	1.93	85	3.05
38.71	41.76	2.84	2.68	93	3.05
41.76	44.81	3.01	2.85	99	3.05
44.81	47.85	2.93	2.64	96	3.04
47.85	50.9	3.04	3.04	100	3.05
50.9	53.95	2.9	2.81	95	3.05
53.95	57.00	2.81	2.81	92	3.05
57.00	60.05	2.29	1.95	75	3.05
60.05	61.57	1.52	1.16	100	1.52

Appendix F

APPENDIX F Missy Northeast Extension Mapping

