TRENCHING AND DRILLING REPORT

on the

METS 1 AND 2 MINERAL CLAIMS

LAIRD AND OMINECA MINING DIVISIONS BRITISH COLUMBIA

on behalf of MANSON CREEK RESOURCES LTD.

GEOLOGICAL BRANCH ASSESSMENT REPORT

16,692 PART I = OF 1 FILMED

TRENCHING AND DRILLING REPORT

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METS 1 AND 2 MINERAL CLAIMS

LIARD AND OMINECA MINING DIVISIONS

BRITISH COLUMBIA

NTS 94 E / 6 W

LATITUDE 57° 21' NORTH LONGITUDE 127° 19' WEST

DECEMBER, 1987

on behalf of

MANSON CREEK RESOURCES LTD. CALGARY, ALBERTA

and

GOLDEN RULE RESOURCES LTD. CALGARY, ALBERTA

by

B. T. EVANS, P.GEOL.

Golden Rule Resources Ltd. #410, 1122 - 4th Street S.W. Calgary, Alberta T2R 1M1

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CERTIFICATE

- I, Bruce Thomas Evans, of 120 Strathdale Close S.W., in the City of Calgary, in the Province of Alberta, do hereby certify that:
- I am a Senior Exploration Geologist with the firm of Golden Rule Resources Ltd., with offices at #410, 1122-4th Street S.W., Calgary, Alberta;
- 2. I am a graduate of Queen's University, B.S.C. (Honours) Geological Science (1982) and have practiced my profession continuously since graduation;
- 3. I am a member in good standing of the Association of Professional Engineers, Geologists, and Geophysicists of Alberta;
- 4. Exploration work conducted on the Mets 1 and 2 Mineral Claims during 1987 was done so under my supervision. Work and recommendations described herein are based upon my interpretation and observations of the Mets Property and knowledge of the region;
- 5. I do not own and do not expect to receive any interest (direct, indirect, or contingent) in the property described herein, and securities I own of Golden Rule Resources Ltd. and Manson Creek Resources Ltd. are through Employee Stock Option Plans or private market acquisitions by myself.

Dated at Calgary, Alberta this $\frac{15}{15}$ day of December, 1987.

Respectfully submitted,

Bruce T. Evans, P.Geol.



1.0 INTRODUCTION

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During the 1987 Summer Exploration season, the Golden Rule Resources Ltd./Manson Creek Resources Ltd. Joint Venture conducted an integrated trenching and drilling program on the Mets 1 and 2 Mineral Claims.

In total, 720 metres (2,362') of backhoe trenching was performed and 6060 metres (19,877') of diamond drilling was completed in forty-one (41) boreholes.

The objective of the 1987 exploration was to enhance and extend reserves for the Mets "A" Zone deposit, and to continue exploration for economic gold mineralization elsewhere on the property.

2.0 PROPERTY STATUS AND OWNERSHIP

The Mets 1 and 2 Mineral Claims are currently under option to Manson Creek Resources Ltd. from Golden Rule Resources Ltd., both of Calgary, Alberta. The claims are described more specifically as follows:

<u>Claim Name</u>	<u>No. of Units</u>	<u>Record Number</u>	Date of Record
Mets 1 Mets 2	20 20	1253 1254	April 3, 1980 April 3, 1980
	40 ==		

3.0 LOCATION AND ACCESS

The Mets 1 and 2 Mineral Claims form a contiguous block located in N.T.S. map area 94E/6W, approximately 510 km (317 miles) northwest of Prince George (figure 1), in the Liard and Omineca Mining divisions. The approximate geographic coordinates of the claims are 57 deg 27' North latitude and 127 deg 19' West longitude (figure 2).

The claims are accessible from Smithers (300 km (186 miles) to the south) to the Sturdee Airstrip via fixed-wing aircraft, then by helicopter to the claims, a distance of 27 km (16.5 miles).

4.0 PHYSIOGRAPHY AND GLACIATION

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The claims lie within the Cassiar Mountains physiographic subdivision of the Interior Plateau. The region is entirely glaciated and is characterized by wide U-shaped drift-filled major valleys and deeply-cut V-shaped upland valleys. Mountain peaks in the area average 1980 m (6,495') ASL and rise fairly abruptly from the major valleys. The topography of areas underlain by Toodoggone volcanic rocks is usually considerably more subdued than areas underlain by Takla Group volcanic rocks.

The southern part of the Mets claims is located over flattopped horseshoe-shaped 1980+ m (6,495') ridges that slopes steeply down into a deeply-cut south-facing basin. The northand east-facing slopes form steep cirques and cliffs as a consequence of ice plucking and seasonal alpine glaciation. The northern part of the claim group is located over an area of relatively gently sloping low hills and ridges that are dissected by a number of tributary streams at the headwaters of the north



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GOLDEN RULE RESOURCES LTD. MANSON CREEK RESOURCES LTD.

LOCATION MAP METS PROPERTY, BRITISH COLUMBIA

Figure 1



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GOLDEN RULE RESOURCES LTD. MANSON CREEK RESOURCES LTD. TOODOGGONE GOLD CAMP BRITISH COLUMBIA fork of Metsantan Creek. Elevations here range from 1400 m (4,592') to 1700+ m (5,576') ASL. In some areas, 30- (100') to 60- m (200') deep, narrow canyons occur along the streams.

5.0 <u>REGIONAL GEOLOGY</u>

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During 1971-1975 the regional geology was mapped by the Geological Survey of Canada at a scale of 1:250,000 under the direction of Dr. H. Gabrielse, with the results published in 1977 as Open File 483.

The British Columbia Department of Mines carried out a mapping program in the Toodoggone area from 1971 to 1984 with a compilation (Preliminary Map 61) published at a scale of 1:50,000 in 1985. This mapping, under the direction of T. G. Schroeter, details the units of Toodoggone Volcanics which had become the focus of gold exploration in the district. The following description of the regional geology is excerpted from his 1981 report:

The Toodoggone area lies within the eastern margin in the Intermontaine Belt. The oldest rocks exposed are wedges of crystalline limestone more than 150 metres thick that have been correlated with the Asitka Group of Permian age. The next oldest rocks consist of andesitic flows and pyroclastic rocks including augitetremolite andesite porphyries and crystal and lapilli tuffs that belong to the Takla Group of Late Triassic age. The Omineca intrusions of Jurassic and Cretaceous age (potassium-argon age of 186 to 200 Ma obtained by the Geological Survey of Canada) range in composition from granodiorite to quartz monzonite. Some syenomonzonite bodies and quartz feldspar porphyry dykes may be feeders to the Toodoggone rocks which unconformably overlie theTakla Group. The "Toodoggone" volcanic (named informally by rocks Carter, 1971) are complexly intercalated volcanic and volcanic-sedimentary rocks of Early and Middle Jurassic

age, 500 metres or more in thickness, along the west flank of a northwesterly trending belt of 'basement' rocks at least 90 kilometres in length by 15 kilometres in width. A potassium-argon age of 186+/- 6 Ma was obtained by Carter (1971) for a hornblende separate from a sample collected from a volcanic sequence 14 kilometres southeast of Drybrough Peak. Four principal subdivisions of 'Toodoggone' rocks have been recognized:

- 1) Lower volcanic Division -- dominantly pyroclastic assemblage including purple agglomerate and grey to green to purple dacitic tuffs.
- 2) Division Middle volcanic -- an acidic assemblage including rhyolites, dacites, 'orange' crystal to lithic tuffs, and quartz feldspar porphyries; includes welded tuff. 'orange' colour of the tuffs resulted The from oxidation of the fine-grained matrix while the rock was still hot. A coeval period of explosive volcanism included the formation of 'laharic' units and intrusion of syenomonzonite bodies and dykes. This event was accompanied by explosive brecciation along zones of weakness, predominantly largescale faults and attendant splays, followed by silicification and deposition of precious and base metals to varying degrees in the breccias. Rounded fragments of Omineca intrusive rocks are rare components in Toodoggone tuffs.
- 3) Upper Volcanic-Intrusive division -- grey to green to maroon crystal tuffs and quartz-eye feldspar porphyries.
- 4) Upper Volcanic-Sedimentary Division-lacustrine sedimentary rocks (sometimes varved), stream bed deposits, and possible local fanglomerate deposits and interbedded tuff beds.

Many Toodoggone rocks have a matrix clouded with fine hematite dust implying a subaerial origin, however, some varieties may have accumulated in shallow water. The host rock for mineralization (division 2) is an orange to chocolate brown colored crystal tuff with varying minor amounts of lithic and vitric ash. Broken crystals of plagioclase and quartz are set in a finegrained 'hematized' matrix of quartz and feldspar. The exact chemical composition(s) and rock name(s) await chemical analyses. Carter (1971) determined the composition of a suite of rocks collected from the Toodoggone area to range from latites to dacites.

To the west, Upper Cretaceous to Tertiary pebble conglomerates and sandstones of the Lower Tango Creek Formation of the Sustut Group unconformably overlie both Takla Group volcanic rocks and Toodoggone volcanic rocks.

The structural setting probably was the most significant factor in allowing mineralizing solutions and vapours to migrate through the thick volcanic pile in the Toodoggone area. The entire area has been subjected to repeated and extensive normal block faulting from Jurassic to Tertiary time. It is postulated that a northwesterly trending line of volcanic centres along a gold-silver-rich 'province' marks major structural breaks, some extending for 60 kilometres or more (for example, McClair Creek system, Lawyers system). Prominent gossans are often associated with structural zones but many contain only pyrite; sulphides occur as disseminations and fracture fillings in Toodoggone and Takla Group rocks. Thrusting of Asitka Group limestones over Takla Group rocks probably occurred during Middle Jurassic time.

Today Toodoggone rocks display broad open folds with dips less than 25 deg. The Sustut Group sedimentary rocks have relatively flat dips and do not appear to have any major structural disruptions.

6.0 <u>PREVIOUS EXPLORATION</u>

Since the staking of the Mets 1 and 2 Mineral Claims in 1980, several small scale mapping and geochemical sampling programs have been performed by Golden Rule Resources Ltd. Strong lineal Au in soil anomalies led to the 1985 Golden Rule/Manson Creek backhoe trenching program which subsequently led to the discovery of the Mets "A" Zone in trench T85-05 (0.341 opt Au/4.0 m). As a follow-up to the trench discovery, a short (244 m (800')) three (3) hole diamond drill program was completed

during September 1985. The 1985 drill program was successful at determining an orientation (strike = 340 deg, dip = 75 to 80 deg W) for the "A" Zone and returned encouraging results in DDH M85-03 (0.299 opt au/11.0 m).

During the 1986 season, the Joint Venture continued its integrated trenching and drilling of the "A" Zone deposit. A total of 1650 m (4428') of drilling was completed at twenty (20) locations. Exploration on the Mets 1 and 2 claims was completed by September 20, 1986, at which point reserves for the "A" Zone stood at 92,000 tons, grading 0.33 opt Au.

Table 1

1986 DRILL PROGRAM RESULTS

		AZM				
<u>HÖLE #</u>	LOCATION	(DEGREES)	DIP	TD	INTERCEPT	GRADE (oz/ton)
86-01	0+00/20+00W	070	-60	303' 92 4'm	284.5' - 307.8'	0.010/23.3'
M-86-02	25+00S/10+00E	070	-50	232' 70 7m	112.5' - 125.0'	0.083/12.5'
86-03	12+50N/10+00W	070	-50	387' 117.96	182.5' - 205.0' m (55.61 - 62.48m	(3.01m) 0.260/22.5' (6.86m) 0.289(15')
M-86-04	25+00N/40+00E	250	-45	252'	(57.15 - 62.48m) (55.0' - 77.5')	0.289/15 (5.33m) 0.228/12.5'
86-05 G @ 252.	25+00N/30+00W .5')	070	-50	76.8m 357' 108.8m	(19.8 - 23.62m) 245.0' - 266.5' (74.68 - 81.33m)	(3.82m) 0.515/21.5' (6.55m)
(70.	. 90m)				(75.9 - 79.7m) (284.0' - 291.5')	0.151/7.5'
₩-86-06 G@169. (51	25+00S/05+00W .0') 5m)	070	-50	292' 89.0m	(36.36 - 38.85m) 163.0' - 186.0' (49.68 - 56.69m)	0.529/23' (6.86m)
M-86-07	25+00N/10+00W	070	-50	297' 90.5m	175.1' - 197.6' (53.37 - 60.23m)	0.545/22.5' (6.86m)

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E <u>)LE</u> #	LOCATION	(DEGREES)	DIP	TD	INTERCEPT	GRADE (oz/ton)
M-86-08 G@98. (29.	50+00S/10+00E 2') 93m)	070	-50	202' 61.56m	82.0' - 124.5' (25.0 - 37.9m) 94.5' - 117.0' (28.8 - 35.7m)	0.527/42.5' (12.9m) 0.887/22.5' (6.86m)
M-86-09	50+00S/5+00W	070	-50	232' 70.7m	94.5' - 109.5' (28.8 - 33.4m) 156.8' - 179.3' (47.79 - 54.65m 164.3' - 179.3'	$\begin{array}{c} 1.247/15' \\ (4.6m) \\ 0.414/22.5' \\ (6.86m) \\ 0.503/15' \\ \end{array}$
M-86-10	25+00S/20+00W	070	-50	327' 99.67m	(50.08 - 54.65m) 211.4' - 238.9' (64.43 - 72.82m)	(4.57m) (4.57m) 0.040/27.5' (8.38m)
N 86-11	25+00S/35+00W	070	-50	352' 107 29m	267.0' - 289.5' (81 38 - 88 24m	0.006/22.5
M 86-12	50+00S/20+00W	070	-50	277' 84.43m	201.0' - 221.0' (61.26 - 67.36m 211.0' - 221.0'	0.214/20') (6.1m) 0.308/10'
1 86-13	805/0+00	070	-50	217' 66 14m	(64.31 - 67.36m) 152.5' - 187.5' (46.48 - 57.15m)) (3.05m) 0.189/35') (10.67m)
M-86-14	100S/5+00W	070	-50	178' 54 25m	no economic :	intersection
N_86-15	80S/15+00W	070	-50	282' 85 95m	(205.0' - 227.5)	0.233/22.5'
M-86-16	100S/15+00W	070	-50	225' 68 58m	no economic	intersection
N= 86-17	25N/50+00W	070	-50	417' 127 1m	305.0' - 322.5' (93.0 - 98.3m)	0.042/17.5'
N-86-18	50N/5+00E	070	-50	174' 53 10m	no economic	intersection
M-86-19	50N/15+00W	070	-50	232' 70 71m	no economic	intersection
N 86-20	75N/5+00W	070	-50	177' 53.95m	no economic	intersection

7.0 PROPERTY GEOLOGY

7.1 Lithologies

i) <u>K-spar Andesite</u>

The oldest unit exposed on the Mets property has been termed previously as a Porphyritic Rhyolite by Golden Rule or as a Trachy-Andesite Porphyry by Schroeter in 1985. In appearance,

this unit has an aphanitic to fine crystalline groundmass which is dark green to red to purple in color and has 1 mm to 4 mm size K-spar phenocrysts which vary in color, and dependent upon alteration in color from white to red. The K-spar Andesite forms the footwall of the Mets "A" Zone.

ii) Porphyritic Dacite

Overlying the K-spar Andesite and making up the hanging wall unit of the "A" Zone is the Porphyritic Dacite. The Dacite unit is fine to medium grained, consists mainly of clear and colorless to white, 1 mm to 2 mm quartz phenocrysts in an aphanitic pink plagioclase groundmass.

iii) <u>Red Tuff</u>

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Exposed at one location on Metsantan Ridge and within the cirque on the western claim boundary is the Red Tuff. Intersected in drill hole above the K-spar Andesite and quite often intercalated with the Dacite unit, the Red Tuff appears as an aphanitic to very fine grained, well sorted unit. Red-brown in color, the tuff is composed mainly of very fine sized strands of quartz and plagioclase. Generally, the tuff is massive but in rare instances, it exhibits slight graded bedding with tops to the west.

iv) Lapilli Tuff/Agglomerate

This Tuff Agglomerate unit is intersected in the western most collared drill holes. The unit is light brown in color and has a fine grained groundmass of quartz and plagioclase. Lithic fragments of elongate and block shape are found throughout, fragments being 5 mm to 10 mm in size and composed mostly of Quartz and Tuff.

v) Quartz-Barite-Breccia

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The Quartz-Barite-Breccia is the host for gold mineralization within the "A" Zone. The Breccia consists of micro-brecciated to coarse (10 cm) fragments of quartz, barite, and porphyritic dacite. Fragments have been re-brecciated and cemented with quartz and barite. Quartz character ranges from chalcedonic to coarse crystalline, white to grey in color. Barite usually occurs as white to pink fine bladed crystals. Native gold is the primary ore mineral present with rare occurrences of electrum, argentite and tetrahedrite. The Breccia generally occurs at the K-spar/Dacite contact but in some instances will cross at both units.

7.2 <u>Alteration</u>

Propylitic alteration is the most common alteration type on the property and proximal to the "A" Zone. Hanging wall porphyritic dacites exhibit propylitic alteration throughout in degrees that can range from moderate to intense; intensity grades upward toward the breccia system. The propylitic envelope consists of an alteration package of chlorite and epidote and to a minor extent, rare calcite and pyrite.

Bleaching and Silicification of hanging wall dacites become evident within 5 m (16') to 20 m (66') of the breccia system after which alteration shifts to argillic, evidenced by the deteriation of the dacite plagioclase component. Adjacent to the breccia zone is an interval of advanced argillic alteration with an assemblage of clay minerals primarily sericite, saussurite and kaolinite. Argillic alteration occurs in both the hanging wall side and footwall side of the breccia system but primarily within the footwall side where the alteration envelope can range from 1 to 40 m in thickness.

Alteration of the footwall K-spar andesites is minimal and restricted to argillic alteration of brecciated andesites and to a minor extent, propylitic alteration. In rare instances, strong andesite and hematite alteration can be found proximal to the breccia zone.

Silicification of the breccia material is extensive and can be found throughout the breccia or at times as a silicified wall, bounding both the footwall and hanging wall sides of the breccia system. Generally, it is within the silicified intervals where the better grade Au mineralization occurs.

7.3 <u>Structure</u>

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The Mets property is situated within a setting which represents a volcanic collapse feature. Photo mosaics reveal a series of overprinted broad circular collapse features. Associated with each collapse are complex sets of ring and radial fractures. Structural trends of the EGX Bonanza, Thesis and BV Zones, coupled with that of the "A" Zone, suggests that these ore grade occurrences appear about radial fractures.

The structural trend of the "A" Zone, being 343 to 351 azimuth, with a 65 to 80 deg west dip. Localized flat, vertical and block faulting provide for minor displacements of the "A" Zone. Trending skew to the "A" Zone at approximately azimuth 290 deg is the "Red Fault". The displacement along the Red Fault is estimated to be 35 m (115') and is left lateral. The Red Fault is vertical and crosses the Baseline at section 75 north.

Between section 50N and 75N, the "A" Zone structure is truncated by the N75 Fault. Orientation of the N75 is mapped a azimuth 050 deg, dipping -77 deg south. The north 75 Fault represents one collapse ring structure with the north of the N75 Fault being the dropped block. Displacement vertically on the N75 Fault is 100 m (330') to 110 m (360'), displacement being mapped between diamond drill holes M86-19 and M87-58. Several linear structural features can be mapped on the Mets property, Golden Furlong and Porcupile occurring on two such structures; Porcupile at azimuth 340 deg and Golden Furlong at 020 deg.

7.4 Mineralization

Gold is the primary ore mineral of the "A" Zone breccia system. Gold occurs as free grains and flakes adjacent to fragments of quartz and barite within the breccia system. Gold grain size ranges from 0.005 mm to 2 mm.

To a minor extent, electrum, argentite and tetrahedrite are present but can only be detected in thin section. Anomalous silver intersections occur within the breccia system proximal to

the footwall contact and usually down section of the gold intersection.

Sulphide mineralization is essentially non-existent but in rare instances, up to 3% fine disseminated pyrite will occur.

8.0 <u>1987 DRILL PROGRAM</u>

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The 1987 Mets project drill program was designed and initiated to accomplish three (3) objectives:

- Delineate and extend current reserves within the "A" Zone deposit;
- 2) Continue testing of the Main Mets structure; and
- 3) Investigate alternative targets within the Mets claims.

In total, forty-one (41) boreholes were completed for a total of 6060 m (19,877') of drilling. Thirty-one (31) drillholes tested the "A" Zone and Main Mets structure, two (2) holes tested the North Zone T-6 showing, four (4) holes tested the 400 South showing (M87-31), and four (4) holes tested the newly discovered Footwall Zone between sections 125N and 350N.

The "A" Zone delineation drilling was performed on 25 m sections, except sections 37+50N and 62+50N (12.5 m) and holes were collared at 15 to 20 m centres along section. Exploration holes on the Main Mets structure were at 100 m centres. Generally, all holes were drilled at azimuth 070 deg and inclined at -45 to -60 deg TCA.

1987 DIAMOND DRILLING PROGRAM

BOREHOLE LOCATIONS

47.11	1	ГОСАТТ	Ö N				
I <u>DLE</u> #	NORTH	EAST	ELEV	<u>AZM</u>	DIP	<u>TD (m)</u>	TARGET
M87-21	27,137.33	20,612.08	1,739,69	070	-45 deg	154.50	North Zone/T-6
137-22	27,209,60	20,618,20	1.723.62	070	-45 deg	163.70	North Zone/T-6
N-17-23	26,515,47	20,940 21	1,926,41	070	-50 deg	84 40	Sec 805/A Zone
M87-24	26 499 47	20 889 83	1 921 33	070	-50 deg	120 40	Sec 805/A Zone
N 7-25	26 545 37	20,000.00	1 916 92	070	-15 deg	18 60	Sec 505/A 7one
1 17-26	26 526 24	20, 879 01	1,010.02	070	-50 deg	124 10	Sec $505/A$ Zone
M87-20	26,020.24	20,073.01	1 020 32	070	-50 deg	169 80	
M87-28	20,402.00	20,000.40	1 023 00	070	-60 dog	167.60	
N 7-20	20,403.43	20,072.33	1,920.09	070	-50 dog	137.00	Sectors/A Zone
N 87-30	20,007.01	20,354.40	1 005 83	070	-50 deg	127 10	
M87-31	26,203,68	20,000.70	1 906 15	070		111 00	
107 - 32	26 63/ 13	21,002.07	1,300.13	070	-50 deg	99 70	N75 7000
1 7 92	20,004.10	20,057.00	1,075.10 1 971 04	070	-50 deg	151 50	N75 Zono
M87-3/	20,020.02	20,014.37	1 995 00	070	-50 deg	203 30	$N75/G_{OO}$ 175N
MR97-95	26,030.30	20,714.00	1,055.00	070	-50 deg	172 80	$\frac{1}{2} \frac{1}{2} \frac{1}$
17-36	20,401.07	20,000.72	1 019 70	070	-65 dog	206 30	Sec BOS/A Zone
M97-97	20,401.07	20,000.72	1 019 61	070	-60 deg	200.00	Sec 505/A Zone
M87-38	26,000.01	20,000.01	1,010.01	070	-60 dog	178 00	
N 7-90	20,400.30	20,040.74	1 031 10	070	-50 deg	93 60	Sectors/A Zone
N 7-40	26 583 46	20,340.23	1 888 89	070	-50 deg	182 00	Sectoos/A Zone
M87/1	20,000.40	20,031.40	1 903 71	070	-60 deg	157 40	
N 7 - 40	20,100.00	20,374.40	1 010 83	070	-50 deg	107.40	
N 7-19	20,212.20	21,027.03	1 914 80	070	-50 deg	105 50	
M Q 7 = A A	26,230.21	21,000.03	1 996 74	070	-50 deg	203 60	$\frac{1}{2} \frac{1}{2} \frac{1}$
Mag7-15	26,500.14	20,701.37	1 866 61	070	-50 deg	108 50	Sec 20N/A 20He
N = 7 - 46	26,000.00	20,044.75 20,831,17	1 860 95	070	-60 dog	195 90	Sectoon/N75
M87 - 47	26,000.00	20,051.17 20,764,30	1 835 07	070	-50 dog	101 16	Sec125N/N75/
1107 -177	20,700.00	20,704.00	1,055.07	070	JO GEE	404.40	Footwall
7-48	26,609,13	20.872.40	1.886.35	070	-50 deg	170.20	A Zone
N 7-49	26,616,49	20,902 28	1,886 71	070	-50 deg	114 90	A Zone
M87-50	26,709.51	20,909,73	1,866.53	070	-45 deg	209.40	Sec $125N/$
		201000110	1,000,00	0.0	10 400	200,10	Footwall Zone
N 7-51	26,709,51	20.909.73	1.866 53	070	-50 deg	246 00	Sec 125N/
	20,100.01	201000110	1,000,00	0.0	00 400	210,00	Footwall Zone
M97-52	26,218,14	21.048.03	1915,40	070	-50 deg	93.60	Sec 400S
7-53	26,476,33	20,956,37	1,935,64	070	-50 deg	93.60	Sec1255/A Zone
Mo7-54	26,465.84	20,924,96	1,931,66	070	-50 deg	127.10	Sec125S/A Zone
M87-55	26,623.67	20,872.51	1,881.74	070	-50 deg	133.20	Sec 50N/A Zone
7-56	26,619.57	20,858.23	1,881.13	070	-55 deg	154.50	Sec 50N/A Zone
N 7-57	26,633.11	20,871.32	1,878.76	070	-50 deg	133.20	Sec62+5N/A Zone
M87-58	26,625.02	20,814.97	1,871.04	070	-60 deg	185.00	Sec 75N/N75
M87-59	26,942.00	21,025.71	1,748.94	250	-50 deg	185	Sec300N/Footwall
N 7-60	Redrill	of first 40	m of M87-	59			

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Table 2

All diamond drill holes intersected the Main Mets structure with the exception of M87-34, M87-21 and M87-22, which were testing the North Zone and M87-50, M87-51, M87-59 and M87-60, which tested the Footwall Zone.

All drill core recovered is stored on the Mets property.

8.1 <u>Analytical Methods</u>

All split drill core and trench channel and grab samples were assayed for Au and Ag by conventional Fire Assay Methods. Selected samples or intervals were analyzed for a 31 element suite using an Inductively Coupled Plasma technique. Selected samples where coarse metallic gold was suspected were re-assayed using a metallic gold process.

Two assay labs were used during the 1987 season:

- 1) Eco-Tech Laboratories of Kamloops, B.C.; and
- 2) Min-En Laboratories of Vancouver, B.C.

In total, 1765 split core samples and 241 trench channel and grab samples were assayed. Results of drill core assays are in appendix 3, and trench data in appendix 4.

In addition to assay analyses, thirty-six (36) thin sections of selected drill core and grabs were prepared and described. Description of thin sections are included in appendix 6.

8.2 <u>Survey Control</u>

Horizontal, vertical and elevation survey control of drill hole collars and surface data points were performed by a three stage, photographic/optical surveying technique.

Initially, during September, 1986, Golden Rule Resources Ltd. shot 1:10,000 and 1:20,000 black and white air photographs of the entire Mets property and much of the surrounding ground. Control points were selected for these photos and then targeted. Once targeted, these points were surveyed by using conventional optical and RF transmission methods. Control point surveying gave correct horizontal, vertical and elevation data with which to reference the Mets property. Topographic maps scaled 1:5,000, 1:1,000 and 1:500 in selected areas were contoured digitally by a Wild AC-1 Stereo Plotter and computer system. Following map generation, all diamond drill hole collars were surveyed and coordinates digitized and subsequently plotted. Downhole surveys were completed by conventional acid and Sperry Sun methods.

During August of 1987, a second set of air photos were shot. This group being photographed in color. All targeted ground control points were checked and confirmed in the second photo set.

9.0 <u>RESULTS</u>

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9.1 <u>"A" Zone</u>

Reserve estimates for the "A" Zone were confirmed at 158,000 tons grading 0.33 opt Au, up from 92,000 tons in 1986.

"A" Zone reserves are contained within a quartz barite breccia that has a mineralized strike length of 140 m(459'), a true thickness of 6 to 10 m, and a vertical extent of up to 75 m

(246'). Orientation of the "A" Zone deposit is 340 azimuth with a 70 deg to 85 deg west dip. The "A" ore shoot exhibits a mild northwest plunge. "A" Zone breccias occur within the Toodoggone volcanics of early Jurassic age, at or near the vertical contact of K-spar andesite flows and an overlying dacite, tuff, agglomerate assemblage. All rock types are cross-cut by major and minor faulting of Jurassic to Tertiary age.

Gold mineralization occurs as free blebs and grains that range in size from 0.005 mm to 2.000 mm. Gold is generally found along quartz and barite fragment margins. Sulphide mineralization within the "A" Zone breccias is practically nonexistent.

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	Results from 1987 "A"	Zone Drilling	Table 3
<u>Hole #</u>	From-To (m)	<u>Au opt/m</u>	
M87-23	34.3 m to 39.3 m Sec 80S / "A" Zone	0.35 opt Au/5.0 m	
M87-40	151.5 m to 152.5 m Sec 25N / "A" Zone	0.106 opt Au/1.0	m
M87-48	79.70 m to 95.00 m Sec 37+50N / "A" Zone	0.244 opt Au/15.3	m
M87-49	47.2 m to 53.2 m	0.285 opt Au/6.0	m
M87-55	72.8 m to 79.9 m Sec 50N/"A" Zone	0.666 opt Au/7.1	m

The most apparent geological feature which controls gold grading within the "A" Zone is attitude and thickness of the breccia. Steep thin breccias generally are of economic grade throughout the "A" Zone. When breccia orientation flattens, such

as at depth and south of Section 80S, ore grade drops off rapidly to sub-economic. At the north end (Section 50N to 75N) of the "A" Zone, a cross cutting (azimuth 050) vertical fault, the N75 Fault, terminates ore grade material.

9.2 N75 Zone ("A" Zone Extension)

Deep drilling north of the N75 Fault on Section 75N has traced a vertical hanging wall/footwall contact to depth. M87deepest hole, intersected a quartz breccia at the 58, the contact. The breccia occurred over a narrow width (4 m) and was 0.860 g/t (0.025/4.0 m). The M87-58 geochemical and anomalous grade Au geological signature resembles strongly that of the breccia intersection in hole M86-19, the significance being M86-19 overcuts M87-55 (0.666 opt Au/7.1 m). The M87-58 occurrence suggests vertical movement on the N75 Fault of approximately 100 m (330'), orientation of the N75 Fault is azimuth 050 with a 78 deg south dip. The potential for continuation of "A" Zone mineralization in this area must be rated as high.

9.3 400 South Showing

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A gold occurrence was discovered at Section 400S in hole 8.030 g/t⁻ M87-31 (0.234 Gpt Au/1.0 m). Follow-up drilling on Section 400S 41149/t⁻ returned one intersection of 0.120 opt Au/1.6 m in hole M87-42, an overcut of M87-31. Gold mineralization at Section 400S is hosted within a narrow quartz breccia at the K-spar andesite/pyroclastic contact. Visible gold was observed in M8731. Follow-up drilling on this occurrence to date has proven discouraging.

9.4 Footwall Zone

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Backhoe trenching between Section 125N and 350N at an approximate elevation of 1760 m (5,773), uncovered a quartz carbonate breccia situated within footwall K-spar andesites. The footwall breccia has been exposed at three (3) points over a 260 m strike length (see map 2 for trench locations):

Table 4

<u>Trench</u> #	Sample #	<u>Au (opt)</u>	Ag (opt)	Type
T87-04 Sec 125N	(1985) 20918 4583	0.18 0.269 0.293		Grab 1.0 m channel 1.0 m channel
T87-05 Sec 300N	$\begin{array}{r} 4509 \\ 4511 \\ 4516 \\ 4543 \\ 4544 \\ 4546 \\ 4547 \end{array}$	$\begin{array}{c} 0.130 \\ 0.578 \\ 0.135 \\ 0.519 \\ 0.248 \\ 0.104 \\ 0.083 \end{array}$	$\begin{array}{c} 0.92 \\ 1.05 \\ 0.34 \\ 3.73 \\ 0.48 \\ 0.37 \\ 0.29 \end{array}$	1.0 m channel 1.0 m channel 1.0 m channel 1.0 m channel 1.0 m channel 1.0 m channel 1.0 m channel
T87-07 Sec 385N	4323	0.155	0.54	1.0 m channel

The Footwall Zone is distinct from the "A" Zone in that it is not emplaced at the andesite/pyroclastic contact but rather is hosted within the footwall K-spar andesites. The breccia has a large carbonate content, and the metal ratios are Ag:Au = 2:1 or greater versus Au:Ag = 10:1 or greater for the "A" Zone.

Orientation of the Footwall Zone is interpreted to be azimuth 340 with a indeterminate dip at this point. Thickness of the breccia occurrence is observed to pinch and swell with a maximum thickness observed on surface of 4.0 m (13'). Drill holes M87-50 and M87-51 tested the Footwall Zone on Section 125N under T87-04. Hole M87-50 intersected 0.563/0.70 m, a vertical 50 m (164') below T87-04, and M87-51 intersected 0.102/0.70 m, a vertical 70 m (230') below the M87-50 intersection (see Section 125N). Within Section 125N, Au mineralization exhibits a steep east dip. Drill holes M87-59 and M87-60 were drilled on Section 300N below T87-05 but failed to intersect the Footwall Breccia.

10.0 <u>RECOMMENDATIONS</u>

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A multiphase integrated trenching and drilling program is recommended for the Mets property. Also recommended, contingent on early success in 1988, is single level underground exploration and bulk sampling of the "A" Zone. Time table/or 1988 exploration would be as indicated in figure 3.

Equipment requirements to perform recommended work on the Mets property are essentially in place, with the exception of a small JKS 300 Diamond Drill to perform the Porcupile and Golden Furlong work.

Mobilization of fuel to the property will be performed during February and March, 1988. Fuel can be transported to Lawyer's Creek by tanker on the new Cheni Gold Mines Road and then north by Cat Train to the property.

Access to the camp and deposit areas should be completed by June 1, 1988 and drill site preparation for the N75 Zone drilling

completed by June 14, 1988. Deep drilling to test for the "A" Zone extensions north of the N75 Fault will be initiated on June 14, 1988 and contingent upon success, up to 3000 m (9,849') of drilling will be completed initially on the N75 Zone. Concurrent with drill start-up, trenching of the Footwall Zone will be started. Up to 250 hours of backhoe time will be used to ascertain the exact nature and orientation of the Footwall Zone. Contingent upon favorable trench exploration of the Footwall Zone, a 3000 m (9,840') diamond drill program will be initiated by mid July 1988 to test the Footwall Zone sub-surface.

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Exploration of the Golden Furlong and Porcupile showings will begin by the second week of July 1988 with backhoe trenching and then a follow-up 800 m (2624') drill program on each showing.

Contingent upon success of the N75 Zone and Footwall Zone, drilling a decision to drive an Adit on the 1850 m (6,068') level to the "A" Zone will be made by mid July 1988. A positive underground decision would require construction of a tote road from the Cheni Road to the project area for heavy equipment haulage. Road construction and equipment mobilization would lead to the 1850 m (6,068') level portal being collared by the first week of September 1988. Assessment and bulk sampling of the "A" Zone would require approximately 435 m (1,427') of underground workings. Time to complete the Adit, drift and cross-cuts is estimated to be 50 days. Sampling and mapping of the workings would be ongoing. Actual bulk sampling of the "A" Zone ore will be performed upon completion of the underground work.

Work recommended for the "A" Zone should be completed by October 31, 1988. Continued favorable results would provide for a prolonging of underground and diamond drill explorations of the Mets deposit, at which point, camp modifications will be necessary to prepare for the winter months.

A detailed outline of expenditures to complete this recommendation is included in appendix 1. To complete the project, it is estimated a total of \$2,261,746 would be required.

Respectfully submitted,

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Bruce T. Evans, P.Geol.



A Stand State



Figure 3

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APPENDIX 1

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PROPOSED 1988 EXPLORATION BUDGETS

	COST DESCRIPTION CO	ST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Personnel:			3 300	1 700		5.000
	Temporary Camp cost	Ψ	-	5,610	2,890		8,500
	Field cost		_	-	-	-	-
	Trans./Char.		-	-			_
Section Se	Travel Expense Exped /Warehouse		-	_			_
44.JF	Computer		-	-	-	-	-
	Contingency Audit		-	-	_		-
F	SUBTOTAL	\$		15,840	8,160		24,000
	Contract Work:						
67 770	Drilling	\$		94,050	48,450	-	142,500
	Core Analyses Trench/Stripp		-	3,960	2,040	-	6,000
18.3	Geochem/Anal.		-		-	_	
	Line Cutting		-	-	-	_ '	 :
	Grid Re-est		-	-	-	-	
	Drill Road/Site Prep.		-	-	_	_	_
節驚	Land Survey		-	-		-	
	Fuel/Haul.			1,485	765		2,250
	SUBTOTAL	\$	_	99,495	51,255	_	150,750
8 .3	TOTAL	\$		115,335	59,415	_	174,750
	Indirect Cost Recovery:						1 050
	Direct @ 10% Contracts/\$200K @ 10%			693 9 949	357 5 126	_	1,050
	Contracts >\$200K @ 5%		-	0,040 	-		-
-	SUBTOTAL	\$		10,642	5,483		16,125
	TOTAL EXPENDITURE	\$	-	125,977	64,898	_	190,875
		==	=======				

N 75 Zone 1988 Exploration Budget

	COST DESCRIPTION CO	ST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Personnel: Permanent	 \$			5,000		5.000
	Temporary	Ψ	-	9,450	15,925	-	25,375
40 B	Field cost			3,225	12,900		16,125
4 5-3	Trans./Char.		-		-	-	
	Travel Expense		-	-			
	Exped./Warenouse Computer		_				-
	Contingency		-				
	Audit		-	-		-	-
	SUBTOTAL	 \$		12,675	33,825		46,500
	Contract Work:						
	Drilling	\$	-		142,500		142,500
	Core Analyses		-		5,000		5,000
	Trench/Stripp.			12,375	6,375		18,750
	Geochem/Anal.		-	1,000		_	1,000
	Grid Re-est		-	-		_	_
	Geophys. Surv.		-	-		_	-
<i>(6</i> .3)	Drill Road/Site Prep.					-	_
	Land Survey			-	-		-
	Fuel/Haul.			720	2,880		3,600
	SUBTOTAL	\$		14,095	156,755		170,850
	TOTAL	\$	-	26,770	190,580		217,350
		==	erris and the state with the same				
	Indirect Cost Recovery:						
	Direct @ 10%		-	322	1,290	. —	1,612
	Contracts<\$200K @ 10% Contracts >\$200K @ 5%		- -	172	15,038	-	15,210
	SUBTOTAL	\$		494	16,328	4444 4444 4444 4444 4444 4444 4444 4444 4444	16,822
	TOTAL EXPENDITURE	\$ ==		27,264	206,908		234,172

FootWall Zone 1988 Exploration Budget

Porcupile/Golden Furlong 1988 Exploration Budget

	COST DESCRIPTION CO	OST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Porconnol:						
	Permanent	\$	_		5,000	-	5,000
-	Temporary	Ψ	-		28,750		28,750
	Camp cost		_	_	17,250		17,250
لاستأ	Field cost		-	-	-		
_	Trans./Char.		-			-	-
資料	Travel Expense		-		-		-
	Exped./Warehouse			-	-	_	-
	Computer			-		-	-
	Contingency		-	-			
	Audit			_			
10 B	SUBTOTAL	\$	-	-	51,000	-	51,000
	Contract Work:						
	Drilling	\$	_		157 440	<u> </u>	157.440
F	Core Analyses	Ψ			6,000	_	6,000
	Trench/Stripp.		_		22,500		22,500
6	Geochem/Anal.		-		2,000	_	2,000
e i	Line Cutting				- -		
601 - 7	Grid Re-est		-	_		-	
	Geophys. Surv.			-	_		-
<i>বে</i> জন	Drill Road/Site Prep.		-	-		_	
10-1-1-1-1-1-1-1-1-1-1-1-1-1-1-1-1-1-1-	Land Survey				-	-	-
4.5	Fuel/Haul.			-	3,600	-	3,600
	SUBTOTAL	\$			191,540		191,540
	TOTAL	\$	-	-	242,540	-	242,540
		= =			Lange State styre benef benef state state ange state State Lange ange State ange		nano bako afin into star star tana tana dala atau tana atau atau star dala dala
	Indirect Cost Recovery:						
	Direct @ 10%				1,725		1,725
	Contracts<\$200K @ 10%	5	-	-	16,904	-	16,904
1946	Contracts >\$200K @ 5%	5	-	-	-		
		-					
\square	SUBTOTAL	\$	-		18,629	-	18,629
		¢	_	_	261 160	_	261 160
	TOTAL EXTENDITORE	₽ ==			201,109		201,105
17 D							

Belle Property 1988 Exploration Budget

	COST DESCRIPTION CO)ST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Personnel:)		dalar 1999) initi dalah initi dalam katu dalam dalam dalam 1999 (1999)
للتستظ	Permanent	\$	-		5,000	-	5,000
傳有	Temporary		-	-	25,375	-	25,375
	Camp cost		-	-	16,125	_	16,125
لاستا	Field cost				20 000		-
(FR)	Trans./Char.		-	_	30,000	_	30,000
	Fyped /Warehouse		_		-	_	_
191.20 191.20	Computer		-	-		-	_
新教	Contingency		-				
	Audit		-	-	-	_	
()	SUBTOTAL	\$	-		76,500		76,500
1.1	Contract Work:						
لقيقا	Drilling	\$	-	-	141,106		141,106
	Core Analyses	-	-	-	8,000	_	8,000
in i	Trench/Stripp.		-	-	18,750	-	18,750
1982, and	Geochem/Anal.		-		2,000	-	2,000
	Line Cutting		-	_	-		-
	Grid Re-est		-	-	-	-	_
	Geophys. Surv.		_	_	_	_	
	Land Survey			_		_	
	Fuel/Haul		_		2,700	-	2,700
	SUBTOTAL	\$	-	-	172,556	-	172,556
	TOTAL	\$		_	249,056	-	249,056
1		= =					
1	Tuding at Cast Banana						
	Diment @ 10%		_	_	4 612		4 612
2	Contracts $\$200K @ 10\%$	6	_		15.381		15.381
	Contracts >\$200K @ 5%	6	-	-	-	-	-
wie w		-					
	SUBTOTAL	\$	-	-	19,993		19,993
	TOTAL EXPENDITURE	\$			269,049	-	269,049
		==					

	COST DESCRIPTION CO	OST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Personnel:						
R 3 1	Permanent	\$	-	-	-		-
18.39 18.39	Temporary	7	-		17,750	-	17,750
ar a	Camp cost		-	-	5,325		5,325
	Field cost		-		_		_
	Trans./Char.				33,600		33,600
	Travel Expense		-		-		_
	Exped./Warehouse		-	-	-		-
	Computer		-				-
	Contingency		-				
	Audit		-	••••	¹		-
1947-1949 1947-1949							
	SUBTOTAL	\$	-		56,675		56,675
87 9							
	Contract Work:						
	Drilling	\$	-	-	-		-
	Core Analyses		-	-	-	-	-
F 3	Trench/Stripp.		-	-	-	-	-
	Geochem/Anal.		-		-	-	-
	Line Cutting		-		-		-
	Grid Re-est				-	-	_
	Geophys. Surv.		-	-	-		
96.99	Drill Road/Site Prep.		-	-	56,050	-	56,050
6 3	Land Survey		-	-	-	-	-
	Fuel/Haul.		-	-	5,850	-	5,850
	Building Material			-	5,600	-	5,600
	Maintenance			-	10,000	-	10,000
	SUBTOTAL	\$	-		77,500	-	77,500
1	TOTAL	\$		_	134,175		134,175
		==		danda ditan peter tetab tetan dang bada adap tetan tetan dang dang		the star the same and an are	
	Indirect Cost Recovery:						
	Direct @ 10%		_	_	3 892	_	3 892
	Contracts ≤ 200 K @ 10%	ζ.	_	_	7,750		7 750
1000 ADV	Contracts $>$ \$200K @ 5%		-		-	-	-
<u>(</u>							
 ار.ع	SUBTOTAL	\$	-	-	11.642		11.642
		Ŧ					,
التعطام	TOTAL EXPENDITURE	\$	-		145,817	-	145,817
		==			arta data ante esta ante atra data		

Road 1988 Exploration Budget

	COST DESCRIPTION CO	ST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
	Personnel: Permanent Temporary Camp cost Field cost Trans./Char. Travel Expense Exped./Warehouse Computer Contingency Audit	\$			9,818 13,923 34,004 - - - - - - -	3,932 5,577 13,621 - - - - - -	13,750 19,500 47,625 - - - - - - - -
	SUBTOTAL	 \$			57,745	23,130	80,875
	Contract Work: Underground Core Analyses Trench/Stripp. Geochem/Anal. Line Cutting Grid Re-est Geophys. Surv. Drill Road/Site Prep. Land Survey Fuel/Haul. SUBTOTAL TOTAL	\$ \$			719,712 14,280 - - - 19,278 753,270 811,015	288,288 5,720 - - 7,722 301,730 324,860	1,008,000 20,000 - - 27,000 1,055,000 1,135,875
	Indirect Cost Recovery:				ita ita ita ita ita ita ita ani	6864 666 966 864 964 966 664 966	
	Direct @ 10% Contracts<\$200K @ 10% Contracts >\$200K @ 5%			-	3,400 3,356 35,986	1,362 1,344 14,414	4,762 4,700 50,400
أفريكا	SUBTOTAL	\$	-		42,742	17,120	59,862
	TOTAL EXPENDITURE	\$ ==:	-		853,757 ======	341,980 ======	1,195,737 =======

"A" Zone 1988 Exploration Budget

COST DESCRIPTION C	OST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
Personnel:						
Permanent	\$				-	_
Temporary		-	-	-	-	-
Camp cost		_		511-1	-	-
fleid Cost Trans (Char		-	8 111		1 222	37 998
Travel Expense		_	-	- 20,002	4,222	-
Exped. /Warehouse		_	_	-		
Computer		-	-	_	-	-
Contingency					-	-
Audit					-	
SUBTOTAL	\$	- .	8,444	25,332	4,222	37,998
Contract Work:						
Drilling	\$	_	_	_	_	
Core Analyses	Ψ	-			-	
Trench/Stripp.					-	_
Geochem/Anal.		-	-		-	_
Line Cutting		-	-	-		-
Grid Re-est		-		-	-	
Geophys. Surv.		-				-
Drill Road/Site Prep	•				-	-
Land Survey		-		_	_	40 000
Mobe/Demobe		40,000	8 888	26 664	Δ ΔΔΔ	39 996
Renovations		_	6,668	20,004	3,334	30,006
10110 (0 0 1 0 110						
SUBTOTAL	\$	40,000	15,556	46,668	7,778	110,002
TOTAL	\$	40,000	24,000	72,000	12,000	148,000
	==				term their term track track their fight	
Indirect Cost Recovery	:					
Direct @ 10%	•	-	844	2.534	422	3.800
Contracts<\$200K @ 10	%	4,000	1,556	4,666	778	11,000
Contracts >\$200K @ 5	%		_	-	_	
SUBTOTAL	\$	4,000	2,400	7,200	1,200	14,800
	*		00 100	80.000	10 000	100 000
TOTAL EXPENDITURE	\$	44,000	26,400	79,200	13,200	162,800
	= =					

General 1988 Exploration Budget

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Toodoggone Total 1988 Exploration Budget						
COST DESCRIPTION CO)ST	PHASE I 1st QTR	PHASE II 2nd QTR	PHASE III 3rd QTR	PHASE IV 4th QTR	TOTAL
Personnel: Permanent Temporary	\$	-	3,300 15,060 10 155	26,518 104,613 89 174	3,932 5,577 13 621	33,750 125,250 112,950
Field cost Trans./Char. Travel Expense			8,444	88,932	4,222	101,598
Exped./Warehouse Computer Contingency		-	-			
Audit SUBTOTAL	 \$		- 36,959	 309,237	 27,453	- 373,548
Contract Work UG Drilling Core Analyses Trench (Stripp	\$	-	 94,050 3,960 12 375	719,712489,49621,04047,625	288,288	1,008,000 583,546 25,000
Geochem/Anal. Line Cutting Grid Re-est		-	1,000	47,625	5,720	25,000
Equipment Maintenance Drill Road/Site Prep. Land Survey Eucl/Haul	•	- - 40.000	- - - 2 205	10,000 56,050 - 35,073	- - - 7 799	10,000 56,050 - 85,000
Mobe/Demobe Renovations Building Material			8,888 6,668	26,664 20,004 5,600	4,444 3,334	39,996 30,006 5,600
SUBTOTAL	 \$	40,000	129,146	1,449,544	309,508	1,928,198
TOTAL	\$ ==	40,000	166,105 ======	1,758,781	336,860 ======	2,301,746
Indirect Cost Recovery: Direct @ 10% Contracts<\$200K @ 10% Contracts >\$200K @ 5%		4,000	1,860 11,677 -	17,811 19,271 60,460	1,784 2,122 14,414	21,455 37,070 74,874
SUBTOTAL	\$	4,000	13,537	97,542	18,320	133,399
TOTAL EXPENDITURE	\$ ==	44,000 ======	179,642 ======	1,856,323	355,180 ======	2,435,145 =======

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1987 EXPLORATION EXPENDITURES

SUMMARY OF 1987 METS PROJECT EXPENDITURES

MAY THROUGH OCTOBER, 1987

TOTAL EXPENDITURES FOR PERIOD:	\$1, ===	183,056.92
CONTRACTOR - DRILLING COSTS		562,879.90
MISCELLANEOUS SUPPORT COSTS		36,832.15
DRAFTING/REPRODUCTIONS		3,564.55
EXPEDITING/FREIGHT/WAREHOUSE		9,432.50
TRENCHING/DRILL SITE PREPARATIONS / CAT COSTS		49,315.47
ASSAY/ANALYTICAL EXPENSES		43,895.30
TRAVEL EXPENSES		13,807.29
HELICOPTOR SUPPORT		104,755.24
FIXED WING SUPPORT		67,907.73
CAMP COSTS CAMP EQUIPMENT FOOD SUPPLIES FUEL		122,179.93
SURVEY COSTS AIR PHOTO WORK GROUND CONTROL		21,022.75
GEOLOGICAL COMPUTER COSTS		1,739.00
SUPPORT STAFF FIELD		99,374.59
SUPERVISORY GEOLOGICAL PERSONNEL	\$	46,350.52