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District Geologist, Smithers	Off	Confid	ential:	90.02.0	07
ASSESSMENT REPORT 18370 MINING DIVISION: On	nineca	a			
PROPERTY: Thorne LOCATION: LAT 56 49 00 LONG 126 38 00 UTM 09 6299263 644465 NTS 094D15E CLAIM(S): Thorne I-VI OPERATOR(S): Asamera Min (US) AUTHOR(S): McCarthy, P.D. REPORT YEAR: 1989, 58 Pages COMMODITIES					
SEARCHED FOR: Gold,Silver,Copper,Lead,Zinc KEYWORDS: Jurassic,Takla Formation,Quartz Veir Sphalerite,Galena	ns,Py	rite,Ch	alcopyr	ite	
WORK DONE: Geochemical, Prospecting PROS 2700.0 ha Map(s) - 1; Scale(s) - 1:10 000 ROCK 153 sample(s); ME Map(s) - 3; Scale(s) - 1:10 000 SILT 61 sample(s); ME Map(s) - 1; Scale(s) - 1:10 000					

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THORNE CLAIMS GEOCHEMISTRY AND PROSPECTING

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Omineca Mining Division

N.T.S. 94 D/15

Latitude 56° 49'N Longitude 126° 38'W U.T.M. 6299500 mN U.T.M. 644600 mE

SUB-RECORDER

RECEIVED

FEB 0 7 1989

VANCOUVER, B.C.

M.R. # _____\$____

by Paul McCarthy

Asamera Minerals Inc.

December 1988

Claim Name	Record No.	Units	Date Recorded
Thorne I Thorne II Thorne IV Thorne V Thorne VI	9515 9516 9517 9518 9788 9789	20 20 20 20 18 10	July 4, 1988 July 4, 1988 July 4, 1988 July 4, 1988 Sept.16, 1988 Sept.16, 1988

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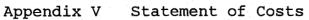
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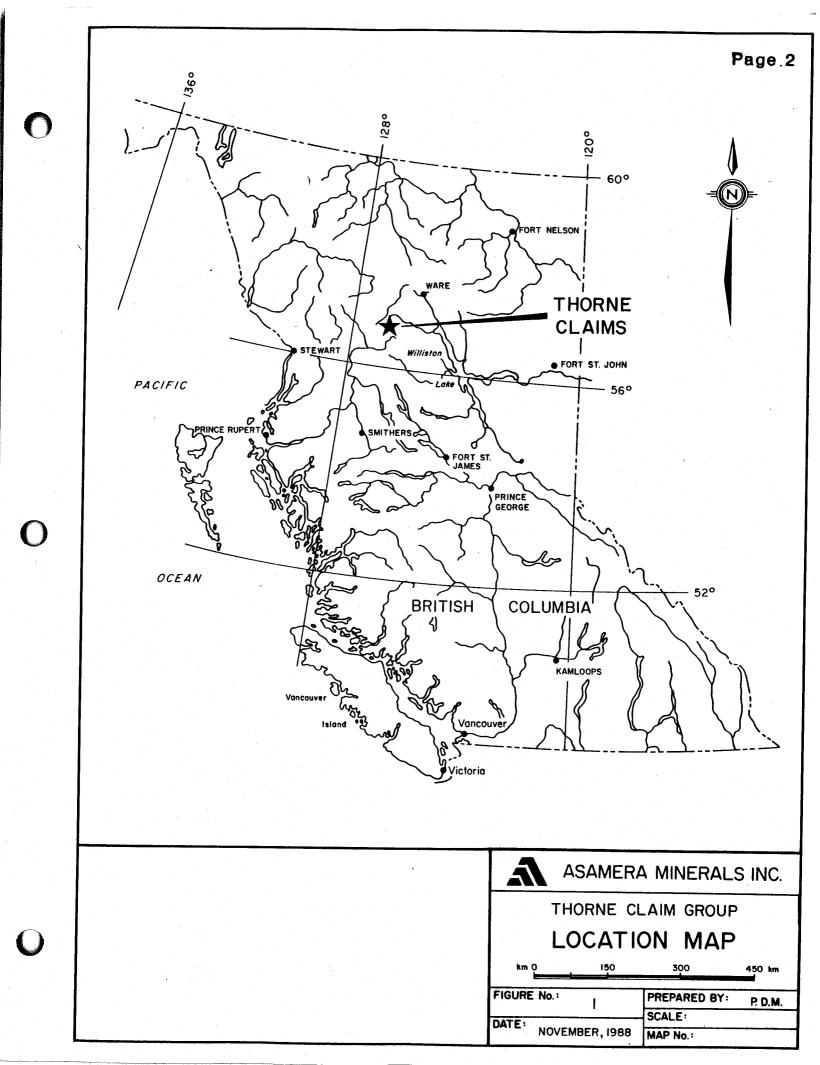


INTRODUCTION

This report summarizes the results of a prospecting and geochemical program conducted by Asamera Minerals Inc. on the Thorne claims in the McConnell Range, north-central British Columbia. The objective of the program was to localized the source of a gold and platinum anomaly detected in a heavy mineral sample during a regional reconnaissance sampling program conducted in 1987.

The program was carried out intermittently from June to late September, 1988 by a variety of individuals and consultants contracted by Asamera. Details of the personnel and their involvement in the work on the Thorne property are included in Appendix IV.

A total of 61 silt, and 153 rock samples were collected and sent to Acme Analytical Laboratories in Vancouver, British Columbia, where they were analyzed for gold, platinum, palladium, rhodium, plus 33 elements by ICP. Selected high-grade rock samples were assayed for more reliable results.



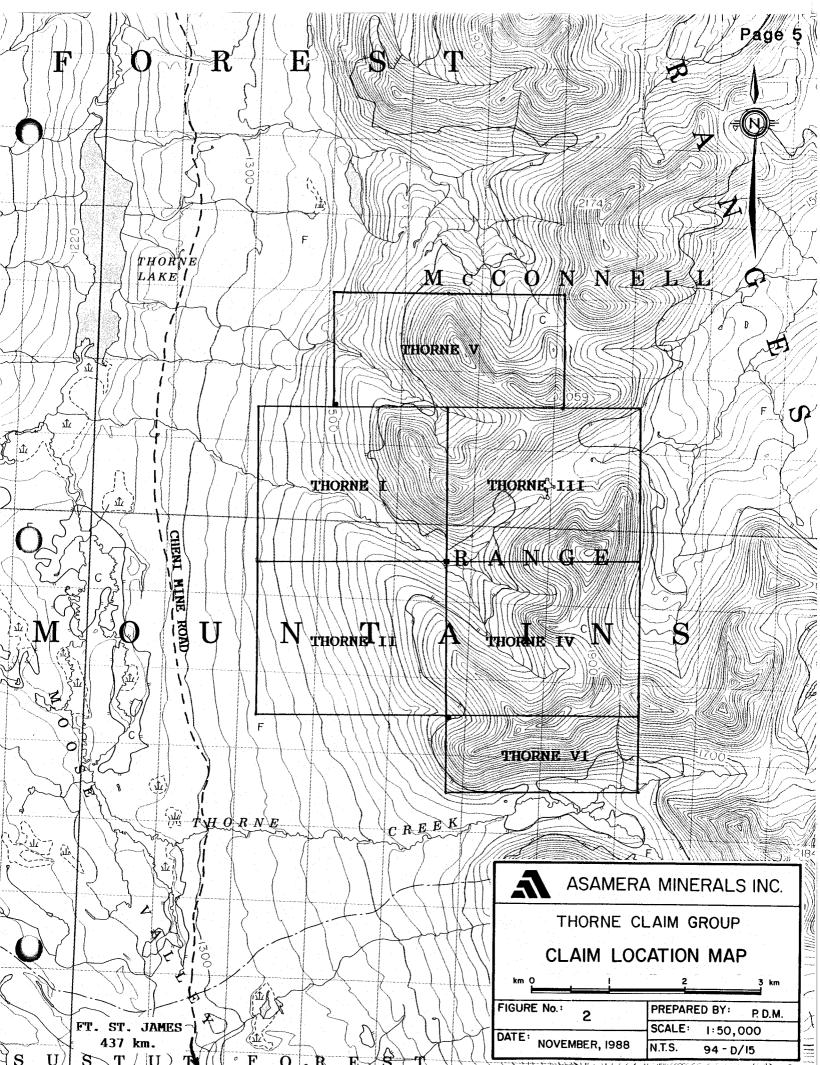
PROPERTY STATUS

The Thorne claim group consists of six mineral claims totalling 108 units or 2700 hectares. All claims are 100% owned by and registered to Asamera Minerals Inc. of Vancouver, British Columbia. Claim data are summarized below:

Claim N	ame		Record Number	Reco: <u>Da</u>	rding te	Mining <u>Division</u>
Thorne	I	20	9515	July	4,1988	Omineca
Thorne	II	20	9516	July	4,1988	Omineca
Thorne	III	20	9517	July	4,1988	Omineca
Thorne	IV	20	9518	July	4,1988	Omineca
Thorne	V	18	9788	Sept.	16,1988	Omineca
Thorne	VI	10	9789	Sept.	16,1988	Omineca

LOCATION AND ACCESS

Access to the claim area is by helicopter from Sturdee Strip, 42 kilometers to the northwest, which is on a regularly scheduled fixed-wing route from Smithers, 270 kilometers to the south (Figure 1). Alternatively, helicopter access can be gained from the Johanson Lake airstrip, 48 kilometers south of the claims. Johanson Lake is 349 kilometers north of Fort St. James along the Omineca Road, a gravelled road open seasonally (Figure 1). Cheni Mines Ltd. has recently constructed a mining road that starts from the terminus of the Omineca Road at Moose Valley, 35 kilometers north of Johanson Lake, and terminates at the Lawyers gold deposit in the southern Toodoggone area, not far from Sturdee Strip. The new road passes within four kilometers west of the centre of the property (Figure 2). Cheni Mines limited access along this road in 1988, and the Sturdee or Johanson routes were used during the 1988 field season. The geographic coordinates of the centre of the claims are 56° 49' north latitude and 126° 38' west longitude in N.T.S. 94 D/15E.



PHYSIOGRAPHY

The property covers the western slopes of the McConnell Range between Thorne Lake in the northwest and Thorne Creek at the southern margin of the claims (Figure 2). The relief on the property varies from 1400 metres to 2100 metres A.S.L. Thorne Creek drains the southern perimeter of the claims, and three unnamed permanently-flowing creeks drain the rest of the area: two creeks into Moose Valley in the west and another that drains to the east into McConnell Creek. The lowest elevations of the property are forested, primarily with spruce and lodgepole pine, gradually grading into sub-alpine vegetation, then alpine meadows above 1700 metres A.S.L. Three peaks dominate the terrain, with steep slopes of outcropping bedrock and scree on west-facing cirques above the meadows. The area has a long winter season, and several snowbanks persist throughout the year on the upper north-facing slopes. Geological fieldwork is generally restricted to the period between mid-June and late September.

The forests do not contain commercially-viable timber, and no plans have been formulated to log in the region.

EXPLORATION HISTORY

Remnants of tent frames and assorted wooden poles were noted in the southeast cirque and along the creek draining the central claim area, but no mention has been made of mining activity prior to 1983, when the area was partially staked, and the debris appear to predate this time.

In the early 1970's B.P. Minerals conducted a regional stream sampling program, following up on two copper-lead-zinc and gold-silver anomalies by staking the Goldway 11 and Goldway 12 claims over what is now Thorne VI and the southern half of Thorne IV, in May 1984. Silt, soil, talus fine, and rock samples were taken in the vicinity of two gossans on the ridge separating Thorne IV and Thorne VI, but the precious metal results were unfavourable, and the claims were allowed to lapse in 1988.

The area now covered by the Thorne I and Thorne V claims was previously covered by the eastern part of the Thor claim group, staked in October 1983 by the Asitka-Gunsteel Resource Corporation. Asitka-Gunsteel and Falconbridge Limited entered into an exploration agreement to determine the paleoplacer gold potential of the Sustut Basin sediments, which outcrop on the western edge of the Thorne I and Thorne V claims and elsewhere in the Moose Valley area to the south and west. A silt, soil, and rock sampling program was completed in summer 1984, but the economic gold-bearing potential of the Sustut sediments on the Thor claims proved to be low, and the claims were allowed to

lapse. However, high gold values were reported in the older Takla volcanic rocks in a shear zone now covered by Thorne I, and this showing was prospected and sampled by Asamera in 1988.

During a regional heavy mineral reconnaissance program conducted by Asamera in 1987, very high gold and platinum values were detected in the stream draining the centre of the property, and the Thorne I to Thorne IV claims were staked in June, 1988 to cover the anomalous drainage. Thorne V and Thorne VI were staked in September, 1988 to cover encouraging gold and silver results detected during the first prospecting trip at the north and south perimeters of the property. Work was terminated in late September due to heavy snowfall in the area.

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PROSPECTING

Regional Geology

The Thorne claims lie in the McConnell Range within the Takla volcanic belt, immediately to the north of the Hogem Batholith, a large, elongate, multiphasic intrusive complex extending from the Nation Lakes 170 kilometres northwest to Dortatelle Creek, 50 kilometres southeast of the claims. Several ultramafic, diorite, monzodiorite, and quartz monzonite stocks intrude the McConnell Range in the vicinity of the claims, and are lithologically similar to intrusions within the Hogem Batholith, probably intruding during the early modal phases of the Batholith in the Late Triassic (Garnett, 1978). The stocks are part of the Fleet Peak Pluton, a regional intrusive complex underlying the axis of the McConnell Range (Woodsworth, 1976). More felsic intrusive rocks, typically pink-coloured, mediumgrained granodiorite and felsic dykes outcrop throughout the east and south McConnell Range, including the central claim area. They are correlative with the Early Jurassic Omineca Intrusions in the northern Hogem Ranges and the Black Lake Intrusions of the southern Toodoggone adjacent to the northwest (Richards, 1988).

The stocks intruded Takla facies, locally the Savage Mountain Formation of Late Triassic (Norian) age, comprised of augite-feldspar porphyry flows, breccias, pillows, and aquagene tuffs, which were slightly metamorphosed during the intrusive

events (Monger,1976; Richards,1976). The youngest rocks in the region are siltstone, sandstone, conglomerate, and coal of the Sustut Group; basinal deposits of Late Cretaceous age.

The Hogem Batholith is bounded to the east by the Pinchi Fault, which bifurcates into several north to northwest trending faults at the north end of the batholith. Two major northnorthwest trending splays, the Ingenika and Moose Valley Faults bracket the claim area to the east and west respectively: the Moose Valley Fault traverses the western margin of the property.

Previous geology in the claim area, by the Asitka-Gunsteel Resources - Falconbridge group in the north (Lehtinen,1984), and BP - Selco in the south (Heberlein,1984) was augmented by reconnaissance mapping during the prospecting program on the property in 1988.

A compilation of the geology known to date is presented in Figure 3. Two major faults, the North Valley Fault and Main Valley Fault follow the west-flowing creek drainages in the north and central claim area. An oblique fault trending southwest from the North Valley Fault likely intersects the Main Valley Fault, although the confluence is obscured by talus and overburden. The claim area appears to have been separated into several blocks by steeply-dipping faults, as indicated in Figure 3. Several north-south and east-west trending shear zones extend away from the major valley faults in the north and central areas of the property. The shears have been infilled with quartz and\or calcite veining with sulphide mineralization, and were the targets of prospecting and sampling as described below.

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PROSPECTING AND ROCK CHIP SAMPLING

Prospecting and rock sampling was done by several groups over the 1988 field season. After the initial prospecting trip in mid June, further prospecting was suspended until the sample analyses were returned in order to determine the best target areas for followup sampling. Followup sampling was done over three field trips, thus there was an overlap in sampling as the most prospective regions were defined. A compilation map of the rock sample locations (Figure 5) has been divided into several "zones" or regions based on local disparities in mineralization and geology in order to facilitate the description of the sampling. Rock sample descriptions are included in Appendix II.

Zone A is in an area of north-trending shears, narrow, but persistent, and widening to the south. This showing was discovered by digging in an area of numerous limonitic rock chips, which overlie small quartz-carbonate veins in the fractures with pyrite-argillic alteration peripheral to the veins. Most samples were taken from float above the shears. Pyrite is the predominant sulphide mineral, with less common sphalerite, chalcopyrite, and galena. Several rock chips had malachite staining and contained hematite along with the sulphide mineralization.

An obliquely-offset fault originates from the North Valley Fault (Figure 3) at the north end of Zone B. The fault zone is over ten metres wide at this point, narrowing towards the top of

the mountain, but extending to the south, where it disappears under talus derived from the south-facing slope of the mountain. Samples 518 and 519 were anomalous in gold (see Figures 6 and 7 for analysis results), and channel samples were subsequently taken in trenches across the fault near the 518 and 519 locations in September (samples 587, 588, 590). The fault zone in this area is marked by a series of 20 to 35 centimetre wide shears with quartz-carbonate veining in a zone of weathered ferricrete and brecciated rock. The samples were from highly oxidized rock with remnant pyrite. Sample 3 from the talus apron at the southern end of the fault zone were limonitic and malachitestained, with chalcopyrite and pyrite in guartz-calcite veins and stringers.

Zone C appears to be a northeast-trending shear zone extending from the Main Valley Fault. At the southern end of the zone a large quartz vein, up to 60 centimetres wide was sampled (samples 509 to 513), and contains pyrite, sphalerite, with trace chalcopyrite. Further up the slope the veining pinches out into narrow quartz-calcite stringers in a mafic volcanic facies, and the samples were taken from malachite-stained rocks (samples 59 to 61).

The samples in Zone D were taken from several large quartzfeldspar veins in fractured volcanic rock that outcrops along the banks and on the bottom of the central tributary of the main creek on the property, within the Main Valley Fault zone. Massive pyrite, sphalerite, and some chalcopyrite were found in

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the quartz-feldspar veins and calcite stringers.

The southern part of the claim block contains a large gossan (Zone F) and a smaller gossanous area near the contact with the granodiorite intrusion (Zone E). The gossans are on a ridge extending along the southern side of a steep-sided cirque, and the initial rock samples were taken from float below the ridge and traced to the side-walls and ridge tops in followup prospecting.

Zone E is at the eastern contact of the granodiorite stock within or proximal to hornfelsed Takla volcaniclastic facies. A limonitic zone ten metres wide covers an east-west trending shear with quartz-carbonate veining in rocks containing hornblende, chlorite, garnet, and tourmaline: probably a skarn zone. The principal sulphide in the rocks is pyrite, with sphalerite, chalcopyrite, malachite, and hydrozincite. A small gossanous shear zone in the eastern part of Zone E may be connected with the larger gossan at the contact by an east-west fault, very hard to trace along the ridge. The samples in this area were taken from limonitic float with guartz-calcite stringers.

Zone F covers a large limonitic gossan, 150 metres in width (talus to ridge) on a steep northwest-facing wall of the cirque. The gossan covers a series of sub-parallel shears with amphibolepyrite and quartz-pyrite veining. Several east-west to southwest trending dacite to trachyte and augite-feldspar porphyry dykes traverse the area. The principal sulphide in the samples is pyrite (euhedral), with sphalerite, galena, and malachite

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staining in a quartz gangue.

The samples in Zone G were taken in pyritic rocks in a small area of quartz - calcite veining in a mafic tuff.

GEOCHEMISTRY

SILT SAMPLING PROCEDURE

A total of 61 silt samples was collected at 100 metre intervals along the active stream channel of the creek draining to the west from the centre of the property, as illustrated in Figure 4. The silt was put into gussetted kraft sample bags and sent to Acme Analytical Laboratories in Vancouver for analysis. Acme was instructed to analyze the samples for Au, Pt, Pd, Rh, Mo, Cu, Pb, Zn, Ag, Ni, Co, Mn, Fe, As, U, Th, Sr, Cd, Sb, Bi, V, Ca, P, La, Cr, Mg, Ba, Ti, B, Al, Na, K, and W.

SILT SAMPLE ANALYSIS METHODS

Sample Preparation

At Acme Laboratories the silt samples are dried at 60°C and sieved to -80 mesh.

Procedure for Gold, Platinum, Palladium, and Rhodium Analysis

Ten grams of material from each sample are fused with a silver niquart in a lead oxide flux. The noble metals are separated from the lead button in a cupellation furnace, and the doré beads dissolved in aqua regia. The resultant solution is analyzed by fire assay/mass spectrographic scan.

Procedure for Multi-Element Analysis

One-half gram of material from each sample is digested with

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three ml. of 3-1-2 hydrochloric acid - nitric acid - water at 95°C for one hour, and diluted to 10 ml. with water. This leach is partial for Mn, Fe, Sr, Ca, P, La, Cr, Mg, Ba, Ti, B, and W, and limited for Na, K, and Al. The resultant solution is analyzed by ICP.

7.3

ROCK CHIP SAMPLING PROCEDURE

A total of 153 rock samples were taken during prospecting traverses where mineralization was found in outcrop or in float, usually from rocks containing limonite, quartz and calcite veining, or altered minerals. Several channel samples were taken across shear zones in gossanous areas. Sample locations are mapped on Figure 5. The rock samples were collected in plastic sample bags and sent to Acme Analytical Laboratories in Vancouver, B.C. Acme was instructed to analyze the samples for Au, Pt, Pd, Rh, Mo, Cu, Pb, Zn, Ag, Ni, Co, Mn, Fe, As, U, Th, Sr, Cd, Sb, Bi, V, Ca, P, La, Cr, Mg, Ba, Ti, B, Al, Na, K, and W.

7.4

ROCK CHIP ANALYSIS METHODS

Sample Preparation

At Acme Labs the rock samples are crushed to -4.8 mm. and then 226.8 gm. is further pulverized to -100 mesh for analysis.

Procedure for Gold, Platinum, Palladium, and Rhodium Analysis

Ten grams of material from each sample are fused with a silver niquart in a lead oxide flux. The noble metals are separated from the lead button in a cupellation furnace, and the doré beads dissolved in aqua regia. The resultant solution is analyzed by fire assay and atomic absorption or mass spectrographic scan.

Procedure for Multi-Element Analysis

One-half gram of material from each sample is digested with three ml. of 3-1-2 hydrochloric acid - nitric acid - water at 95°C for one hour and diluted to 10 ml. with water. This leach is partial for Mn, Fe, Sr, Ca, P, La, Cr, Mg, Ba, Ti, B, and W, and limited for Na, K, and Al. The resultant solution is analyzed by ICP.

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RESULTS

8.1 Silt Sampling

Silt sample locations and gold, platinum, and palladium analyses are reported in Appendix Ia and on Figure 4. Platinum and palladium analyses were generally very low in the silt samples taken above the anomalous heavy mineral sample taken in 1987 from the stream draining the centre of the claims. Sample TST 011, in the south fork of the stream, with 17 ppb Pt, was the only sample with an elevated platinum group metal value.

Several elevated gold values were found in the three forks of the central creek. Samples TST 02, 06, 11, 15, and 16, in the southern fork, had over 25 ppb Au, and TST 015 with 240 ppb Au, had the highest silt value on the property. This stream drains below three gossanous shear zones discovered during prospecting, and the source of the gold in the silts likely derived from them. Samples TST 018, in the central fork, and TST 038, in the north fork also had gold values over 25 ppb. The area above these samples has had only cursory prospecting coverage, and the possible sources for the gold values remains to be discovered. Several other silt samples, in the main stream below those previously-mentioned, also had over 25 ppb gold. Some of the gold in these samples probably derived from the gold-bearing shear zones on the south-facing slopes above the creek (see section 8.2).

8.2 Rock Chip Sampling

Rock chip sample locations are mapped on Figure 5, and the results reported in Appendix Ib and Figures 6 and 7. Seventy rock samples were analyzed for platinum, palladium, and rhodium, but platinum and rhodium content in all samples was low. The only sample with elevated palladium content was THR 67, with 87 ppb, in a float rock sample containing magnetite and chalcopyrite, with malachite staining.

High gold and silver values were found in float and outcrop in several areas of the property. The gold and silver analyses indicate that there are two separate mineralized zones on the claims - a north gold-rich area (Zones A,B,C) within shears and fault structures associated with the two northwest-southeast trending faults in the west-flowing creek drainages, and a southern silver-base metals concentration associated with gossanous shear zones in the southern cirgue wall.

Most of the highest gold values (THR 07,13; TR 261,262; PS 519,590; 39561) were from samples at the northern edge of the claims, near the North Valley fault, in Zones A and B, although a few high gold values were also found in the southern claim area (PS 526,541). Many (but not all) of the high gold values in the northern samples had associated high arsenic assays, although arsenopyrite was not identified in the samples during field inspection.

The largest concentration of high silver values was detected at the southern edge of the claims (Zones E,F). The highest silver values were found in rocks with base metal values over 10,000 ppm, but were not specifically correlated with either copper, lead, or zinc. Almost every rock sample taken from the large gossan in the southeast corner of Thorne IV had high silver values - the three samples that had lower silver results (one with high gold) also had low base metal content. The small oxidized zone in the eastern part of Zone E nearest Zone F also had some silver-rich rocks, and several silver analyses over 35 ppm were detected in rocks within the northern claim area, mostly associated with high gold or base metals. Most silver-rich samples from the gossans in the southeast claim area had lower arsenic values as compared to those from the northern zones.

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CONCLUSIONS

9.0

Two principal areas of mineralization are apparent on the Thorne property. Very promising gold and some high silver and base metal results were detected in oxidized rocks containing sulphides in quartz and calcite veins and stringers in faults and shear zones, probably linked to large northwest-trending faults that traverse the north and central claim area. A large gossanous shear zone in the southeast claim area has high silver-base metals content with sporadic elevated gold values, and constitutes the other prominent mineralized area on the property. The structural or genetic link, if present, between the two mineralized areas is unknown.

The precious metals - base metals mineralization on the Thorne claims is encouraging enough to warrant followup geological and physical work leading to a drill program in 1989.

RECOMMENDATIONS

- The factors governing the formation of the two major areas of mineralization in the northwest and southeast claim area are unknown at this time. Detailed mapping should be done to determine the structural control over mineralization, the extent of mineralized structures, and the type of deposit.
- 2) Most anomalous gold, silver, and base metal results have derived from oxidized sulphide-bearing rocks, and trenching is needed to expose unoxidized rocks in the prospective areas. Channel samples should be taken across mineralized structures in shear zones and veins, both to determine the structural character of the mineralized zones, and their mineralogy. Several targets should be trenched:
 - i) the gold showing in Zone A.

- ii) several trenches across the fault in Zone B. Deep trenches should be dug at the northern terminus of the fault at the North Valley Fault to expose unoxidized rock, and in the talus-covered slope at the southern extension of the fault.
- iii) the shear? zone in Zone C.
- iv) the eastern part of Zone E, to determine the source of the high gold, silver, and base-metals values in locally derived float.
- v) the shear zone under the large gossan in Zone F.
- 3) Time constraints in 1988 limited prospecting in the far north and much of the eastern claim area. The region north of the North Valley fault (Thorne V) has obvious potential for further gold mineralization similar to the showings in Zones A and B, possibly in shears and fault zones obliquely trending from the main fault. The south-facing slope below

the ridge at the Thorne IV - Thorne VI boundary line is also prospective, given its proximity to the gossanous shear zones with high gold, silver, and base metal content. Silt samples in the central and north forks of the creek draining that prospecting should be done above these samples to determine the gold sources.

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A P P E N D I X I

Laboratory Reports

Ia: Silt Sample Results

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ACME ANALITICAL LABORATORIES LTD.

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GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR WN FE CA P LA CR MG BA TI B W AND LINITED FOR NA K AND AL. AU DETECTION LINIT BY ICP IS 3 PPM. - SAMPLE TYPE: P1-P2 SOIL P3-P4 ROCK AU** PT** PD** & RH** BY FA-MS.

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DATE RECEIVED: JUNE 27 1988 DATE REPORT MAILED: ASSAYER. .. D. TOYE OR C. LEONG, CERTIFIED B.C. ASSAYERS ASAMERA MINERALS INC. File # 88-2275 Page 1 SAMPLE Mo Cu Pb Zn Ag Ní Co Nn Fe As U Au Th Sr Cđ Sb Bí V Ca P La Cr Ma Ba Ti B Al Na ĸ W Auss pres pres pher PPN PPN PPN PPM PPN PPN PPN PPM 8 PPM PPM PPM PPM PPM PPM PPM PPM PPM % % PPN PPN \$ PPN % PPM 1 1 3 PPN PPB PPB PPB PPB TST 001 95 11 74 637 4.35 1 . 1 - 19 15 11 71 107 1.31 .058 5 ND 1 1 2 4 7 47 1.00 74 .16 8 2.24 .02 .05 4 2 TST 002 1 109 84 .2 21 18 701 4.85 13 5 ND 1 80 2 2 2 119 1.47 .065 8 51 1.08 83 .17 4 2.42 .02 .07 1 47 6 2 TST 003 78 .2 627 4.71 12 1 96 B 20 16 5 ND 71 2 4 116 1.38 .056 6 50 1.04 74 1 1 .17 2 2.29 .02 .05 1 6 3 2 3 TST 004 1 124 1 78 .2 20 18 671 4.54 15 5 ND 1 84 1 2 3 110 1.62 .063 6 48 1.10 76 .16 2 2.57 .02 .07 1 11 2 5 2 TST 005 1 141 13 79 .4 22 21 648 5.14 16 5 ND 2 82 2 2 2 128 1.63 .069 7 60 1.12 79 .17 4 2.54 .03 .07 1 13 2 TST 006 1 122 13 88 .2 20 18 648 4.76 15 -5 ND 1 82 2 2 117 1.60 .069 1 52 1.09 83 .17 1 4 2.52 .02 .07 39 2 3 2 7ST 007 103 ß 76 .3 20 17 640 5.08 12 1 6 ND 1 77 2 2 2 126 1.50 .066 7 57 1.06 80 .17 8 2.39 .02 .06 2 13 2 **TST 008** 10 79 .1 19 18 665 1.72 13 1 117 5 ND 1 89 2 1 2 112 1.68 .070 1 49 1.08 90 .16 5 2.65 .02 .07 54 10 2 1 2 TST 009 1 119 13 79 .2 19 17 676 4.28 12 5 ND 86 7 1 1 2 2 104 1.58 .064 44 1.10 91 .17 2 2.63 .02 .07 1 10 2 2 4 11 TST 010 1 117 11 .3 20 17 684 4.17 13 5 ND 1 90 1 2 4 99 1.62 .061 1 41 1.09 92 .16 4 2.64 .02 .07 10 2 6 2 1 TST 011 1 109 77 .2 19 16 644 4.37 10 ND 88 - 5 1 1 2 2 105 1.64 .063 6 43 1.06 95 .16 8 2.60 .02 .07 35 17 2 1 2 TST 012 101 11 .2 19 639 4.74 12 8 16 5 ND 2 85 116 1.60 .067 1 1 2 4 1 48 1.05 84 .16 10 2.47 .02 .07 11 2 Ł 2 TST 013 92 10 90 .2 17 14 559 5.31 8 5 ND 1 11 2 2 132 1.53 .056 1 2 7 55 1.04 78 .17 10 2.33 .02 .06 6 1 2 3 2 TST 014 6 79 .2 19 12 89 1 109 16 644 4.32 5 ND 1 1 2 2 104 1.61 .059 6 42 1.09 94 .16 2 2.57 .02 .07 2 6 1 4 2 TST 015 94 1 80 .1 19 15 597 5.82 10 1 5 ND 1 84 2 2 2 143 1.58 .063 6 60 1.02 85 .17 2 2.42 .02 .07 240 1 6 2 TST 016 105 78 21 5 1 6 .1 16 643 5.05 - 14 ND 2 88 1 2 2 124 1.62 .072 1 53 1.10 97 .16 11 2.53 .02 .07 28 2 2 TST 017 52 3 72 .1 18 12 637 4,96 7 5 ND 68 1 1 1 2 2 130 1.42 .052 8 62 .94 69 .17 2 2.10 .02 .05 1 3 3 1 2 7ST 018 55 8 71 .1 19 13 642 4.35 8 5 ND 1 73 2 2 111 1.43 .052 .93 1 1 1 53 77 .15 12 2.23 . 02 .05 1 99 2 2 TST 019 1 61 9 73 .1 19 12 636 4.22 8 5 ND 1 73 1 2 107 1.46 .052 6 50 .96 2 72 .15 2 2.27 .02 .06 1 3 2 2 2 TST 020 1 55 1 73 .1 19 12 623 4.71 1 5 ND 1 73 2 2 122 1.50 .052 1 58 1 .90 68 .16 6 2.20 .02 .06 2 1 1 6 TS7 021 9 1 60 71 .1 18 13 628 4.15 -5 ND 71 106 1.43 .050 .92 -7 1 2 2 8 49 69 .15 2 2.19 1 .01 .06 2 TST 022 11 10 82 .1 21 1 14 730 4.88 12 5 ND 1 85 1 2 4 122 1.70 .054 8 57 1.04 79 .16 8 2.65 .02 .07 1 2 4 3 4 TST 023 70 15 73 20 693 4.26 15 79 1 .1 14 5 ND 1 2 3 1 106 1.56 .053 7 51 . 97 77 .15 2 2.43 .02 .07 1 .22 2 3 2 TST 024 1 74 19 77 .2 20 15 710 4.48 11 5 ND 1 -82 1 2 2 110 1.58 .054 1 51 .98 87 .15 .02 12 2 2.51 .07 1 2 4 2 TST 025 60 10 68 20 13 734 4.03 83 1 .1 9 5 ND 1 2 2 95 1.57 .054 9 43 .93 96 .13 2 2.43 .02 .08 1 2 3 2 **TST 030** 1 50 8 68 .1 15 456 3.18 9 6 5 ND 64 1 2 90 1.11 .057 8 39 .74 1 2 17 .12 7 2.10 .01 .06 1 22 TST 031 1 81 4 81 .2 19 12 633 3.80 7 6 ND 1 11 1 2 2 101 1.28 .056 8 46 .98 95 .07 .11 2 2.84 .01 1 R 2 TST 032 74 15 76 .2 18 12 622 3.78 5 71 1 8 ND 1 1 3 3 101 1.16 .056 8 46 .97 87 .11 2 2.58 .01 .06 1 3 1 5 2 **TST 033** 75 11 72 .2 18 12 571 3.64 1 5 ND 71 2 2 96 1.17 .055 1 1 2 8 43 .95 93 .11 11 2.63 .01 .06 1 17 2 4 2 **TST 034** 79 9 79 .3 19 12 584 3.75 5 ND 74 1 6 1 1 2 2 97 1.23 .057 8 46 1.00 97 .11 4 2.73 .01 ..07 5 2 TST 035 696 3.67 1 71 79 .1 18 12 -5 70 2 - 3 97 1.18 .056 8 45 .94 92 .11 4 2.56 .01 . 06 1 - 3 2 - 3 2 7ST 036 1 68 8 71 .1 17 11 621 3.49 1 5 ND 1 67 1 2 2 96 1.20 .055 8 45 . 89 93 .06 .10 2 2.45 .01 1 2 TST 037 6 95 .1 21 15 994 4.53 5 82 1 74 9 ND 1 2 2 2 119 1.39 .055 8 49 1.10 93 .14 7 2.76 .02 .07 2 2 2 5 2 TST 038 70 1 106 .1 21 17 1426 5.17 12 5 ND 1 91 2 2 134 1.59 .055 1 2 7 54 1.25 94 .15 5 2.95 .02 .07 32 1 3 3 ż 2 TST 039 1 58 1 95 .1 21 15 889 4.65 6 5 ND 1 66 2 2 120 1.40 .052 6 44 1.44 66 .21 2 2.39 .03 .07 3 2 TST 040 1 56 9 96 20 14 301 4.67 ND 1 70 2 2 2 122 1.29 .053 6 46 1.28 60 .19 .1 1 5 6 2.22 .03 .08 1 1 - 2 2 - 2 STD C/FA-5X 18 58 39 132 6.6 67 29 1058 4.20 41 19 37 47 18 16 18 58 .50 .089 40 57 .93 179 .07 7 38 1.98 .07 .14 14 97 103 100 24

	. (O									ASI	MER	А МІ	NER	ALS	INC	. (E #	88-	2275	5										. 1		re 2
SANFLE‡	No PPN	Cu PPM	Pb PPN	Zn PPN	Ag PPM	NÍ PPM	Co PPN	Nn PPH	Fe 3	As PPN	U PPN	Au PPN	Th PPN	Sr PPN	Cd PPN	SD PPM	Bi PPM	V PPN	Ca \$	P	La PPN	Cr PPH	Ng B	Ba PPN	Ti t	B PPM	Al t	Na ł	X t	W PPM	Au** PPB	Pt** PPB	Pd** PPB	Rb** PP8
TST 041 TST 042 TST 043 TST 044 TST 045	1 1 1 1	55 69 74 51 64	9 11 14 9 22	75 111 105 70 85	.1 .1 .2 .1 .1	17 17 17 17 20	11 14 14 13 15	814 941 669	4.15 4.68 4.30 4.73 5.32	5 6 5 7	5 5 5 5 5	ND ND ND ND ND	2 1 2 2 1	65 77 85 72 86	2 2 3 2	2 2 2 2 2	2 2 2 2 2	129 115 127	1.44 1.59 1.36	.048 .051 .052 .051 .051	8 8 8 7	58 56 47 61 67	.79 .94 .95 .89 1.13	85 93 96 79 74	.16 .17 .16 .19 .21	2 3 4	2.13 2.63 2.90 2.21 2.80	.01 .02 .02 .02 .02	.07 .07 .08 .07 .08	1 1 1 1 1	7 8 7 85 45	4 3 4 3	5 5 2 3 2	2 2 2 2 2 2
TST 046 TST 047 TST 048 TST 049 TST 050	1 1 1 1	76 58 72 54 72	14 15 12 11 11	70 74 67 60 71	.1 .1 .2 .1	15 15 17 16 18	14 13 14 12 14	608 625 564	5.83 4.11 5.06 3.99 6.86	8 5 8 6	5 5 5 5 5	ND ND ND ND ND	1 1 2 2 2	81 89 80 73 76	1 1 2 2	2 2 2 2 2 2	2 2 2 2 2	98 130 102	1.60 1.44	.057 .051 .053 .053 .063	8 6 7 7 9	67 40 57 44 86	.98 .97 1.03 .95 .94	93 87 84 80 78	.20 .11 .20 .18 .21	2 2 3	2.43 2.70 2.52 2.27 2.27	.02 .02 .02 .02 .02	.07 .07 .07 .07 .07	1 1 1 1 1	10 6 31 6	3 4 3 2 4	2 4 2 2 2	2 2 2 2 2 2
TST 051 TST 052 TST 053 TST 054 TST 055	1 1 1 1 1	83 70 77 73 70	20 11 15 11 6	75 68 68 66 65	.1 .1 .3 .3	19 18 17 17 16	15 14 14 14 14	632 642 629	4.55 4.94 4.60 4.25 4.22	8 6 8 7 5	5 5 5 5 5	ND ND ND ND ND	6 2 1 2 1	90 83 82 82 81	1 2 1 2 1	2 2 2 3 2	2 2 2 2 2	127 118 109	1.60 1.56 1.55	.062 .055 .055 .055 .055	8 8 8 7	56 51 46	1.08 1.02 1.01 1.01 1.00	98 91 95 94 92	.20 .20 .19 .18 .18	13 6 7	2.68 2.51 2.46 2.43 2.45	.03 .03 .02 .02 .02	.08 .07 .07 .08 .07	1 1 2 2 1	9 8 51 5	3 4 3 3 3	3 -2 3 3 2	2 2 2 2 2
TST 056 TST 057 TST 358 TST 059 TST 060	1 1 1 1	46 76 60 51 57	12 12 14 19 12	58 66 62 61 62	.1 .1 .1 .1	19 16 17 17 16	12 14 13 12 12	641	4.36	4 6 7 4 5	5 5 5 5 5	ND ND ND ND ND	2 1 1 1 3	71 83 78 72 67	1 1 1 2	2 2 2 2 2 2	2 2 3 2 2	108 101 114	1.55 1.46 1.40	.052 .055 .053 .052 .054	9 8 9 9	52 46 44 53 76	.82 1.01 .94 .91 .87	109 99 100 90 84	.18 .18 .18 .19 .20	5 2 3	2.10 2.50 2.29 2.20 2.01	.02 .02 .02 .02 .02	.08 .08 .07 .07 .07	1 1 1 1 2	4 4 3 11 18	3 2 3 2 3	2 2 2 2 2 2 2	2 2 2 2 2 2
TST 061 TST 062 TST 063 TST 064 TST 065	1 1 1 1	56 59 52 56 64	11 14 14 13 21	65 65 58 60 65	.2 .1 .1 .1	17 18 16 16 18	13 13 12 12 13	670 631 623	5.47 5.27 4.15 4.62 5.38	7 4 4 7 6	5 5 5 5 5	ND ND ND ND ND	2 1 2 1 2	71 74 72 70 73	1 1 2 1	2 2 2 2 2 2	2 2 3 2 2	137 108 120	1.38 1.36 1.33	.052 .052 .052 .054 .055	9 8 8 9	73 63 49 57 66	.89 .94 .91 .90 .93	87 93 91 89 94	.19 .19 .18 .18 .19	4 2 8	2.12 2.21 2.15 2.16 2.22	.02 .02 .02 .02 .02 .02	.07 .07 .07 .07 .07	1 1 1 1	78 3 5 8 2	2 3 3 3 3	2 2 3 3 3	2 2 2 2 2 2
STD C/FA-51	17	57	38	132	7.1	67	28	1054	3.99	40	15	8	35	45	16	17	20	55	.47	.085	38	55	.91	173	.07	35	1.90	.06	.14	12	96	101	98	24

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APPENDIX I

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Laboratory Reports

Ib: Rock Sample Results

			ASAMERA	MINERALS IN	NC. OILE # 88-2275		ge 3
SAMPLE	NO CU PD Zn PPN PPN PPN PPN	Ag NI Co Mu Fe As PPM PPM PPM & PPM				Ba Ti B Al Na X W Au PPN & PPN & & & PPN PI	rs Pt+s Pd+s Rb+s PB PPB PPB PPB
THR 01 THR 02 THR 03 THR 04 THR 05	1 195 13 2167 1 26 21 749 1 296 269 10428 1 3191 8 63 1 585 15 96	1.5 17 13 2230 3.79 11 .3 3 2071 1.32 30 2.8 7 9 1923 2.53 88 4.0 47 19 350 5.22 4 1.0 18 37 491 36.88 4	0 5 ND 8 5 ND 4 5 ND	1 17 9 1 25 105 1 55 1		$\begin{array}{cccccccccccccccccccccccccccccccccccc$	23 3 3 2 29 4 2 2 37 2 3 2 93 4 8 2 2 2 5 2
THR C6 THR C7 THR C8 THR J9 THR 10		.1 29 22 1103 5.83 3 38.7 3 4 280 25.81 6313 8.4 373 121 206 21.37 28 1.1 11 3 262 2.56 156 .1 20 17 1011 5.69 8	3 5 ND 9 5 ND 5 ND	2 48 56 3 109 3 1 18 1	8 39 108 .26 .045 5 38 .11 2 3 55 .72 .028 2 100 .89 2 3 29 1.75 .003 2 54 .68	4 .05 5 .89 .01 .04 3 900 5 .06 10 1.11 .01 .05 1 46 1 .02 5 .62 .01 .01 1 900	10 3 10 2 64 9 22 2 65 6 23 2 98 2 2 2 8 3 5 2
THR 11 THR 12 THR 13 THR 14 THR 15	14 540 512 2127 3 293 130 2061 42 452 21387 1581 20 497 21478 2915 20 20699 203 223	7.9 8 6 1952 16.40 1521 22.9 1 1 151 21.72 2797 36.9 2 4 507 20.06 7114	5 ND 6 3 5 2	2 33 13 4 16 8 1 2 65 94 13	2 3 120 .49 .047 2 46 1.14 10 2 84 .08 .026 2 13 .11 12 8 68 .19 .031 2 11 .14	2 .07 4 1.13 .01 .93 1 81 7 .15 2 2.69 .01 .08 1 66 4 .03 2 .73 .01 .10 1 414 5 .03 6 .89 .01 .07 3 162 3 .27 4 3.09 .01 .01 1 335	69 2 14 2 44 2 6 2 22 2 6 2
THR 15 THR 17 THR 19 THR 19 THR 20	302 7258 264 866 4 1 7830 95 218 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3 3	33.0 84 31 1985 6.08 38 .4 1 1 53 2.62 54 23.2 14 5 165 12.02 2002	5 ND 5 ND 5 ND	1 54 4 9 10 3 3 2 3 1 21	2 3 105 5.64 .039 2 305 2.02 3 2 1 .03 .003 41 1 .02 21 28 58 26 .03 .025 8 16 .10 1	2 .02 2 .54 .01 .01 1 200 4 .05 3 2.73 .01 .03 1 112 17 .01 2 .35 .01 .16 1 2 19 .01 2 .47 .01 .10 1 372 12 .01 6 .96 .01 .06 2 111	22 B 13 2 22 2 2 2 2 28 2 6 2
THR 21 THR 22 THR 23 THR 25 THR 25		34.9 1 4 87 16.76 931 .5 8 18 2525 5.69 10 1.4 12 405 4.70 9 1.1 2 2 157 2.25 14 .5 2 5 3648 2.23 7	5 ND 5 ND	1 76 7 2 1 107 1 2 1 10 1 2	2 2 155 4.34 .049 3 10 1.77 2 2 98 1.91 .053 2 12 1.20 1 2 2 2.71 .102 2 1 .12 3	10 .04 3 .39 .01 .06 1 92 8 .19 2 6.38 .01 .07 1 17 .16 4 2.66 .13 .08 2 5 35 .01 8 .38 .03 .11 1 3 65 .01 2 .95 .01 .05 1	6 2 3 2 56 2 2 2
THR 27 THR 28 THR 29 THR 30 THR 31	1 120 15 253 2 406 7 19 4 15 26 100 1 187 8 49 1 15 20 133	.2 4 7 3153 3.97 29 .2 29 27 231 3.93 20 .1 5 4 496 1.45 17 .1 3 12 586 5.21 2 .1 10 5 497 13.40 2	5 ND 5 ND	1 4 1 2	2 2 58 1.20 .054 2 18 .77 1. 2 2 25 .15 .008 2 9 .65 6 2 2 89 1.01 .080 3 6 1.35 15		1 2 2 2 1 3 4 2 1 2 2 2 1 2 2 2 1 2 2 2 1 2 2 2 1 2 2 2 1 2 2 2 1 2 2 2
THR 32 THR 33 THR 34 THR 35 THR 36	18 152 88 2132	.5 7 7 384 9.28 7 .5 3 10 57 3.18 5 22.7 1 8 812 4.50 78 1.0 14 12 762 2.91 20 3.8 12 6 391 2.88 210	6 ND	3 33 1 2 3 5 1 3 3 22 1 4 1 19 23 2 1 24 2 5	4 4 21 1.60 .068 6 3 .35 7! 2 4 58 1.09 .021 2 35 1.25 11	2 .01 2 .35 .03 .09 1 43 15 .01 3 .79 .01 .19 1 81	3 2 2 2 7 2 2 2 3 2 2 2
THR 37 STD C/FA-5X	201 77 1099 557 17 58 42 131	6.3 6 4 168 2.86 573 6.5 67 28 1066 4.13 40	5 ND 18 8 3	1 27 4 8 37 47 16 16		4 .06 5 .50 .01 .05 1 816 8 .07 36 1.98 .06 .14 11 103	

- ASSAY REQUIRED FOR CORRECT RESULT for Cu, Pb > 10,000 pm

			\bigcirc									ASA	MERA	A MI	NER/	ALS	INC.	C	Fili	S #	88-2	2275												Jg	e 4	
SANPLE		No PN	Cu PPN	Pb PPN	Zn PPN	•	NI PPN	CO PPN	Na PPN	Fe	As PPN	U PPN	Au PPN	Th PPM	Sr PPN	Cd PPN	Sb PPN	Bİ PPN	V PPM	Ca 3	P	La PPN	Cr PPN	Ng t	Ba PPN	Ti z	B PPM	Al S	Ha	K S	V PPK	Au** PPB	Pt** PPB	Pd** PPB		
THR 39 THR 39 THR 40 THR 41 THR 42		5 1 1 1	722 33 98 9 201	18 3 5 47	60 33 54 51 71	.1 .1 .1	39 102 18 25 14	20 13 14 6 9	662 636 775 4475 579	11.89 3.04 6.68 2.30 2.38	169 2 9 2 92	5 5 5 5 5	ND ND ND ND ND	1 1 1 1 1	2 126 31 15 6	1 1 1 1 1	2 2 2 2 2 2	2 2 3 2	145	2.67 4.04 3.47		2 4 2 2 2	229	3.72 2.41 2.58 .38 .92	19 57 4 135 16	.04 .11 .22 .01 .03	4 2 2	4.07 2.01 2.43 .20 1.01	.01 .07 .02 .01 .01	.03 .07 .03 .03 .02	1 1 1 1	72 1 1 1 127	2 1 5 1 1	2 2 5 2 2	2 2 2 2 2 2	
THR 43 THR 44 THR 45 THR 46 THR 47		1	114 160 185 148 160	7 7 2 10 17	47 25 36 43 70	.1 .2 .1 .1	12 28 21 43 17	15 23 23 24 21	646 356 453 578 613	4.08 3.45 3.97 5.47 4.23	3 8 5 4 7	5 5 5 5	ND ND ND ND ND	1 1 1 1 1	126 129 88 119 107	1 1 1 1	2 2 2 2 2	2 2 2 2 2 2	67 93 141	3.45 2.81 2.38 2.41 3.34	.033 .068	2 2 2 2 2	26 25 31	1.65 .89 1.15 2.32 1.45	21 19 24 16 17	.29 .19 .23 .35 .34	11 2 2	3.96 3.31 2.75 4.38 3.58	.17 .23 .20 .20 .15	.05 .11 .11 .07 .08	1 1 1 1	3 2 2 2 2	1 4 5 5	1 5 6 7 6	2 2 2 2 2	
THR 48 THR 49 THR 50 THR 51 THR 52			61 65 30 100 133	6 11 5 6 7	18 55 109 61 65	.1 .1 .2 .1 .1	10 25 23 17 14	9 16 17 17 20	581 618 940 757 888	2.39 3.90 5.01 4.38 5.49	5 9 9 2 2	5 5 5 5 5	ND ND ND ND ND	9 1 1 1 1	24 26 36 209 436	1 1 1 1	2 2 2 2 2	2 2 2 2 2	124 147 103		.039 .047	7 2 2 2 3	38 59 21	.96 1.73 2.42 1.82 2.11	8 9 5 58 201	.17 .27 .15 .30 .35	47 10 7	2.34 3.49 2.63 3.97 3.76	.04 .03 .02 .13 .10	.04 .03 .02 .07 .10	1 1 1 1	3 2 27 3 6	3 5 5 5 5 5	2 5 4 6	2 2 2 2 2 2	
THR 53 THR 54 THR 55 THR 56 THR 57		1 1 1 1	274 214 45 86 46	3 9 2 7 8	18 43 56 46 13	.2 .1 .1 .1 .1	25 13 21 11 2	21 22 17 14 4	242 552 1013 661 126	2.44 4.50 4.90 3.67 .84	14 3 2 2 3	5 5 5 6	ND ND ND ND ND	1 1 4 6	156 83 65 97 79	1 1 1 1	2 2 2 2 2	2 2 2 2 2 2	86 127 107	2.75 2.19 10.13 4.78 4.15	.069 .039 .040	3 2 2 6 4	50	.51 1.15 2.21 1.47 .24	17 32 6 30 10	.25 .18 .24 .25 .09	3 9 10	3.34 2.87 2.39 3.30 2.65	.30 .17 .02 .05 .02	.04 .11 .03 .09 .06	1 1 1 1	7 3 19 6 4	9 2 5 4 1	8 6 5 2	2 2 2 2 2	
THR 58 THR 59 THR 60 THR 61 THR 62		1	296 283 368	137 147 6 133 5452	43665 282 385 84 6336	5.0 .4 10.7	12 5 20 7 1	4 18 6	1059 192 796 273 4085	15.11 5.33 4.61 9.37 2.06	177 328 4 468 32	5 5 5 5 5	ND ND ND ND	1 1 1 1	44 6 67 7 4	696 3 4 1 38	2 2 3 2	2 2 2 6 2	215	.17 4.01 .14	.018 .009 .074 .018 .005	2 2 3 2 3	22 92 54 188 5	.97 .16 2.19 .20 .34	8 1 36 1 22	.01 .02 .12 .05 .01	13	1.29 .38 2.48 .56 .67	.01 .01 .08 .01 .01	.03 .01 .08 .01 .02	3 1 1 1	195 199 7 438 2525	3 4 3 4 1	2 11 11 12 2	2 2 2 2 2	
THR 63 THR 64 THR 65 THR 66 THR 67		19 1	322 18 41 111 1414	22 51 13 9 54	45 106 33 56 129	.1 2.3 .3 .2 20.2	21 1 11 11 613	23 3 14 15 390	323 364 373 370 249	4.84 1.88 2.32 3.46 47.03	11 2 3 7 582	5 5 5 5 5	ND ND ND ND ND	1 1 1 7	32 106 93 40 35	1 1 1 5	2 2 2 2 2	2 2 2 2 2 21	20 75	3.83 1.79	.037	3 4 2 2 2	7 21	1.04 .48 .96 1.31 .58	10 31 3 12 3	.25 .08 .22 .36 .01	2 15	1.64 1.32 2.99 2.14 .76	.07 .02 .01 .06 .01	.08 .04 .01 .07 .02	1 1 1 1 1	11 12 1 964	4 1 3 4 5	7 2 4 4 81	2 2 2 2 2	
STD C/FA-5	1	18	57	40	128	6.5	65	28	1059	4.08	40	15	8	37	48	17	16	19	58	.47	.090	40	57	.93	178	.07	38	1.98	.07	.13	10	97	92	101	21	

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ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: SEP 1 1988 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Sep 1. 6./.

ASSAY CERTIFICATE

- SAMPLE TYPE: Pulp

ASSAYER:

C. D.TOYE OR C.LEONG, CERTIFIED B.C. ASSAYERS

ASAMERA MINERALS INC. FILE # 88-2275R

SAMPLE#	Cu	Pb	Zn	Ag
	%	%	%	OZ/T
THR 03	.02	.02	1.08	.10
THR 07	.10	.36	.38	1.47
THR 08	1.35	.01	.01	.41
THR 13	.04	2.28	.19	.91
THR 14	.05	2.23	.35	1.24
THR 15	1.98	.01	.02	.46
THR 16	.70	.03	.10	1.41
THR 19	.43	.10	.08	7.19
THR 20	1.18	.04	.47	1.46
THR 34	1.01	.01	.01	.69
THR 58	.05	.02	5.87	.24
THR 67	2.24		.01	1.02

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GEOCHEMICAL ANALYSIS CERTIFICATE

852 E. HASTINGS ST. VANCO

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3NL 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 NL WITH WATER. THIS LEACH IS PARTIAL FOR NN FE SR CA P LA CR NG BA TI B W AND LINITED FOR NA K AND AL. AU DETSCTION LIMIT BY ICP IS 3 PFN. - SAMPLE TYPE: ROCK AU** ANALYSIS BY FA+AA FROM 10 GM SAMPLE.

						avui a	5 1112		Л	ANA 1131	1		1		55.		\bigcap	1														
DATE	RECEI	IVED	: AU	G 23 1	988	DATE	RE	POR	T MAIL	ED:	Hw	30	/8e	2	ASSA	YER.	<u>ب.</u>	him		.D.1	OYE	OR	C.L	EONG	;, C	ERTI	FIE	DB.	.c.	ASS	AYERS	;
							ASA	MERA	MINER	AL I	NC.	PROJ	JECT	THO	RNE	LK.	1	File	; ≓.	88-3	815		Page	1								
SAMP	15#	MC PPM	Cu 2PM	PD PPM	Zn PPM	Ağ PPN	NÍ ?PM	CO PPN	NA PPM	Fe A 3 PP		U AU H PPN		Sr PPH	Cđ PPN	SD PPM	BI PPM	7 PPN	Ca ł		La PPN	Cr PPN	Nç Ş	Ba PPN	Ti z	B ??N	Al B	Na .3	л १		Au** PPB	
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88 TH 88 TH 89 Th 88 Th 88 Th 88 Th	R 255 R 257 R 258	4 12 7 6 5	42 51 93 30 20383	87 235 145 224 116	29 50 35	6.3 35.9 6.4 15.8 19.7	3 3 2 3 110	5 7 1 1 55	2931 8. 113 3. 1578 12. 156 3. 771 9.	74 72 32 93 91 11	3 (1]	ND ND 2	2 2 3 1 2	9 4 39 7 36	2 1 1 5	2 5 5 5 2	12 98 9 13 52	68 7 67 18 116	.07 .01 .08 .14 1.83	.004 .084 .007	2 2 3 2 2	3 7 4	1.30 .04 .39 .05 2.14	23 5 32 15 10	.01 .01 .27 .01 .10	3 2 5	2.55 .14 1.40 .43 1.93	.01 .01 .01 .01 .02	.14 .03 .55 .16 .08	4 7 4	234 1210 154 2732 1415	
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	274A 274B 275	2 8 1 1 101	425 1347 1028 102 104		396 16635 47 200 156	3.6 3.8 .3	15 12 14 3 23	31 36 15 10 134	6047 7. 511 6. 78 31. 204 6. 32 13.	6 413 3 56 3 11	8 5 5	ND ND ND	1 2 5 1 3	95 34 2 4 1	5 213 1 2 7	2 2 2 2 7	2 2 19 2 53	42 1 139 359 108 6		.036 .003 .022 .068 .001	5 6 2 2 2	40 70	1.14 1.25 .05 1.08 .01	10 5 12 15 7	.02 .22 .05 .21 .01	19 2 2	1.44 1.16 .21 1.09 .03	.01 .01 .01 .02 .01	.09 .08 .01 .15 .02	4 1 2 1 2	534 99 892 29 396	
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- ASSAY REQUIRED FOR CORRECT RESUL for G. PS Zn >10,000 pph

Ag>35pm

	C)					AS	AMERA	MI	NER	AL 1	NC.	PRC	JEC	т тң	0	E LÞ	κ.	FILI	E #_	88-3	8815								(3
SAMPLE‡	Ho PPM	CU PPN	PD P?N	Zn PPM	Ag PPM	Nİ PPM	Co PPN	Hn PPN	īe }	As PPN	U PPH	Au PPN	Th PPN	ST PPM	Cđ PPN	Sb PPN	Bİ PPM	V P PN	Ca Z	P Z	La PPN	- Cr PPN	Ng t	Ba PPN	Ti ł	B PPN	Al t	Na ł	r z	¥ Au** PPN PPE	
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GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR MA R AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM. - SAMPLE TYPE: ROCK AU** PT** PD** RH** BY FA-MS.

Cu Pb Zn λq Co Ma Fe As U Au Th Sb W Au** Pt** Pd** Rh** Sr Cd Bi V Ca P La Cr Mg Ba ŤÍ Al Na В Ā PPN PPN PPM PPM PPN PPN PPN PPN 3 PPM PPN PPH PPN PPN PPM PPN PPM PPN PPN PPN ş \$ \$ PPH ł PPN \$ PPN 2PB PPB PPB PPB \$ 2 88 TR 266 41 2003 120 164 3.5 26 76 529 19.34 42 ND 5 4 - 5 2 6 2 106 .12 .024 49 1.78 2 3.14 2 4.02 .01 .01 121 1 19 2 89 TR 267 42 9738 38 106 7.3 101 41 469 3.98 45 5 ND 2 137 2.48 1.200 1 54 2 2 20 2.37 19 1 .05 20 2.18 .01 .02 1 539 7 27 2 88 TR 271 114 1278 154 375 3.0 72 135 250 10.57 39 ND 72 5 1 2 2 3 105 .65 .094 51 1.28 21 .21 2 2.23 .03 .04 1 72 4 7 18 2 83 TR 277 3 13986 1529 852 125.0 51 30 866 26.12 1589 5 ND 5 3 - 9 6 390 31 .07 .021 2 14 1.27 5.02 5 3.21 .01 .03 1 205 2 25 2

ASSAY CERTIFICATE

- SAMPLE TYPE: Pulp

ASAMERA MINERAL INC. PROJECT THORNE LK. FILE # 88-3815R

SAMPLE#	Cu	Pb	Zn	Ag
	چ	%	%	OZ/T
88 TR 253	.02	.03	.01	1.08
88 TR 256	.01	.02	.01	1.09
88 TR 259	1.87	.01	.01	.67
88 TR 261	.08	.06	.11	1.20
88 TR 262	.04	.05	.05	1.41
83 TR 272	25.13	.01	.10	4.90
88 TR 274A	.13	.01	1.74	.17
88 TR 274B	.11	.01	.01	.19
88 PS 512	.01	.03	3.82	.73
88 PS 514	1.47	.01	.08	.39
88 PS 524 88 PS 525 88 PS 526 88 PS 532 88 PS 533		.01 14.21 .04 .12 .03	.09 1.77 .04	.13 18.60 11.74 1.81 .47
88 PS 534 88 PS 535 88 PS 536 88 PS 537 88 PS 538	1.72 2.21 .46 3.55 1.02	.08 .02 6.04 .06 .89	.10 .20 .02	2.14 3.75 10.21 3.64 2.34
88 PS 539	1.77	.02	.08	8.61
88 PS 540	1.58	2.07	7.97	
88 PS 541	.20	.28	.10	
88 TR 277	1.42	.20	.11	

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GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3NL 3-1-2 HCL-HR03-H20 AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 NL WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SE CA P LA CE MG BA TI B W AND LIMITED FOR MA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPN. - SAMPLE TYPE: ROCK AU* AMALISIS BY ACID LEACE/AA FROM 10 GM SAMPLE.

Sept 17/83 DATE RECEIVED: SEP 13 1988 DATE REPORT ASSAYER. ASAMERA MINERALS INC. File # 88-4448 SAMPLE Pb Zn λσ Ni Co Xn Σe Sr Cđ Sb Xo Cu λs Th Bi Ca 2 La Cr χđ Ba Ti 11 ¥ Au* U λu ٧ В Xa PPH PPN PPN PPN PPN PPN PPN PPN Y PPH PPH PPH PPH PPH PPH PPN PPN PPN ł \$ PPX PPM Ł PPN 2 PPN Ł ł PPN PPB Ł 039559 787 6 120 2151 4.1 25 11 5043 11.05 105 -5 XD 2 12 12 2 2 138 .31 .081 4 99 2.99 46 .01 2 4.40 .01 .14 1 - 44 039560 - 5 879 72 1256 26 12 5089 10.73 293 5 11 119 .19 .059 123 2.74 6.2 HD 2 4 3 2 3 65 .01 2 4.21 .01 .13 3 89 039561 15 74454 38 324 72.0 302 36 1074 9.86 1 5 17 11 48 1.65 .085 25 1.32 6 1 2 84 3 8 . 01 2 1.85 .02 .09 1 9320 039562 10 110 162 38 6.1 10. 33 162 5.59 1862 5 XD 3 2 11 15 .05 .010 2 6 .07 9 .02 1 1 2.25 .01 .07 4 2700 316 5 13 039563 1 247 456 2.9 11 15 4679 14.18 488 ND 2 1 2 2 187 .56 .112 2 17 1.95 13 .32 2 3.97 .01 .19 1 360 039564 11 31 70 156 2.2 9 42 156 6.98 475 5 XD 3 1 3 2 22 .04 .011 2 8 .07 5 . 02 2 .18 .01 .03 4 870 039565 305 36 1196 37 4567 8.23 17 5 XD 27 12 1 .2 23 2 2 175 .90 .083 15 2.98 21 .34 1 3 2 4.38 .01 .12 1 23 039566 1 314 1290 1809 . 72 21 3652 16.44 157 5 3 9 9 5 202 .27 .062 3.8 2 2 2 363 3.20 8 .18 2 4.17 .01 .03 4 1270 039567 1 710 11525 16747 8.4 1 12 3060 3.61 20 5 ND 1 84 93 2 5 70 9.91 .033 3 14 1.21 13 .09 2 2.55 .01 .11 1 65 039573 61 48 37 341 1.3 33 15.51 .010 6 10 3709 2.63 3 6 ND 3 146 , , 4 11 1.17 81 .01 2 .79 .01 .01 1 42 039574 465 7 16 83 .1 5 9 655 13.76 - 32 5 ND 2 9 132 .17 .062 2 127 4.21 8 . 04 3 5,95 .01 . 02 3 10 1 2 2 039575 9 205 803 1109 3.8 5 9 85 4.09 202 5 ND 15 2 27 . 32 65 1 2 71 144 .14 .043 .08 2 .54 .01 .11 2 2 60 039576 34 541 3080 17114 47.3 27 131 36 16.00 535 9 ND 2 3 216 2 61 10 .03 .008 2 11 .05 1 .02 2.13 1 380 .01 .07 039577 57 131 417 516 7.1 17 9 14 233 1.29 38 5 ND 10 2 20 .7 .15 1 8 1.99 .008 2 7 .01 2 .23 .01 .04 1 77 4 1561 23051 10141 158.2 039578 11 4 336 1.47 37 5 ND 1 25 189 2 310 184 2.30 .021 2 12 .23 9 . 02 10 .36 .01 .06 1 109

STD C/AU-R 18 58 41 132 6.7 64 29 1033 3.98 40 16 7 36 47 18 21 23 57 .46 .090 38 53 .86 172 .06 32 1.85 .06 .14 11 530

- ASSAY REQUIRED FOR CONNECT RESULT for Cu Pb 30 >10,000 ppm

ACME ANALYTICAL LABORATORIES LTD.

852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716

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GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MM FE SR CA P LA CR MG BA TI B W AND LIMITED FOR WA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM. - SAMPLE TYPE: ROCK AU** AMALYSIS BY FA+AA FROM 10 GM SAMPLE.

S PPN 2 \$ % PPM PPB PS 587 4 109 60 202 24 124 1195 14.79 3551 4.0 5 2 1 19 2 2 8 79 .32 .040 10 . 82 14 .08 2 4 1.76 .01 .08 2 2120 PS 588 2 108 54 156 3.5 28 133 998 16.51 2207 5 ND 2 23 -1 2 4 72 .61 .035 2 5 .70 5 .11 2 1.64 .01 .08 1 1310 PS 589 1 4 5 19 .1 1 23 18092 1.97 65 5 ND 95 1 1 2 2 7 27.80 .002 3 1 .13 2 .01 2 .18 .01 .01 1 42 PS 590 13 1694 54 2802 6.5 13 24 1738 17.27 324 5 ND 2 8 19 2 16 61 .42 .018 2 25 1.06 5 .05 2 2.04 .01 .05 1 4610 STD C 17 59 44 133 7.2 67 31 1047 4.17 44 17 7 37 48 18 15 19 58 .50 .087 39 52 .94 183 .07 33 1.96 .06 .13 11 -

APPENDIX II

Rock Sample Descriptions

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ROCK SAMPLE DESCRIPTIONS

	ROCK SAMPLE DESCRIPTIONS
SAMPLE NO.	DESCRIPTION
THR 01	Rock grab: rusty quartz and calcite veinlet with pyrite, non magnetic
THR 02	Float: rusty quartz with pyrite, non magnetic
THR 03	Float: rusty quartz with pyrite and hydrozincite, non magnetic
THR 04	Float: hornblende-epidote porphyry with malachite, moderately magnetic
THR 05	Float: fragmented greenstone with jasper? and epidote, non magnetic
THR 06	Float: carbonate veining with iron stain, hematite, pyrite, and malachite, non magnetic
THR 07	Float: rusty with malachite stain, non magnetic
THR 08	Float: rusty with malachite stain, strongly magnetic
THR 09	Float: rusty with quartz, non magnetic
THR 10	Rock grab: calcite stringers, non magnetic
THR 11	Rock grab: rusty with malachite stain
THR 12	Rock grab: from a rusty zone 0.5 metre wide, non magnetic
THR 13	Float: rusty quartz, non magnetic
THR 14	Rock grab: rusty banded quartz and carbonate, non magnetic
THR 15	Rock grab: rusty iron staining with malachite, moderate to strongly magnetic
THR 16	Rock grab: spongy and rusty rock with malachite, non magnetic
THR 17	Rock grab: rusty quartz, non magnetic
THR 18	Rock grab: rusty quartz with malachite and chalcopyrite, non magnetic

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SAME	PLE NO.	DESCRIPTION
THR	19	Rock grab: spongy and rusty rock with pyrite and trace malachite, non magnetic
THR	20	Rock grab: spongy and rusty rock with massive pyrite, non magnetic
THR	21	Rock grab: altered porphyry adjacent to spongy orange rock, non magnetic
THR	22	Rock grab: rusty, silicic orange rock containing pyrite, in a one metre wide zone, weakly magnetic
THR	23	Rock grab: in a volcanic rock with quartz and malachite, weak to moderately magnetic
THR	24	NOT USED
THR	25	Rock grab: rusty with pyrite, possible arsenopyrite, non magnetic
THR	26	Rock grab: rusty banded metavolcanic rock with calcite and quartz veining and pyrite, non magnetic
THR	27	Float: altered intrusive, non magnetic
THR	28	Float: rusty altered porphyry with pyrite, non magnetic
THR	29	Float: rusty with quartz and pyrite, non magnetic
THR	30	Float: rusty altered porphyry, non magnetic
THR	31	Rock grab: rusty banded metavolcanic rock with calcite and quartz veining and pyrite, moderate to strongly magnetic
THR	32	Rock grab: iron-welded breccia, weakly magnetic
THR	33	Rock grab: rusty with pyrite, possible arsenopyrite, non magnetic
THR	34	Float: quartz stringer in intrusive rock with malachite and chalcopyrite
THR	35	Float: quartz-carbonate veining in mafic host, with hydrozincite and possible copper, non magnetic

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SAMPLE NO.	DESCRIPTION
THR 36	Float: rusty with quartz and pyrite, non magnetic
THR 37	Float: rusty with blue metallic mineral
THR 38	Float: massive pyrite, non magnetic
THR 39	Float: fine to medium grained mafic intrusive, weakly magnetic
THR 40	Float: quartz and calcite stringers in basic volcanics, weakly magnetic
THR 41	Float: quartz-carbonate with pyrite, non magnetic
THR 42	no description
THR 43	no description
THR 44	Rock grab: rusty tuff with pyrite, non magnetic
THR 45	Rock grab: rusty tuff with pyrite, non magnetic
THR 46	Rock grab: rusty tuff with pyrite
THR 47	Rock grab: rusty tuff with quartz and calcite veining and pyrite, non magnetic
THR 48	Rock grab: fractured volcanic rock with feldspar veining, non magnetic
THR 49	Rock grab: fractured volcanic rock with tiny quartz veins, non magnetic
THR 50	Rock grab: fractured volcanic rock with tiny quartz veins, non magnetic
THR 51	Rock grab: fractured volcanic rock with tiny quartz veins, non magnetic
THR 52	Rock grab: fractured volcanic rock with calcite veining
THR 53	Rock grab: basic volcanic rock with quartz stringers and pyrite, non magnetic
THR 54	Float: porphyry with quartz stringers and pyrite, weakly magnetic

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SAMPLE NO.

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DESCRIPTION

THR	55	Rock grab: fractured volcanics with quartz veining, non magnetic
THR	56	Rock grab: from a quartz-feldspar vein 20 cm. wide, non magnetic
THR	57	Rock grab: from a quartz-feldspar vein 20 cm. wide, non magnetic
THR	58	Rock grab: from quartz-carbonate vein with massive pyrite, zinc, and trace copper, non magnetic
THR	59	Rock grab: orange-weathered mafic porphyry with quartz-carbonate veining, non magnetic
THR	60	Rock grab: orange-weathered mafic porphyry with quartz-carbonate veining, non magnetic
THR	61	Rock grab: quartz-carbonate with malachite, non magnetic
THR	62	Float: quartz with copper and zinc, non magnetic
THR	63	Rock grab: volcanic breccia with occasional massive pyrite, non magnetic
THR	64	Rock grab: quartz vein stockwork in altered intrusive, containing pyrite; non magnetic
THR	65	Rock grab: volcanic rock with epidote veining, non magnetic
THR	66	Float: rusty porphyry with pyrite
THR	67	Float: with magnetite, malachite, strongly magnetic
TR 2	250	Quartz-epidote shear zone filling 30 cm. in width with specks of chalcopyrite
TR	251	Gossanous shear zone to 15 cm. in width, 20% deeply weathered pyrite
TR	252	Chip sample across 60 cm. in gossan zone with propyllitic rock laced with limonitic stringers

SAMPLE NO.	DESCRIPTION
TR 253	Collection of limonitic fragments derived from 5 to 15 cm. wide limonitic zone in shear
TR 254	Fractured rock with thin quartz stringers in 10 to 15 cm. wide zone of deeply weathered pyrite. Continuation of THR 17 and THR 18
TR 255	45 cm. chip sample of hanging wall. Propyllitic, limonitic rock, formerly greenstone
TR 256	10 cm. chip sample with quartz and deeply weathered pyrite
TR 257	20 cm. chip sample across propyllitic and pyritized footwall
TR 258	Thin (5 cm.), deeply weathered gossan with minor relict pyrite in highly shattered volcanics. Wall rock selvage weakly bleached, local strong Mn staining
TR 259	Chalcopyrite pod in augite porphyry
TR 260	Chips of gossanous rock floating on loose dirt collected across a 6 metre wide zone sandwiched between exposures of highly fractured greenstone
TR 261	2.5 metre wide chip sample across a clay gouge zone, unearthed by hammer-trenching. Rock contains yellowish iron sulphate and fine granular quartz and clay
TR 262	2.5 metre wide chip sample, similar to, and continuous with TR 261
TR 263	Gossanous float chips taken below TR 261 and 262
TR 264	One metre wide chip sample across argillically altered pyritic zone
TR 265	Sample from 45 cm. wide quartz-breccia zone in massive augite porphyry
TR 266	One metre wide section of very rusty weathering pyritic vein, with a core of coarse-grained pyrite and chlorite

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SAMPLE NO.	DESCRIPTION
TR 267	Sample from 5 to 8 cm. wide shear zone with prominent lenses of Fe-rich hornblende, pyrite and chalcopyrite
TR 268	Bleached chips off ridge, adjacent to THR 25, 26, and 35
TR 269	Sample from yellowish-rusty clay alteration zone on surface
TR 270	Sample from prominent float train of gossanous fragments, completely weathered out
TR 271	Sample from 20 cm. quartz-chlorite-hornblende- pyrite vein
TR 272	Grab: from coarse-grained float with calcite, quartz, and chalcopyrite
TR 273	Grab: from 9 metre wide zone with quartz, calcite, chlorite, Mn ankerite, pyrite, and minor chalcopyrite
TR 274	Grab: from float train of pieces 15 to 30 cm. across with garnet and pyrite
TR 275	Grab: from float train of limonitic and pyritic rocks
TR 276	Grab: rock with rusty stringers of limonite, quartz, and K-feldspar
TR 277	Grab: from base of shear vein with pyrite and amphibole
TR 278	Grab: from block of quartz-pyrite
PS 508	Float: rock with pyrite, carbonate, chalcopyrite
PS 509	Grab: weathered pyrite vein
PS 510	Chip across 75 cm. weathered quartz vein with limonite and pyrite
PS 511	Chip across one metre with pyrite, quartz, and limonite

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SAMPLE NO.	DESCRIPTION
PS 512	Grab: from east side of vein, one metre below PS 509, with chlorite, pyrite, coarse sphalerite, hematite, and manganese
PS 513	Chip across 50 cm. poorly exposed vein 15 metres west of PS 512 and parallel to that vein
PS 514	Float: rounded rock containing quartz, pyrite, and chalcopyrite
PS 515	Float: from talus, with quartz stockwork, carbonate, and pyrite
PS 516	Float: angular quartz-pyrite boulder with carbonate stringers
PS 517	Siliceous rock with pyrite from a small gossan
PS 518	Quartz-pyrite in fault breccia
PS 519	Rusty rock with quartz
PS 520	Grab: from talus, with rusty patches
PS 521	Sample from 10 cm. rusty zone in gully
PS 522	10 cm. sample with quartz and limonite
PS 523	Float: ankeritic rock fragments with minor galena and pyrite
PS 524	Float: in talus, containing chalcopyrite
PS 525	Float: rusty rock from talus containing galena, cerrusite, and malachite
PS 526	Float: from talus, containing carbonate, chalcopyrite, and malachite
PS 527	Float: from talus 100 metres east of PS 523, containing quartz, pyrite, and chalcopyrite
PS 528	Float: from talus, with quartz stockwork
PS 529	Float: from talus below rusty seam, containing quartz and pyrite

SAMPLE NO	. DESCRIPTION
PS 530	Sample taken near pink granite contact, with quartz stockwork, similar to PS 528
PS 531	Float: granitic rock in talus with quartz stringers and pyrite
PS 532	Float: from talus below large gossan, containing quartz and pyrite
PS 533	Float: from talus below large gossan, and 25 cm. below a vein; containing quartz and pyrite
PS 534	Float: from talus below large gossan, and 25 cm. below a vein; containing quartz, carbonate, chalcocite, chalcopyrite, and pyrite
PS 535	50 cm. chips near footwall of shear, with carbonate, pyrite, chalcopyrite, malachite, and azurite
PS 536	Fractured rock at top of talus slope, with quartz, galena, malachite, chalcopyrite, and pyrite
PS 537	Float: from talus 100 metres north of gossan, within an area of mineralized float with no obvious source above. Contains quartz, pyrite, and chalcopyrite
PS 538	Float: talus sample of silicified rock with quartz, galena, pyrite, and malachite; 15 metres north of PS 537
PS 539	40 cm. chip sample in feldspar porphyry with pyrite, chalcopyrite, malachite, and azurite in a vein
PS 540	Float: talus sample below intrusive with quartz, galena, chalcopyrite, pyrite, and sphalerite
PS 541	Float: quartz, pyrite, and malachite in a weathered rock
PS 587	Rock chip sample across 45 cm. in a trench dug near sample PS 518. Sample contains 30 cm. quartz and 15 cm. pyrite in a rusty clay gouge with malachite
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SAMPLE NO.	DESCRIPTION
PS 588	Rock chip sample similar to PS 587, but 40 cm. to the north from PS 587
PS 589	Float: 10 metres up creek from PS 587 in an ankeritic rock with 5% pyrite
PS 590	Rock chip sample across 60 cm. in a trench dug 47.5 metres north from location of samples PS 587 and PS 588. Sample contains 30 cm. of quartz and 30 cm. of clay gouge, pyrite, ferricrete, and malachite
039559	Float: limonitic rock near TR 261
039560	Float: limonitic rock near 039559
039561	Float: limonitic talus sample near TR 522
039562	Sample taken near PS 518, in a fault zone with ferricrete, from an oxidized rock containing pyrite
039563	Sample taken near PS 518, in a fault zone with ferricrete, from an oxidized rock containing pyrite
039564	Sample taken below a slump in a highly oxidized zone, near sample 039563
039565	Sample of gouge breccia in a shear zone one metre wide
039566	Resampling of PS 519
039567	Selected samples of quartz vein in footwall of fault
039573	Highly altered rock with carbonate, epidote, and trace pyrite, possibly a skarn
039574	Quartz vein in gossan
039575	Sample contains jasper with euhedral pyrite
039576	Limonitic rock with quartz and pyrite

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SAMPLE NO.	DESCRIPTION
039577	Sample from cherty vein with jasper and sulphides
039578	Limonitic rock with quartz, carbonate, and pyrite

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APPENDIX III

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Statement of Qualifications

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STATEMENT OF QUALIFICATIONS

- I have worked during the past twelve years for several mining exploration firms, including RioCanex Inc., Minequest Exploration Associates Ltd., Asamera Inc., and Asamera Minerals Inc.
- 2. I have supervised, designed, and conducted several geochemical and prospecting programs, including the work done on the Thorne claims contained in this assessment report.
- 3. I wrote this assessment report, using information supplied by the various sub-contractors hired by Asamera to work on the claims, and personal work experience on the property in 1988.
- 4. I am a member of the Geological Association of Canada.

Paul D. в McCarthy,

A P P E N D I X I V List of Personnel

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LIST OF PERSONNEL

NAME	COMPANY	WORK PERIOD			
K.Bilquist	Lone Trail Prospecting Ltd.	June 20 - 24, 1988			
R.Bilquist	Lone Trail Prospecting Ltd.	June 20 - 24, 1988			
G.Dickson	Asamera Minerals Inc.	Sept. 7 - 8, 1988			
G.Graham	Asamera Minerals Inc.	June 20 - 24, 1988			
P.McCarthy	Asamera Minerals Inc.	Aug. 13 - 19, 1988			
T.Richards	Tom Richards Prospecting Ltd.	Aug. 13 - 19, 1988			
P.Suratt	Tom Richards Prospecting Ltd.	Aug. 13 - 19, 1988 Sept. 23 - 26, 1988			

Prospecting: K.Bilquist, R.Bilquist, G.Dickson, G.Graham, P.McCarthy, T.Richards, P.Suratt

Silt Sampling: G.Graham, P.McCarthy

A P P E N D I X V

Statement of Costs



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STATEMENT OF COSTS THORNE CLAIMS June 20,1988 to December 1,1988

Fees and Wages

Disbursements

P.McCarthy	12	days	at	\$136.36	per	day		1636.3	2
G.Graham	5	days	at	\$125.00	per	day		625.0	0
G.Dickson	3	days	at	\$360.00	per	day		1080.00	0
Lone Trail Prospecting Ltd.							1505.00	0	
Tom Richards Prospecting Ltd.					•	16940.90	6		

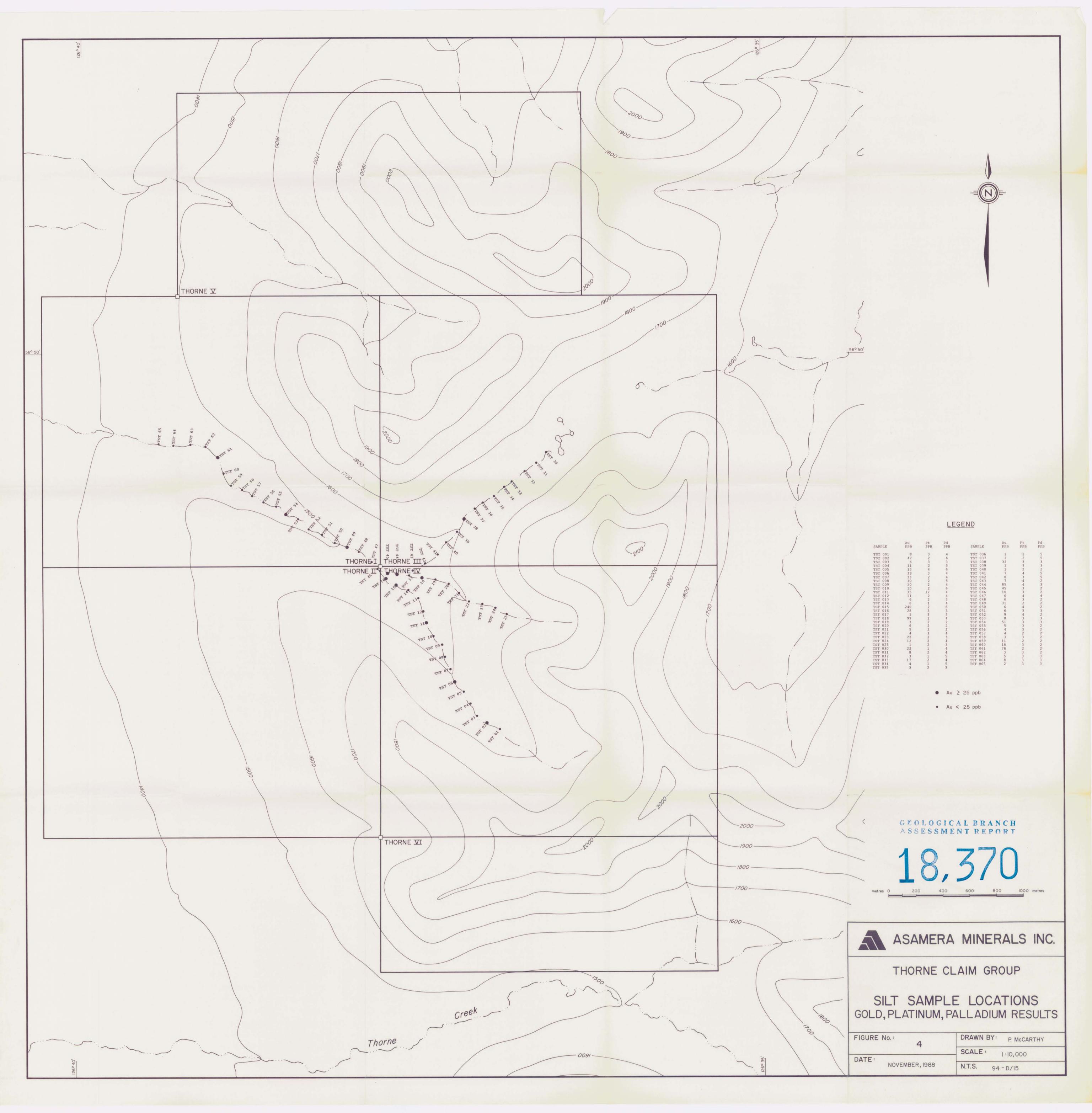
\$ 21787.28

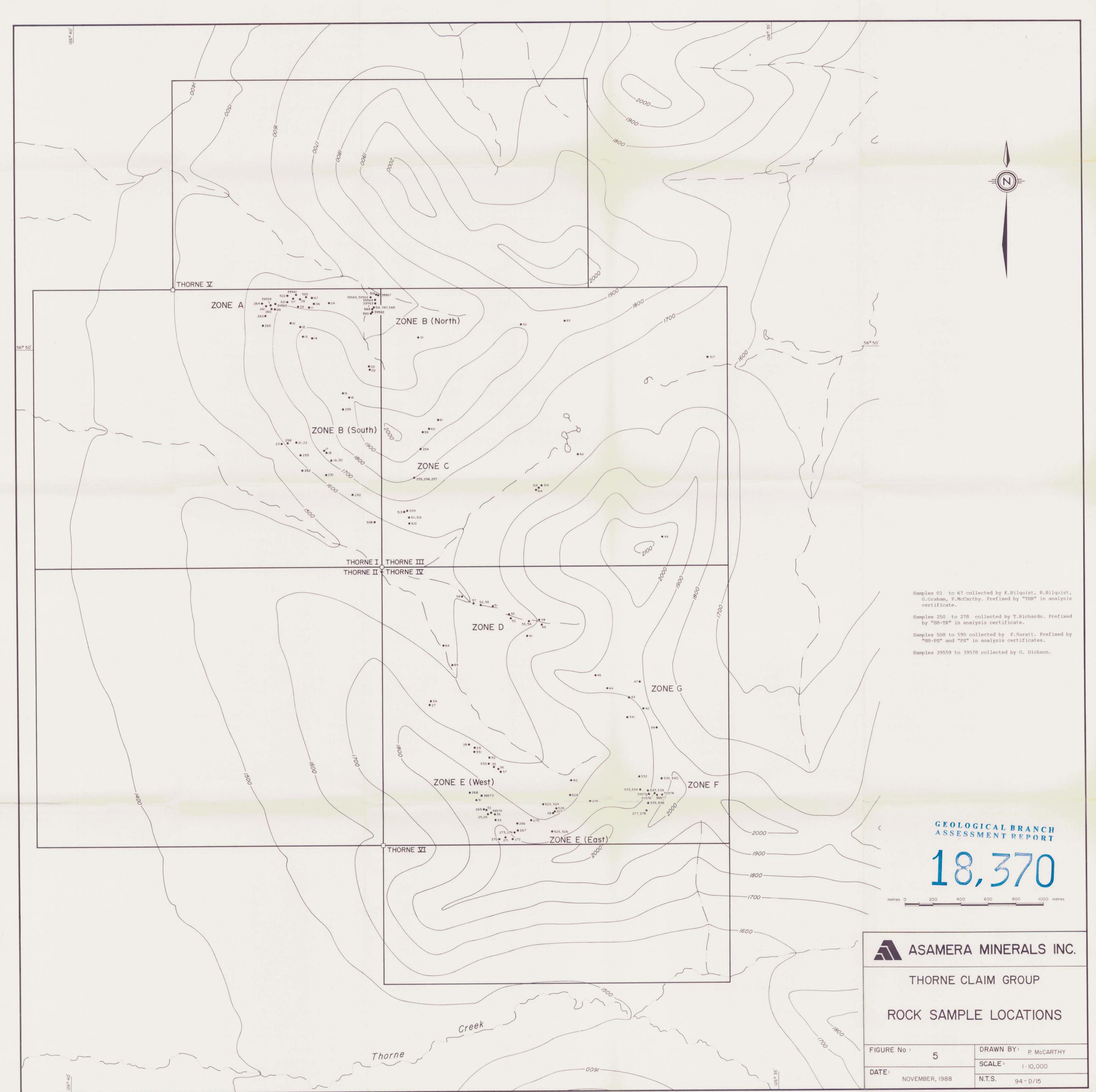
Accomodation	680.04
Camp supplies and consumables	2828.94
Commercial flights	4302.62
Drafting and related costs	966.16
Fuel and lubricants	207.32
General supplies	105.52
Geochemical analysis	5342.06
Helicopter charter	10333.41
Shipping	151.83
Vehicle rental	793.53
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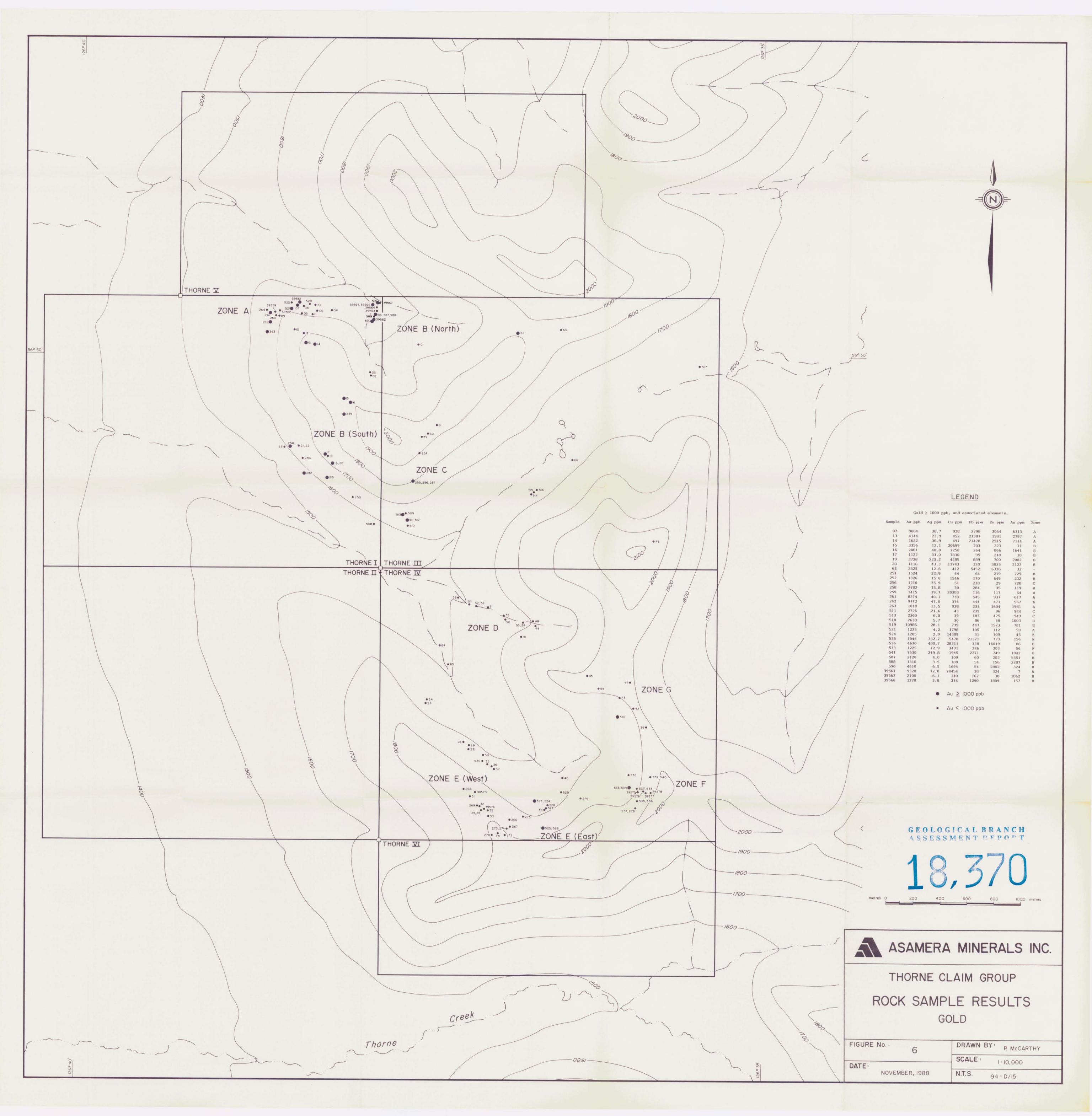
\$ 22359.26

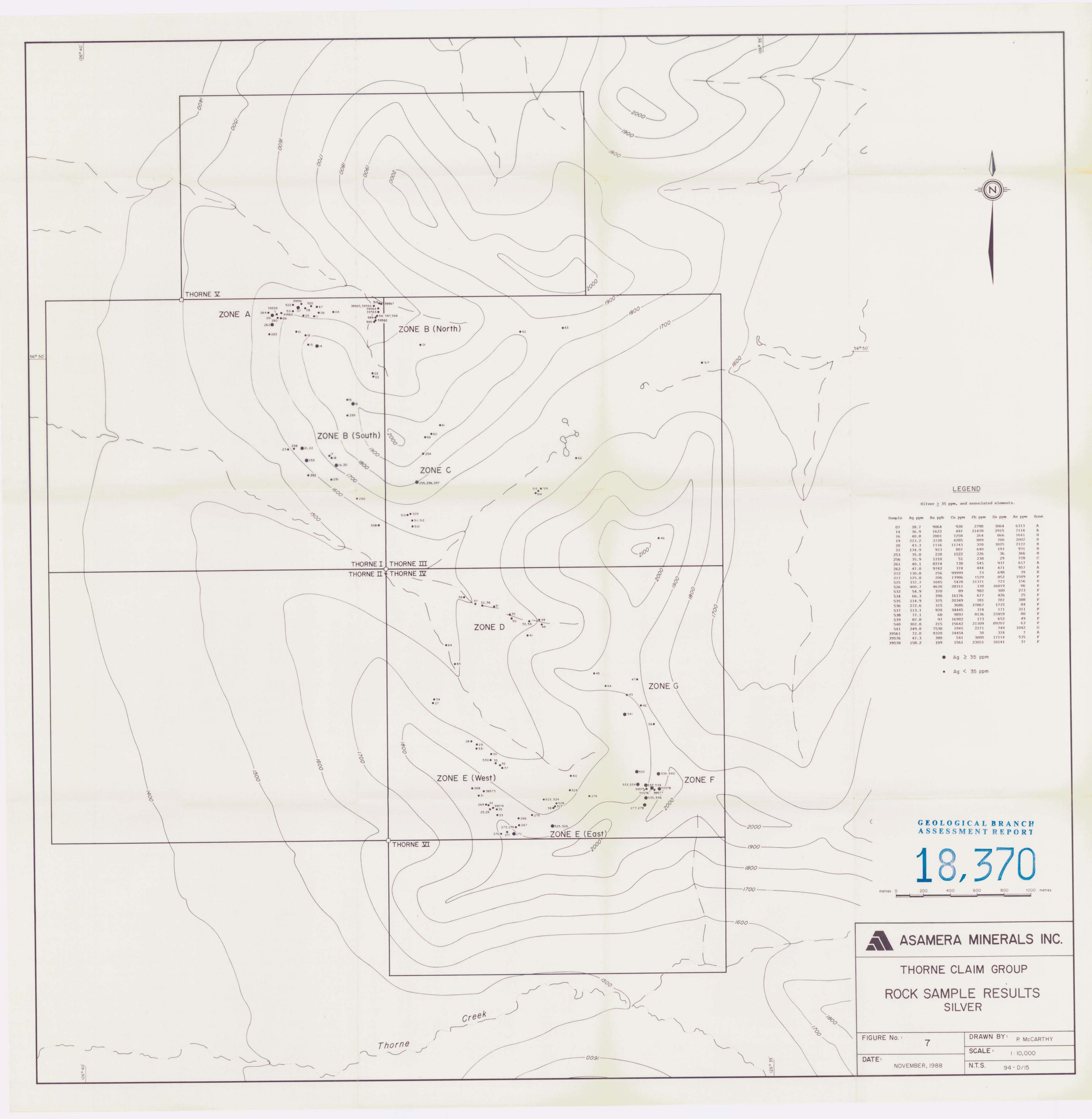
TOTAL COSTS:

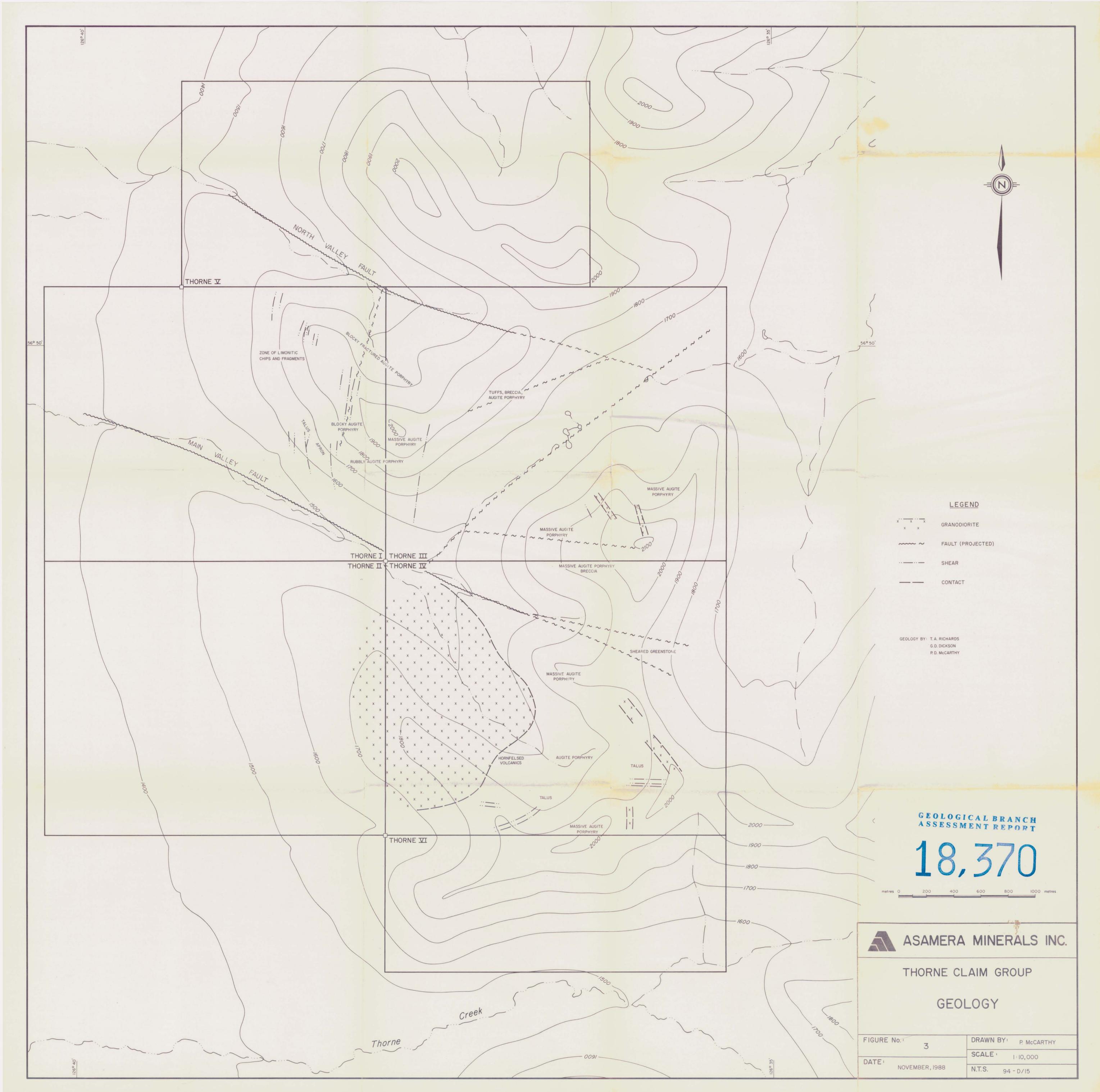
\$ 44146.54











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