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ASSESSMENT REPORT ON THE WREN CLAIM GROUP

RUTHERFORD CREEK AREA

LILLOOET MINING DIVISION, B.C.

NTS 92 J/6E AND 7W

SUB-RECORDER
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 VANCOUVER, B.C.

BY

R. A. GONZALEZ, M.Sc., F.G.A.C.

DECEMBER 1989

Claim Name Record No. Units Anniversary Date

SPARROW	3817	12	SEPT 01, 1987
JAY	3819	1	SEPT 21, 1987
ROBIN	3820	1	SEPT 21, 1987
CROW	3821	1	SEPT 21, 1987
WREN	3835	20	OCT 05, 1987

LOCATION: 50° 16' N, 123° 00' W
 OWNERS: CASTLE MINERALS INC.
 (SPARROW, JAY, ROBIN, AND CROW CLAIMS)
 OPERATOR: JIM McDONALD (WREN CLAIM)
 CONSULTANT: CASTLE MINERALS INC..
 PROJECT GEOLOGIST: ARCHEAN ENGINEERING LTD.
 R. GONZALEZ, P.ENG.

GEOLOGICAL ASSESSMENT REPORT
 1989/12/29

**ASSESSMENT REPORT ON THE WREN CLAIM GROUP
RUTHERFORD CREEK AREA
LILLOOET MINING DIVISION, B.C.
NTS 92 J/6E AND 7W**

SUMMARY

In early 1987, **CASTLE MINERALS INC.** optioned 1 Modified Grid claim (Wren Claim), comprised of 20 units in the Lillooet Mining Division. Three 2-post claims and one additional Modified Grid claim (Sparrow Claim), totalling 12 units, were later added to Castle's land position. The claims are located in the valley of Rutherford Creek, approximately 15 km southeast of the village of Pemberton.

Access to the claims is via a graded-gravel logging road along the north side of Rutherford Creek. A bridge crosses the Creek immediately north of the claims where additional logging roads traverse the claims.

The property was staked to cover a large, gold-bearing shear zone found in the lower slopes immediately south of Rutherford Creek. This area was first staked in the late 1970's and various cursory development programmes, including geochemical sampling, a test I.P. survey, and trenching. The I.P. survey outlined an anomalous area within the shear; however, this anomaly was never drill tested. Trenching over a portion of the I.P. anomaly exposed a silicified, pyrite-bearing shear zone carrying low gold values.

The results of these earlier programmes and an exploration programme carried out by Castle in 1988 outlined a 200 m wide shear zone which contained up to 780 ppb gold in soils and visible flakes of angular gold from panned soils.

Between 15 May and 15 July, 1989, Castle Minerals completed five diamond-drill holes totalling 219.8 m (721 feet). The results of Castle's 1988 soil and rock geochemical survey were used as a guide for target selection. This report summarizes the results of this small diamond drilling programme.

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1.0 INTRODUCTION

In early 1987, a dialogue was established between the claim holder, Mr. Jim McDonald, of Vancouver, B.C., and **CASTLE MINERALS INC.** in which the latter agreed to option 1 Modified Grid Claim, comprised of 20 units. During the initial exploration, it became apparent that the economic potential of the area was considerable and additional claims were required. **CASTLE** purchased or optioned three 2-post claims and one Modified Grid claims comprised of 12 units, contiguous to the original Wren Claim. These claims form a block collectively called the Wren Claim Group.

Between 15 May and 15 July, 1989, five, short diamond-drill holes totalling 219.8 m (721 feet) were completed to test five areas geochemically anomalous in gold. This report summarizes the results of that programme and discusses the various occurrence of gold as indicated from the drilling.

1.1 LOCATION AND ACCESS

The Wren Mineral Group is a gold prospect located on the south side of the Rutherford Creek in southeastern British Columbia. The claims are located in moderately steep, mountainous terrain approximately 120 km north of Vancouver (Figure 1). The town of Pemberton is 10 km north, and the Village of Whistler is approximately 15 km to the south. Terrestrial co-ordinates for the centre of the claim block are as follows:

50° 16' North Latitude
123° 00' West Longitude
NTS 92 J/6E and 7W

The property is at an elevation which ranges from 800 m (2600 feet), along Rutherford Creek, to over 2150 m (7060 feet) at the ridge top which divides the east flowing Rutherford Creek and Soo River drainages.

Access to the property is along a low-maintenance, dry weather, logging road which trends westward along the north side of Rutherford Creek. This road connects with the Vancouver-Pemberton Highway (B.C. Highway 99) approximately 10 km south of the town of Pemberton. The north boundary of the Wren Mineral Group is immediately south of a logging bridge which crosses Rutherford Creek. The lower, northern, portions of the claim group was logged during 1986 and 1987 and several logging roads cross the property.

1.2 CLAIM INFORMATION

The property is located in the Lillooet Mining Division and is comprised of two Modified Grid claims, totalling 32 units, and three 2-post claims (Figure 2). All claims are contiguous. The Wren Claim is held by an option agreement with the recorded holder, Mr. Jim McDonald of Vancouver, while the Sparrow, Robin, Jay, and Crow are recorded in the name of Castle Minerals Inc. For claim information, see Table 1.

TABLE 1

CLAIM STATUS

MODIFIED GRID CLAIMS

CLAIM NAME	RECORD NO.	UNITS	ANNIVERSARY DATE
WREN	3835	20	5 OCTOBER
SPARROW	3817	12	21 SEPTEMBER

2-POST CLAIMS

CLAIM NAME	RECORD NO.	ANNIVERSARY DATE
JAY	3819	21 SEPTEMBER
ROBIN	3820	21 SEPTEMBER
CROW	3821	21 SEPTEMBER

CASTLE MINERALS INC.

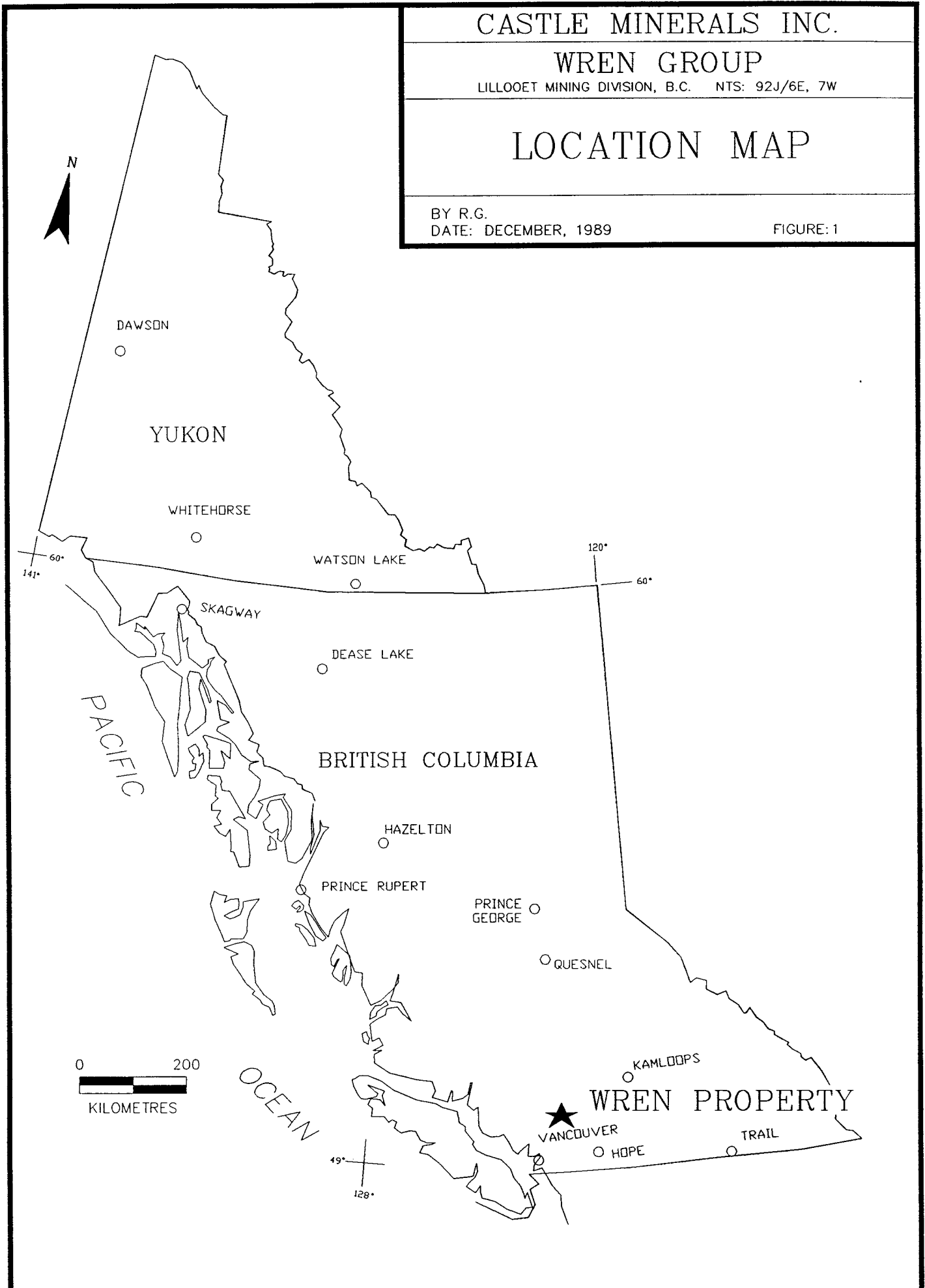
WREN GROUP

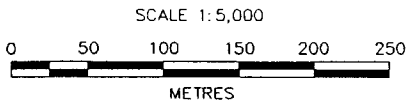
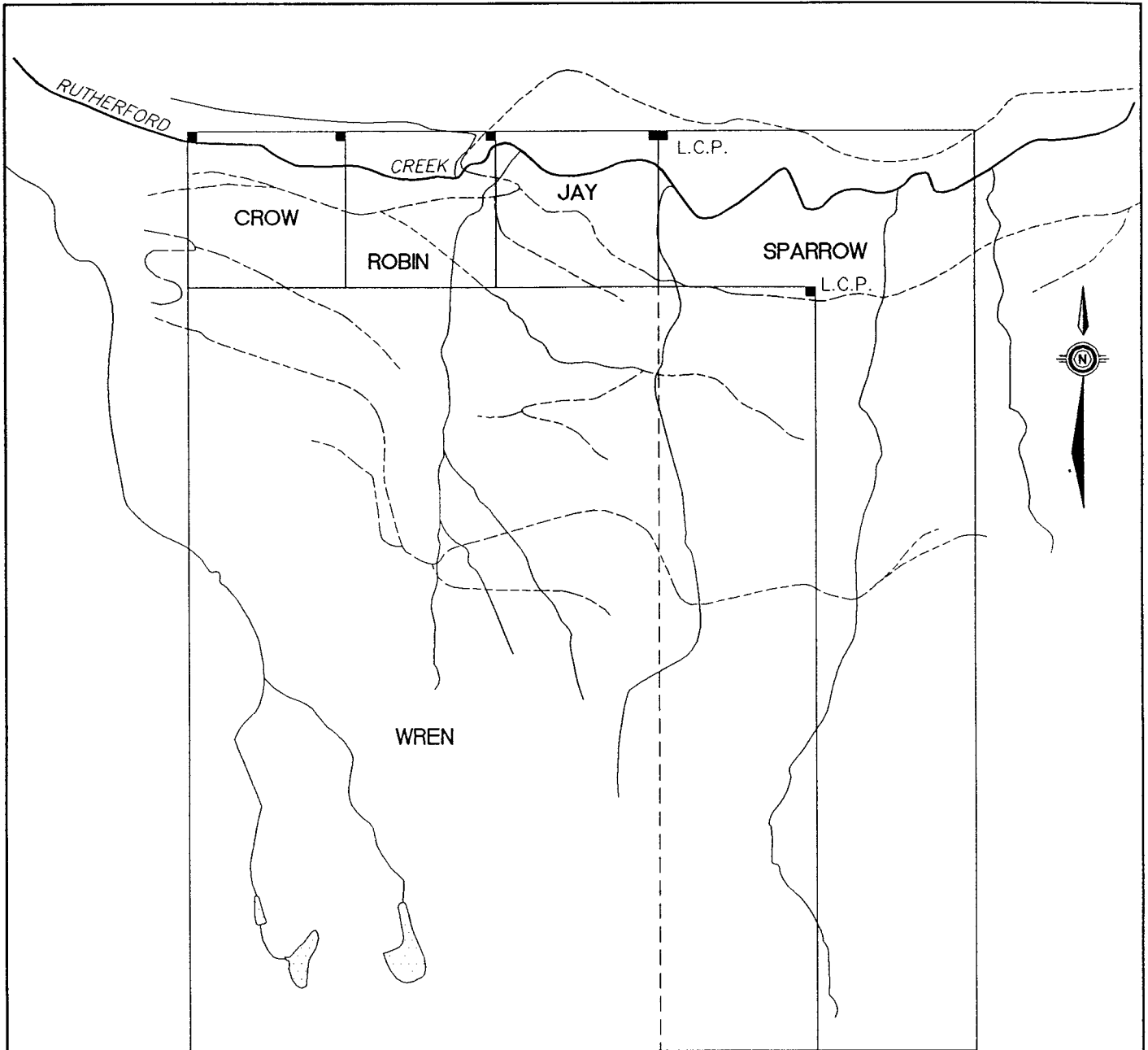
LILLOOET MINING DIVISION, B.C. NTS: 92J/6E, 7W

LOCATION MAP

BY R.G.
DATE: DECEMBER, 1989

FIGURE: 1





CASTLE MINERALS INC.	
WREN GROUP	
LILLOOET MINING DIVISION, B.C. NTS: 92J/6E, 7W	
CLAIM MAP	
BY: R.G./p.s. DATE: DECEMBER, 1989	FIGURE: 2

1.3 PHYSIOGRAPHY, CLIMATE AND VEGETATION

The Wren prospect is located in the Pacific Ranges Subdivision of the Coast Mountains Physiographic Province (formerly referred to as the Coast Plutonic Complex). The area surrounding the claims has a rugged topography with surface elevations ranging from 600 to over 2100 m (2000 to 7000 feet). Mountains rise abruptly on either side of Rutherford Creek valley; the highest peak on the property is approximately 2150 m (7060 feet) above sea level.

The climate during the summer is generally warm although brisk winds are common on unprotected ridges and peaks. The weather station at Pemberton Meadows (elev. 655 m) records a mean rainfall of 741 mm/year, a mean snowfall of 2824 mm/year, and a mean daily temperature varying from a low of -6.10C to a high of 18.60C. However, conditions are more severe at higher elevations. The area's climate is likened to that of the western interior of British Columbia (Drysdale, 1916).

Treeline is approximately 1600 m on north facing slope. At lower elevations cedar, cottonwood, white pine, Douglas fir, and hemlock fir are common with Douglas and hemlock fir being more common at higher elevations. Alpine fir, mosses and grasses are found above treeline.

2.0 GEOLOGY

2.1 REGIONAL GEOLOGY

The geology of the Pemberton map-area has been described by Woodsworth (1977), Cairnes (1925), Camsell (1918), and Drysdale (1916). The Rutherford Creek area has been studied in some detail by Woodsworth (1977), among others. The area is underlain mostly by granitoid rocks of the Coast Plutonic Complex and highly deformed volcanic and sedimentary rocks of Lower Cretaceous age.

The Coast Plutonic Complex consists largely of plutonic rocks and subordinate gneiss and migmatite, mostly of uncertain age. The plutonic rocks are dominantly quartz diorite to granodiorite, with some diorite and quartz monzonite. Regionally metamorphosed Late Triassic to Early Cretaceous sedimentary and volcanic rocks form northwest-trending roof pendants within the plutonic framework.

Highly deformed Lower Cretaceous aged stratified rocks are common with meta-volcanic rocks greatly predominate over meta-sedimentary strata. The volcanic rocks are mainly pyroclastic and are comprised of greenish tuffs and breccias, reddish brown to maroon breccia-conglomerates, and purplish breccias. Thin beds of brittle shale or siltstone are often interbedded with the volcanics.

A chain of late Tertiary and Quaternary calc-alkaline volcanic centres extends north through part of the Coast Plutonic Complex.

As in other parts of the Coast Mountains, the dominant structural trend is northwesterly. Foliation in plutonic rocks are generally northwest with steep dips. Schistosity in pendants is usually parallel or subparallel with contacts. Schistosity is rare in the meta-volcanics. It appears that deformation has been largely concentrated in narrow northwest trending zones, leaving the intervening areas with well preserved original textures suggest that deformation was controlled by deep-seated major structural features.

The geology of the area is not simple. Multiple deformation has rendered most of the rocks schistose and tightly compressed in complex repetitive folds. A subtlety of rock differences, and obscurity of bedding, facies changes in some formations, and a variation in intensity of hydrothermal alterations all combine to make a complex relationship which poor exposures, at lower elevations, further compounds.

The geology within the Wren Claim block appears to be divided into three parts, sediments to the east, volcanics in the centre, and quartz diorite to the west. The sediments are mostly fissile shales, black carbonaceous argillites, and minor calcareous shales or limestone. The volcanics are dominantly flows and pyroclastic andesites with minor flow breccias. The principal plutonic rock is a coarse-grained, quartz diorite within several high-level, quartz monzonite stocks intruding the quartz diorite.

3.0 PREVIOUS WORK SUMMARY

The area was first staked in the mid-1970's by the Rainbow Syndicate, a syndicate consisting of Newmont Exploration of Canada Ltd. (40%); Union Oil Company of Canada Ltd. (Calgary) (40%); Bethlehem Copper Corporation (20%); and John McGoran, (geologist). The area was staked as the GL Claims after a regional stream sediment sampling programme identified anomalous zinc and gold in the Rutherford Creek drainage. From 1977 to 1980, the property was geologically mapped and soil sampled. A geochemically anomalous area 200 X 250 metres was outlined and contained values up to 780 ppb gold. Panning the soils within the anomalous area returned visible flakes of angular gold. In 1980, an I.P. survey (a single-line, test survey) was conducted over the anomalous area and a 100 m long anomaly, believed to be disseminated pyrite, was outlined. This anomalous zone was below the area where gold had been panned from the soils. A gasoline powered, underground-slusher was mobilized onto the property, and a small trench was dug across the anomalous zone. This trench exposed a silicified, pyrite-bearing shear zone, but rock samples from the trench carry only low gold values. Two drill holes were proposed to test the I.P. anomaly at depth; however, the Syndicate was dissolved prior to the drilling, and the property was returned to Mr. McGoran who later allowed the claims to lapse.

As soon as the ground was open to staking, the core area of the original GL claims was covered by the Wren Claim and optioned to **CASTLE MINERALS INC.**

In 1987, **CASTLE MINERALS** relocated the Syndicate's trench and established a grid over the northern portions of the property east of the trench. Logging activity, especially road building, has exposed the shear zone in several widely spaced road cuts and consequently greatly enlarged its surface dimensions of the original showing. Grid lines 50 m (164 feet) apart were cut over the lower slopes of the Wren and Sparrow Claims. The grid was established to expand the area of known gold mineralization. All grid lines were soil or rock chip sampled at 20 m intervals. In addition to the grid sampling, all logging roads crossing the claim group were sampled at 20 m or 40 m intervals. Approximately 14 line km of grid lines and road traverses were sampled and a total of 899 samples were collected and analysed.

Results of the geochemical programme were very encouraging. Samples ranged from 1 ppb to 5690 ppb. With an anomalous threshold arbitrarily set at 100 ppb gold, over 15% of the samples are anomalous. The results of the geochemical survey were sufficiently encouraging to warrant drill testing some of the anomalous areas. A small programme of 5 drill holes totalling approximately 200 metres was proposed, and in mid-May a BBS-1 drill rig was mobilized to the property.

4.0 DIAMOND DRILL PROGRAMME

Previous geochemical soil sampling has identified an area approximately 200 X 250 metres which is highly anomalous with respect to gold. Angular gold particles were also recovered by panning the soils within the anomalous area. A ground geophysical survey (I.P. survey) outlined a pyrite zone near the centre of the anomalous area, and a small trench, constructed in the area of highest gold values, exposed a silicified shear zone containing pyrite. Subsequent road building has greatly expanded the surface exposure of the silicified shear zone.

Exploration by **CASTLE MINERALS** confirms the previous work and indicates that several shear zones, the widest is approximately 750 metres wide, are present and extend beyond the claim boundaries in both the northern and southern directions.

Five, short drill holes, totalling 219.6 metres (721 feet) were completed. Table II summarizes the drill data and Figure 3 shows the 1989 drill hole locations. All of the core was split, logged, and assayed. On the average, all core was divided into 3 metre (10 feet) sections, split lengthwise, and sent to Acme Analytical Labs. Ltd. for assaying. The core is in temporary storage in West Vancouver. Diamond-drill logs are presented in Appendix A, and assay results are presented in Appendix B.

TABLE II
SUMMARY OF DIAMOND DRILL DATA

Core diameter: AQ wire line

HOLE NO.	AZIMUTH	DIP	DEPTH
89-1	2280	-700	32.6 m (107')
89-2	2600	-600	60 m (197')
89-3	2550	-600	61.2 m (201')
89-4	1550	-470	43.3 m (142')
89-5	2600	-450	22.5 m (74')

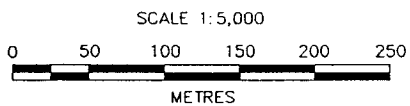
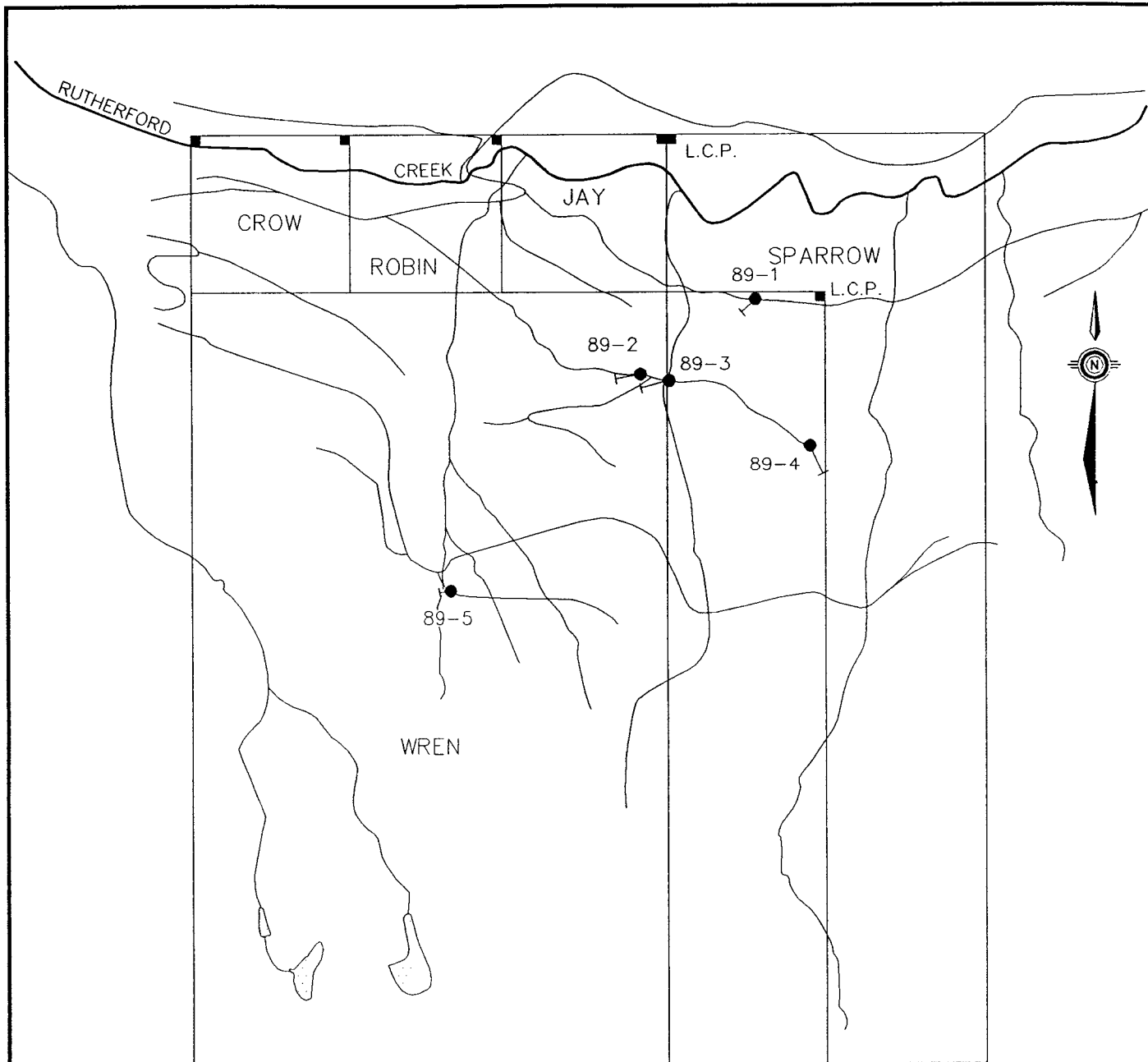
Hole 89-1 was collared to test a gossanous outcrop within an area of anomalous soil and rock samples. The rock was highly fractured and core recovery was poor; however, the bedrock appeared to be a silicified, light greyish-green andesite containing approximately 2 per cent pyrite. Although gold values were generally low, fault gouge (at 21.3 and 27.4 metres) returned 102 and 140 ppb Au and 488 and 722 ppm W respectively.

Holes 89-2 and 89-3 were collared a few tens of metres apart and were designed to test an area of high gold geochemistry in rock and soil samples. Hole 89-2 was collared west of Hole 89-3 and was to test what was believed to be a narrow, pink-coloured monzonite dyke within a zone of highly anomalous soil samples. Except of the first 10 metres, which was andesite, the entire hole returned monzonite with minor, narrow (1-3 cm) gold-bearing quartz veins. Hole (89-3) was to test the faulted contact between volcanics to the east and plutonics to the west. This hole intersected andesite to 21.9 metres with the balance of the hole in monzonite. Minor pyrite and pyrrhotite and traces of chalcopyrite were present in the volcanics. The monzonite showed minor propylitic alteration and traces of chalcopyrite. Although the volcanics were commonly sheared, there was no indication of a fault contact between the two igneous rocks. As in Hole 89-2, gold was confined to narrow quartz veins up to 3 cm wide.

Hole 89-4 was drilled to test a large area on anomalous soil samples. The core is composed of argillite to 10.7 m, and the balance of the core was altered andesite. Two 10-foot sections of the argillite contained numerous, quartz stringer which returned gold assays of 185 and 600 ppb.

Hole 89-5 was drilled to test a gold-sulphide and gold bearing shear zone up slope from the area trenched by the previous claim holders. The core contained altered andesite with approximately five per cent pyrite and only traces of gold. The anomalous surface samples were not explained.

The results from the drilling programme indicate that gold is present within fault gouge and appears to be associated with pyrite, in narrow, quartz-veins in the plutonic rocks, and in quartz stringers within an argillite host.



CASTLE MINERALS INC.

WREN GROUP

LILLOOET MINING DIVISION, B.C. NTS: 92J/6E, 7W

DRILL HOLE LOCATIONS

BY: R.G.
DATE: DECEMBER, 1989

FIGURE: 3

5.0 COSTS STATEMENT

CASTLE MINERALS INC.
 WREN CLAIM GROUP
 RUTHERFORD CREEK AREA
 15 MAY TO 15 JULY 1989

GENERAL COSTS

FOOD & ACCOMMODATION:	
4 persons, 67 man-days @ \$45/day	\$12,060.00
SALARIES	21,353.00
VEHICLES/TRANSPORTATION	14,886.00
SUPERVISION:	5,750.00
ENGINEERING AND CONSULTING:	2,802.86
CAMP FUEL:	250.00
ROAD MAINTENANCE & DRILL SITE PREP.	1,300.00
DRILL MOB.-DEMOB.	2,000.00
DRILL FUEL, OIL, & MAINTENANCE	2,500.00
DIRECT DRILLING COSTS	
incl. drill bits, repairs & drill steel	
721 feet @ \$15.54/ft.	11,207.07
HELICOPTER:	
Pemperton 206B, 2 hrs. @ \$600/hr	1,200.00
ASSAYING:	
Acme Analytical Labs.	
70 core samples @ \$14.62 ea	1,023.50
TOTAL COSTS	\$ 76,332.43

6.0 REFERENCES

- Cairnes, C.E., 1925; Pemberton area, Lillooet District, British Columbia: Geol. Surv. Can., Summary Report, 1924, Pt. A, p. 76-99.
- Camsell, C. 1918; Reconnaissance along the Pacific Great Eastern Railway between Squamish and Lillooet: Geol. Surv. Can., Summary Report, 1917, Pt. B, p. 12-23.
- Drysdale, C.W., 1916; Bridge River Map Area, Lillooet Mining Division, B. C.: Geol. Surv. Can. Summary Report, 1915, p. 75-85.
- Gonzalez, R.A., 1988; Geologic Report on the Wren Claim Group, Rutherford Creek Area, Lillooet Mining Division, B.C.: Engineer's Report for Castle Resources Inc. 1988 Prospectus, 19 pp.
- Gonzalez, R.A., 1988; Assessment Report on the Wren Claim Group, Rutherford Creek Area, Lillooet Mining Division, B.C.: B.C. Dept. of Mines Assessment Rpt., 25 pp.
- McCann, W.S., 1922; Geology and mineral deposits of the Dridge River map-area, British Columbia: Geol. Surv. Can., Mem. 130, pp. 115
- McGoran, John, 1978; Geological Report-G.L. Mineral Claim-Record No 552-Rutherford Creek Area: B.C. Dept. of Mines, Assessment Rpt., 6976.
- McGoran, John, 1979; Geochemical Report-G.L. 1 - G.L. 5 inclusive-Rutherford Creek Area: B.C. Dept. of Mines, Assessment Rpt., 7648.
- McGoran, John, 1979; Rainbow Syndicate Report-1978 Field Season: Unpub. Rpt., 16 pp.
- Woodsworth, G.J., 1977: Geology Pemberton (92J) map-area, Geol. Surv. Can., Open File 482.
- Woodsworth, G.J., Pearson, D.E., and Sinclair, A.J., 1977: Metal distribution patterns across the eastern flank of the Coast Plutonic Complex, south-central British Columbia: Econ. Geol. v. 72, p. 170-183.

7.0 CERTIFICATE

I, R. A. Gonzalez, do hereby certify that:

1. I am a geologist and reside at 2784 Lawson Ave., West Vancouver, British Columbia.
2. I am a graduate of The University of New Mexico, U.S.A.; with a B.Sc. in geology (1965) and an M.Sc. in geology (1968).
3. I have practiced my profession since 1965 in Canada and abroad as indicated on the following page.
4. I am a Fellow in the Geological Association of Canada, Registration Number 4523.
5. I am a registered member of the Association of Professional Engineers of the Province of Manitoba, Registration Number 3970.
6. I have based this report on a personal examination of the property and on information obtained from the Geological Survey of Canada and engineering reports and other support documents provided by **CASTLE MINERALS INC.**
7. I have no interest, nor do I expect to receive any interest, either directly or indirectly in the securities or properties of **CASTLE MINERALS INC.**
8. I have no past or present, direct or indirect interest in any of the listed Mineral Claims or in any other property within the Lillooet Mining District.
9. This report may be used by **CASTLE RESOURCES LTD.** or their agents for a Statement of Material Facts or Shareholders' newsletter, etc. either in whole or in part.

Dated at Vancouver, British Columbia, this 27th day of DECEMBER 1989:



R. A. Gonzalez M.Sc., F.G.A.C.

8.0 STATEMENT OF PROFESSIONAL QUALIFICATIONS**R.A. GONZALEZ, M.Sc., F.G.A.C.****ACADEMIC**

1965	B.Sc. in Geology	The University of New Mexico, U.S.A.
1968	M.Sc. in Geology	The University of New Mexico, U.S.A.

PROFESSIONAL

1985	Adder Exploration & Dev. Ltd.	President
1983	Archean Engineering Limited	Overseas Manager
1980-1983	Placer Development y Cia. Ltd. (Chile)	Ass't Exploration Manager
1977-1980	Consultant attached to the Geological Survey of Malaysia	Ass't Project Manager on a C.I.D.A. supported mineral exploration survey over Peninsular Malaysia
1977	Registered with the Association of Professional Engineers of the Province of Manitoba	
1975-1977	Province of Manitoba	Resident Geologist for the Manitoba Dept. of Mines.
1971-1975	Giant Mascot Mines Limited	Senior Geologist
1970-1971	New Jersey Zinc (Canada) Ltd.	Exploration Geologist
1968-1970	Anaconda American Brass Ltd.	Research Geologist
1965-1966	Mex-Tex Mining Co. (U.S.A)	Geologist

9.0 APPENDIX A - DIAMOND DRILL LOGS

PROPERTY CASTLE MINERALS INC.
WREN CLAIMS

DIAMOND DRILL RECORD

HOLE NO 89-1 PAGE 1 OF 2

LATITUDE	DIPS - collar	AZIMUTH	STARTED
DEPARTURE	-	228°	COMPLETED
ELEVATION	-	AQ	LENGTH
SHEET NO.	-	CLAIM	LOGGED BY
TARGET			DATE
Anomalous AU in soil samples			

SECTION [ft]		ROCK DESCRIPTION	MINERALIZATION SUMMARY	ASSAYS				
FROM	TO			SAMPLE NUMBER	INTERVAL	WIDTH	AU [oz/t]	
0	20	Casing no core						
				801	20-30			
20	27	Light greyish-green, medium-grained andesite. Core highly broken with few fragments larger than 1cm. Some sections contain euhedral plagioclase 15% of total ground mass with crystals approximately 1mm long, -2% pyrite mostly along fracture (joints?) surfaces	-2% pyrite along joint surfaces	802	30-40			
		Minor -0.5% magnetite at 27'	-0.5% diss magnetite at 27'. Minor diss	803	40-50			
		21'-27' 1 foot of core recovered	FeOx (siderite?)	804	50-57			
		27'-30' 1.5 feet of core recovered	throughout	805	72-77			
		Fault at 19.5-20'		806	77-87			
		30'-32' 1 foot of core recovered		807	87-92			
		32'-37' 0.5 feet of core recovered						
		37'-47' 0.1 feet of core recovered						
		47'-49' 1 foot of core recovered						
		49'-52' 2.5 feet of core recovered						
		52'-72' 0 feet of core recovered						
		72'-75' 2 feet of core recovered						
		75'-77' 0.6 feet of core recovered						
		77'-80' 2.3 feet of core recovered						
		80'-82' 1.2 feet of core recovered						
		82'-92' 1.5 feet of core recovered						
		92'-97' 0 feet of core recovered						
		97'-107' 0 feet of core recovered						

Diamond Drill Record

HOLE NO. 89-1

Page 2 of 2

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from m ft	to m ft		from m ft	to m ft		Thickness mm	Angle to core	minerals in decreasing abundance
27	30	Fault zone-abundant clay gouge. Light grey 50% clay	27	30	3-5% discontinuous veinlets of pyrite. Veinlets -0.5mm wide			
30	79	Light greyish-green lithic andesite tuff with rounded fragments upto 4mm long	30	79	Intensely altered to a mixture of light grey mica(?) (probably a mylonite) with minor calcite along fractures			-1% diss. euhedral pyrite
		37-47 Fault zone with no core recovered						
		47-49 Core ground to pea sized fragments	72	75	Core fractured at 75°t.c.a. and 25°t.c.a..Also most intense			5% diss. pyrite
		52-72 Fault zone with no core recovered			zone of mylonization			
79	83	Massive light greenish-grey andesite						1-2% pyrite crystals +20% euhedral plagioclase
		81-83 Highly broken core						crystals upto 1mm long. No apparent sulphides
83	92	Core recovery is -15% fragments are plate like 1-4mm thick with fractures possibly at 45°t.c.a. fragments seldom with surface area +1cm. Medium grained, greenish lithic andesitic tuff. fragments average 3-4mm in diameter and are usually rounded although some appear strained and angular						
92	107	No core						

Diamond Drill Record

LOCATION:		Diamond Drill Record				HOLE NO. 89-2	Page 1 of 3	
AZIMUTH:	260°	DIPS - collar	-60°	CONTRACTOR:		PROPERTY:		
ELEVATION:		-	m	LOGGED BY: Ralph Gonzalez		CLAIM NO.		
LENGTH:	197 feet	-	m	DATE:		SECTION NO.		
CORE SIZE:	AQ	-	m			STARTED:		
PURPOSE: High Au geochemical respose in rock and soil						COMPLETED:		
Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from (ft)	to (ft)		from (ft)	to (ft)		Thickness mm	Angle to core	minerals in decreasing abundance
0	3	Casing - no core						
3	27	Light greyish-green andesite: porphyritic texture with plag. crystals upto 0-5cm in length. Weak to moderately magnetic. Locally pink quartz? amygdules upto 2mm	3	18	Fractured @ 50°t.c.a. with epidote alteration and discontinuous quartz veins perpendicular to fracture. Alteration zone range to 1.5cm. Fracture density 20-25/m	1-3mm	50°	Quartz veinlets with pyrite parallel to veinlet and extending into wall rock ±0.5cm Pyrite crystals anhedral to subhedral and 1-2mm across
					from 7.5ft to 18ft intense epidote alteration. 20% of core in 1cm veins and -0.5mm veinlets. Approx. 1% pyrite with an increase to 2-3% in more altered sections	6cm		@ 14' Quartz vein - no sulphides!
						14cm		@ 14' FeOx altered core
			18	27	Same as above except reduced -5% epidote alteration	1-3mm		Quartz, plag, epidote veins with -1% pyrite. Only trace amounts of pyrite in the andesite
			26	26.5	10cm FeOx and muscovite altered core			
27	31.5	Light greyish-green, massive, silicified andesite				0.5	50°	75% pyrite as euhedral to subhedral disseminate crystals and as anhedral grains adjacent to quartz veinlets

Diamond Drill Record

HOLE NO. 89-2

Page 2 of 3

Section		ROCK DESCRIPTION	Interval		ALTERATION. MINERALIZATION etc.	VEINLETS		
from (ft)	to (ft)		from (ft)	to (ft)		Thickness mm	Angle to core	minerals in decreasing abundance
			27	31.5	The lower contact is fractured and altered to quartz, plag. micas and uralitized hornblende FeOx increases toward to lower contact and pyrite decreases over the last foot			
31.5		Lower contact 30° t.c.a. fractured			Traces -0.5% chalcopyrite with FeOx along the rim. Crystals are usually euhedral			
31.5	141	Coarse grained greyish pink monzonite. Hornblende (mafics) are altered to chlorite			All of the core shows moderate-to weak propylitic alteration with local phyllic alteration			
		41.5-43.5 Andisite with 5-8% pyrite similar to above			Local phyllic alteration			
		Monzonite fractured 30° t.c.a. averaging 5-7/m Most of the core is weak to very weakly magnetic	57	57	4cm of quartz vein contact 85° t.c.a. Mafics altered to chlorite and a yellowish epidote (?)			
			84	89	Colour change increase in pink (FeOx increase) and intense alteration of the mafics to micas (sericite)			Sulphides weathered to FeOx
			98	98	4cm quartz vein 40° t.c.a.			
			109	109	3cm quartz vein			
			106	106	2cm quartz vein 90° t.c.a.			
			111	112	Sericite alteration			
			120		3cm quartz vein			
			137		3cm quartz vein pyrite			
			127	128	Bleached core with slight increase of chalcopyrite			

Diamond Drill Record

HOLE NO. 89-2 Page 3 of 3

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from (ft)	to (ft)		from (ft)	to (ft)		Thickness mm	Angle to core	minerals in decreasing abundance
141	197	Coarse grained monzonite grey to pink in colour, non-magnetic to very weakly magnetic. The grey coloured sections reflect and increase in sericite alteration. Generally, however, the core is weakly propylitically altered with mafics + epidote. The sericite development may in part be due to quartz veins 141-145 Sericite altered monzonite with 1ft quartz vein @ 143-144 157-172 FeOx abundant and fractures @ 20/m with fractures @ 75° t.c.a.	135	137	Bleached core with muscovite Chalcopyrite 0.5% with FeOx rims FeOx rims increase in the more altered sections. 141-148, 157-172, 176-180			
			143	144	28cm quartz vein 85° t.c.a.			
			172	172	Poor core recovery (50%) between 162-172 feet 3cm quartz part of			

Diamond Drill Record

LOCATION:				HOLE NO. 89-3	Page 1 of 2
AZIMUTH: 255°	DIPS - collar	-60 °	CONTRACTOR:	PROPERTY:	
ELEVATION:	-	m °	LOGGED BY: Ralph Gonzalez	CLAIM NO.	
LENGTH: 201 feet	-	m °	DATE: June 30, 1989	SECTION NO.	
CORE SIZE: AQ	-	m °		STARTED:	
PURPOSE:				COMPLETED:	

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from (ft)	to (ft)		from (ft)	to (ft)		Thickness mm	Angle to core	minerals in decreasing abundance
0	10	Casing, no core recovered						
10	31	Mixture of light green, fine grained andesite; greyish, porphyritic andesite; and pink to brownish, coarse grained monzonite			-0.1 pyrite and local pyrrhotite, odd grain of chalcopyrite. Most of the andesitic is weakly to moderately magnetic-monzonite is non-magnetic			
		11ft and again at 12ft 5-6cm core of monzonite						
		14.5-16ft Monzonite						
		16-26 Shear zone 10% core						
		26-27 Porphyry andesite						
		27-30 Shear zone with 30% recovery						
31	72	Grey, moderately grained (lapilli?) tuff. with inclusion upto 4mm across			Sulphides are 1% pyrrhotite after pyrite and very minor chalcopyrite, pyrrhotite forms			30° and 0° wispy veinlets and fracture filling veinlets @30° t.c.a., epidote and quartz are the principal material with
		The bottom 1ft is more silicified and contains 5-10% muscovite. At the contact with the underlying unit is 7cm quartz vein			casts and along veinlets where they are highly altered to FeOx			minor pyrrhotite
					Some fragments(?) upto 3cm are incorporated in the tuff and are altered entirely to epidote			
72	201	Coarse grained, pink to grey monzonite fractures @ 70° with density of 20/m			Mafics are altered to chlorite and epidote			Traces -0.1% of chalcopyrite
					Pervasive micas are common in the grey sections			

Diamond Drill Record

HOLE NO. MRO 89-3 Page 2 of 2

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from	to		from	to		Thickness mm	Angle to core	minerals in decreasing abundance
			116	116	3cm quartz vein-increased mica for 4cm on either side of quartz vein			
			127	127	7cm quartz vein-contact 90°t.c.a.			
			128	135	Sericite alteration. The core is grey in color and FeOx alteration. Commonly extends 2-3cm on either side of fractures Fractures average 10/m 45-60°t.c.a. 0.1% chalcopyrite with 0.5% chalcopyrite in FeOx zones			
			142	152	Sericite alteration-the core is grey in color with 8 fractures per metre. Fractures 40°t.c.a. several of which have 1-2mm thick epidote veinlets along the face			

Diamond Drill Record

HOLE NO. 89-4

Page 2 of 2

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from	to		from	to		Thickness mm	Angle to core	minerals in decreasing abundance
92	142	Light greenish-grey andesite with -2% plagioclase crystals upto 4mm long. Core is massive with fractures at 35° to core and 9/metre None of the core was magnetic	115	117	Vuggy core with MnOx filling the cavities			
EOH			121	142		1-2	0,10	Stringers of quartz, carbonate and epidote
			137	142	The only sulphides seen: 1-3% pyrite euhedral crystals 1-2mm across. The most abundant amount is at the top 137' decreasing downward			

Diamond Drill Record

Section		ROCK DESCRIPTION	Interval		ALTERATION, MINERALIZATION etc.	VEINLETS		
from (ft)	to (ft)		from (ft)	to (ft)		Thickness mm	Angle to core	minerals in decreasing abundance
43	74	Light grey, f/g massive andesite slightly metamorphosed with plagioclase and sulphides parallel to schistosity 51-60 Lithic tuff with no visible sulphides 68-70 Light green tuff with local clots and stringers of chlorite Some of the more greenish material is very talc like to the touch 70-75 core lost			Pyrite occurs as stringers and clusters parallel to the schistosity 30-40° to core Pyrite locally may be as high as 5% but usually averages about 1-2% Disseminated and clusters of pyrite parallel to schistosity			Discontinuous quartz and quartz carbonate veinlets 1-3mm across the schistosity 70° to core 5 veinlets/meter

10.0 APPENDIX B - ASSAY RESULTS

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: JUN 7 1989

DATE REPORT MAILED:

June 12/89

SIGNED BY:

C. Long

D. TOYE, C. LEONG, J. WANG: CERTIFIED B.C. ASSAYERS

CASTLE MINERALS INC. PROJECT WREN CLAIMS File # 89-1369

89-1

SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Am	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPM
801 20-30	2	211	16	201	.3	8	12	825	4.19	2	5	ND	2	23	1	2	2	41	.21	.074	4	10	2.55	135	.01	2	3.50	.03	.10	1	45
802 30-40	1	159	12	215	.2	15	17	958	4.91	2	5	ND	1	26	1	2	2	59	.45	.074	5	21	3.22	205	.03	2	4.55	.03	.09	1	34
803 40-50	1	193	10	188	.3	2	8	225	2.98	7	5	ND	1	12	1	2	2	21	.16	.073	2	1	.99	71	.01	2	1.75	.04	.23	1	54
804 50-57	2	173	10	100	.1	6	9	307	3.44	4	5	ND	1	14	1	2	2	33	.11	.066	2	5	1.21	96	.01	2	2.31	.05	.31	1	48
805 72-77	3	214	10	100	.2	3	12	645	3.63	6	5	ND	2	19	1	2	2	20	.20	.081	4	1	1.59	88	.01	2	2.34	.04	.17	1	43
806 77-87	1	121	30	102	.4	27	24	1143	3.58	8	5	ND	1	67	1	3	2	59	2.15	.064	4	57	3.47	79	.12	2	4.01	.01	.04	1	14
807 87-92	1	151	12	169	.2	5	9	983	3.97	6	5	ND	2	36	1	3	2	34	.22	.080	10	3	2.48	118	.01	2	3.78	.06	.19	1	19
STD C/AU-R	18	63	41	132	6.6	73	31	1022	3.83	42	22	7	37	51	19	15	18	59	.47	.090	38	56	.98	173	.07	35	1.83	.06	.13	12	490

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: CUTTING AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: MAY 29 1989

DATE REPORT MAILED: June 1/89

SIGNED BY: *C. Long* D. TOYE, C. LEONG, J. WANG: CERTIFIED B.C. ASSAYERS

CASTLE MINERAL INC.

File # 89-1242

*returning
sludge
from
fault
zones.*

SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPB
89-1 70'	9	771	22	170	1.3	50	34	592	8.05	2	5	ND	1	56	1	3	2	30	.29	.085	3	20	1.18	7	.01	12	1.91	.05	.16	488	102
89-1 90'	11	525	13	194	.2	63	27	614	6.48	6	5	ND	1	36	1	4	2	24	.22	.075	2	14	1.01	15	.01	2	1.66	.04	.15	722	140

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: JUN 29 1989

DATE REPORT MAILED: July 3/89

SIGNED BY: *C. Long*...D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

CASTLE MINERALS LTD.

File # 89-1855

SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	AU*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPM
89-2 3-12	1	101	16	137	.1	7	11	1055	3.60	2	5	ND	1	52	1	2	3	92	.65	.062	3	10	1.07	254	.17	11	2.02	.05	.64	1	18
89-2 12-20	3	87	17	169	.1	6	13	1156	2.96	4	5	ND	1	41	1	2	2	49	.49	.058	4	20	1.06	68	.07	13	1.67	.02	.12	1	5
89-2 20-27	3	41	11	140	.1	8	13	1182	2.71	2	5	ND	1	49	1	2	2	48	.61	.063	4	9	.93	74	.06	5	1.62	.02	.13	1	5
89-2 27-31.5	1	72	16	149	.1	8	14	1583	4.67	2	5	ND	1	11	1	2	2	58	.49	.052	5	12	1.38	35	.01	3	2.26	.01	.13	1	2
89-2 31.5-42	4	64	12	101	.2	6	9	853	2.53	2	5	ND	1	21	1	2	3	31	.20	.028	5	8	.76	91	.02	7	1.40	.02	.12	1	3
89-2 42-52	1	74	10	80	.1	5	6	728	2.32	2	5	ND	1	31	1	2	2	30	.55	.041	7	20	.58	132	.03	12	1.18	.02	.13	1	6
89-2 52-62	2	35	11	40	.1	5	3	451	.90	6	5	ND	1	29	1	2	2	5	.45	.027	7	5	.18	116	.01	4	.58	.02	.13	3	4
89-2 62-72	2	37	10	40	.2	4	3	389	.88	5	5	ND	1	33	1	2	2	6	.53	.036	6	6	.21	137	.01	4	.62	.02	.11	3	12
89-2 72-82	2	42	6	36	.2	4	3	349	.81	2	5	ND	1	35	1	2	2	4	.72	.034	6	5	.13	203	.01	3	.49	.02	.12	1	8
89-2 82-92	2	41	11	40	.1	4	3	380	.91	7	5	ND	2	27	1	2	2	5	.44	.035	8	18	.15	119	.01	6	.55	.02	.12	3	139
89-2 92-102	3	37	11	41	.4	5	3	417	.99	5	5	ND	2	29	1	3	2	4	.40	.032	9	5	.13	168	.01	9	.54	.02	.12	3	10
89-2 102-112	3	62	10	36	.3	6	3	394	.91	5	5	ND	2	36	1	2	2	4	.40	.032	7	7	.14	178	.01	9	.57	.02	.14	3	16
89-2 112-122	6	30	7	42	.3	6	3	528	.97	3	5	ND	2	19	1	2	2	3	.36	.036	9	7	.11	98	.01	10	.49	.02	.14	3	4
89-2 122-132	1	73	8	30	.1	3	2	491	.83	2	5	ND	1	55	1	3	2	3	1.31	.034	7	22	.10	104	.01	7	.44	.02	.14	1	13
89-2 132-142	2	77	7	40	.3	4	3	444	.80	3	5	ND	2	42	1	2	2	3	.90	.045	7	5	.15	87	.01	7	.51	.02	.12	1	5
STD C/AU-R	18	60	43	132	6.9	71	31	1015	4.09	41	22	7	37	50	19	15	21	60	.49	.093	39	55	.84	183	.07	38	1.96	.06	.13	12	490

alteration?
zone

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: JUL 4 1989

DATE REPORT MAILED:

July 7/89

SIGNED BY: *C. Long* . . . D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

CASTLE MINERAL INC File # 89-1899

SAMPLE#	Mc	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	PPM	PPB	
89-2 142-152	4	36	7	35	.1	6	2	348	.66	2	8	ND	1	25	1	3	4	3	.26	.033	6	7	.11	84	.01	4	.44	.02	.12	3	6
89-2 152-162	3	26	11	38	.1	4	3	421	.80	4	7	ND	2	28	1	2	2	5	.22	.031	7	22	.12	106	.01	5	.64	.02	.14	4	2
89-2 162-172	5	49	12	46	.3	6	3	528	.98	28	5	ND	3	20	1	3	2	5	.51	.033	10	6	.09	112	.01	12	.57	.02	.14	3	5
89-2 172-182	4	31	13	48	.2	7	2	406	.84	7	5	ND	1	33	1	3	2	5	.42	.036	8	8	.15	102	.01	5	.67	.03	.16	1	2
89-2 182-187	2	23	10	40	.1	6	3	359	.70	2	5	ND	1	42	1	2	3	4	.80	.035	8	6	.15	90	.01	5	.57	.03	.13	1	2
89-3 10-16	1	42	12	70	.1	6	8	710	2.27	2	5	ND	1	52	1	2	4	42	1.07	.053	7	28	.70	92	.06	8	1.33	.05	.19	1	2
89-3 16-27	1	86	16	105	.2	5	8	1117	2.67	2	5	ND	1	51	1	2	3	64	.62	.053	6	9	.90	288	.17	11	1.98	.09	.91	1	1
89-3 27-33	2	60	22	99	.1	7	9	945	2.11	2	5	ND	1	34	1	2	2	38	.65	.054	6	11	.81	105	.09	2	1.57	.05	.35	1	7
89-3 33-43	1	47	24	150	.1	8	10	1257	2.75	2	5	ND	1	33	1	2	2	47	.73	.058	6	11	1.19	92	.09	4	1.87	.04	.32	1	3
89-3 43-53	1	56	12	77	.4	5	9	864	2.71	2	7	ND	2	89	1	2	2	70	1.00	.061	6	22	.90	366	.16	8	1.99	.11	.59	1	2
89-3 53-63	2	53	20	108	.2	6	10	833	2.56	2	5	ND	1	55	1	2	3	62	.82	.070	5	11	.99	211	.14	5	1.74	.07	.41	1	3
89-3 63-73	2	47	72	133	.1	5	6	638	1.97	2	5	ND	1	38	1	2	2	29	.61	.043	6	7	.54	198	.08	2	1.02	.04	.27	1	9
89-3 73-83	2	12	8	45	.1	3	3	463	.91	2	5	ND	2	36	1	2	2	6	.36	.032	7	6	.23	80	.01	9	.78	.03	.13	2	2
89-3 83-93	2	13	5	39	.1	4	3	456	.85	3	5	ND	2	34	1	2	2	5	.59	.032	8	6	.19	82	.01	12	.73	.03	.14	1	1
89-3 93-103	2	17	8	33	.2	3	3	292	.69	2	5	ND	2	50	1	2	2	4	.31	.035	5	5	.14	156	.02	14	.61	.04	.15	3	2
89-3 103-113	3	22	7	41	.1	5	3	400	.82	2	5	ND	1	44	1	2	2	4	.39	.031	5	6	.17	140	.02	2	.63	.04	.15	1	3
89-3 113-123	3	47	7	42	.1	5	2	453	.93	2	5	ND	1	32	1	2	3	4	.45	.035	8	10	.16	147	.01	10	.64	.04	.15	2	46
89-3 123-133	4	74	6	38	.1	5	2	554	.94	3	5	ND	1	37	1	2	2	4	.80	.036	9	6	.13	98	.01	20	.53	.03	.15	1	41
89-3 133-143	1	68	6	34	.1	2	3	354	.73	4	5	ND	1	47	1	2	2	4	.92	.039	8	4	.15	222	.01	2	.52	.02	.11	1	10
89-3 143-153	2	66	6	33	.1	5	3	317	.74	2	5	ND	1	55	1	2	2	4	.92	.036	8	7	.14	253	.01	7	.57	.03	.13	1	2
89-3 153-160	1	81	6	40	.4	3	3	359	.75	2	5	ND	3	41	1	2	2	4	.53	.041	7	5	.17	82	.01	11	.60	.03	.13	1	7
STD C/AU-R	18	59	44	132	7.2	69	31	1020	4.04	42	19	7	37	50	18	15	24	60	.47	.091	39	55	.82	182	.07	35	1.93	.06	.14	11	520

*gt₃ veins
with some
well rock
alteration
adjacent
to gt₃ veins*

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: P1 CORE P2 SLUDGE AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: JUL 18 1989

DATE REPORT MAILED: July 25/89

SIGNED BY: *C. Long* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

CASTLE MINERALS INC.

File # 89-2237

Page 1

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPM	Co PPM	Mn PPM	Fe %	As PPM	U PPM	Au PPM	Th PPM	Sr PPM	Cd PPM	Sb PPM	Bi PPM	V PPM	Ca %	P %	La PPM	Cr PPM	Mg %	Ba PPM	Ti %	B PPM	Al %	Na %	K %	W PPM	Au* PPB
89-2 192	13	13	12	5	.3	28	1	108	1.19	21	5	ND	1	1	1	4	2	1	.01	.004	2	33	.01	13	.01	2	.02	.01	.03	9	11
89-4 14-24	2	36	14	104	.1	5	8	859	7.13	11	5	ND	3	20	1	2	2	28	.34	.150	6	3	.74	27	.03	3	3.22	.02	.16	1	185
89-4 24-34	1	60	18	141	.2	9	13	1135	6.85	9	5	ND	2	18	1	4	2	52	.26	.083	4	6	1.41	33	.03	5	3.87	.01	.10	1	600
89-4 34-44	1	63	37	211	.1	10	16	2226	6.84	8	5	ND	2	32	1	4	2	100	.46	.116	3	8	2.08	27	.07	4	4.32	.05	.09	1	6
89-4 44-54	1	102	34	213	.3	10	23	2290	7.08	12	5	ND	1	15	2	6	2	86	.33	.088	2	7	2.21	29	.08	2	4.62	.02	.09	1	18
89-4 54-64	1	56	41	171	.1	9	17	2044	5.34	11	5	ND	1	16	1	2	2	65	.37	.093	3	5	1.94	42	.09	2	3.74	.03	.14	1	4
89-4 64-74	1	86	14	123	.1	7	14	1544	5.22	3	5	ND	2	17	1	4	2	43	.90	.070	4	4	1.53	40	.05	2	3.24	.02	.16	1	1
89-4 74-84	1	50	11	117	.3	9	14	1427	4.66	15	5	ND	2	10	1	5	2	36	.33	.078	3	4	1.61	34	.06	2	3.03	.02	.15	1	3
89-4 84-94	1	113	11	144	.3	8	16	1690	5.28	8	5	ND	1	15	1	2	2	48	1.33	.076	3	4	1.59	37	.06	2	3.35	.02	.17	1	3
89-4 94-104	1	70	11	137	.1	7	16	1646	5.51	8	5	ND	1	11	1	2	2	41	1.33	.072	2	3	1.39	30	.06	2	3.07	.02	.14	1	2
89-4 104-114	1	49	8	142	.2	7	21	1406	5.26	4	5	ND	1	9	1	2	2	32	.57	.078	2	3	1.47	32	.05	2	2.95	.01	.17	1	4
89-4 114-124	1	28	10	125	.2	6	12	1084	3.02	5	5	ND	1	37	1	4	2	30	1.26	.074	2	2	1.75	30	.09	2	2.31	.03	.10	1	2
89-4 124-134	1	24	6	111	.1	4	12	1070	3.08	2	5	ND	1	37	1	2	2	29	1.72	.073	2	2	1.67	44	.10	2	2.25	.03	.12	2	1
89-4 134-144	1	23	6	119	.1	5	11	1113	2.92	6	5	ND	1	35	1	2	2	25	1.22	.073	2	2	1.56	57	.07	2	2.07	.02	.11	1	3
STD C/AU-R	18	58	38	132	6.9	69	31	1024	3.95	42	20	7	36	48	19	14	22	61	.47	.097	38	53	.93	172	.07	35	1.98	.06	.13	11	490

CASTLE MINERALS INC. FILE # 89-2237

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPM	Co PPM	Mn PPM	Fe %	As PPM	U PPM	Au PPM	Th PPM	Sr PPM	Cd PPM	Sb PPM	Bi PPM	V PPM	Ca %	P %	La PPM	Cr PPM	Mg %	Ba PPM	Ti %	B PPM	Al %	Na %	K %	W PPM	Au* PPB
DDH#4 15-20 A	3	94	12	135	.1	9	13	813	6.99	20	5	ND	4	34	1	2	2	24	.37	.114	10	4	.60	84	.03	2	2.94	.01	.13	5	11
DDH#4 20-25 A	3	96	11	94	.1	8	9	668	7.26	16	5	ND	4	27	1	2	2	21	.35	.136	10	5	.51	62	.02	2	2.98	.02	.15	7	8
DDH#4 15-20 B	2	89	12	144	.1	9	14	836	7.49	16	5	ND	4	37	2	2	2	24	.42	.117	10	5	.61	97	.03	2	3.19	.02	.13	2	6
DDH#4 20-25 B	4	105	12	99	.1	10	9	698	7.05	11	5	ND	4	33	1	2	2	22	.36	.132	10	6	.47	84	.02	2	2.94	.02	.22	11	3

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 NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: JUL 18 1989

DATE REPORT MAILED:

July 21/89

SIGNED BY.....

C. Leung

D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

CASTLE MINERALS INC.

File # 89-2235

SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPB
89-5 5-15	1	113	20	62	.5	7	13	929	5.01	10	5	ND	1	15	1	2	2	25	.14	.090	2	3	2.13	44	.10	2	2.19	.03	.10	1	3
89-5 15-25	1	134	38	61	1.7	16	16	1049	4.40	7	5	ND	1	14	1	2	2	40	.18	.075	2	42	2.64	34	.12	2	2.56	.03	.07	1	5
89-5 25-35	1	127	15	80	.3	41	30	1150	4.92	4	5	ND	1	41	1	2	2	117	.90	.088	3	202	4.59	4	.20	2	4.41	.01	.01	1	1
89-5 35-45	1	115	32	60	.8	37	24	650	4.54	8	5	ND	1	34	1	2	2	85	.50	.090	2	147	3.23	19	.10	3	3.00	.01	.06	1	4
89-5 45-55	3	69	20	10	2.4	11	17	57	4.22	17	5	ND	1	16	1	3	2	9	.15	.077	2	8	.15	29	.01	2	.78	.04	.13	1	10
89-5 55-65	1	155	573	306	1.3	41	25	2115	5.35	3	5	ND	1	9	2	2	2	109	.32	.098	2	181	4.76	10	.05	2	4.64	.01	.04	1	1
89-5 65-75	2	201	25	258	.8	9	15	648	4.71	16	5	ND	1	12	1	2	2	29	.15	.088	4	6	1.96	24	.01	3	2.47	.03	.12	1	37
STD C/AU-R	19	62	40	133	7.4	73	31	1038	4.11	42	21	8	39	52	20	16	23	64	.45	.090	42	55	.93	181	.08	38	2.05	.06	.14	13	530