PARTO OF SHEET

### FLUIDS PROGRAM - 3/7-2



FLUIDS AND CEMENT

 $T_{ t G}$ . COLENC

- 1. 36" PHASE interval sea bed 85 to 145 m. 30" casing at  $\pm$  140 m.
  - 1.1. Drill with sea water, allowing returns to go to the sea Displace sea water with 6-8 m³ gelly slugs of mud before each connection. When the 36" hole is drilled, pump to fill up well with thick mud before making a wiper trip.

Criculation with viscous mud into the hole for cleaning before pulling out to run 30" casing.

To avoid any incident, fill the hole and circulate. It is necessary to get permanently in reserve a volume of thick mud corresponding to the hole volume.

1.2. Composition of high viscosity mud for starting up

Sea water : 1000

CMC extra HV : 2 - 5 Kg to adjust rheology and reduce

filtrate

Caustic soda : 3 - 6 Kg

Wyoming bentonite: 80 - 120 Kg if possible prehydrated in

300 1 FW

1.3. Characteristics of mud

Weight : 1.04 - 1.06

f.viscosity : 140 - 160

filtrate : 10 - 15

pH : 8.5 - 9

1.4. Provide mud 250 m<sup>3</sup>

- 2. 26" PHASE interval 145 to 760 m. 20" casing at + 750 m.
  - 2.1. Drill out cement and shoe with sea water and returns to the sea. After drilling the shoe, displace sea water by mud and continue drilling 17 1/2 hole to 760 m. After logging opening 26" hole.

Before running in the 20"casing, the hole control with 26" bit with no mud return, requires an important volume of mud. To avoid any incident, fill the hole and circulate. It is imperative to get permanently in reserve, a volume of mud corresponding.

2.2. Composition of bentonite and lignosulfonate mud

Sea water : 1000 1

CMC HV or LV : 4 - 6 Kg to adjust rheology reduce filtrate

FCL : 6 - 8 Kg Caustic soda : 5 - 7 Kg w.bentonite : 50 - 70Kg

d.detergent : 1 - 2 1 if "bit balling" or torque

2.3. Characteristics of mud

Weight : 1.10 - 1.15

Solids : 7 - 11

Funnel viscosity: 65 - 70

Plastic viscosity

or "N" : 15 - 20 / 0.55 - 0.60

yield point or "K": 10 - 15 / 1.30 - 1.60

gel 10 sec : 2 - 5

gel 10 min . 10 - 15

filtrate 30 min: 5 - 8

pH : 9 - 9.5

alkalinity Pf : 0.1 - 0.3

2.4. Provide mud 1100 m<sup>3</sup>

4.

- 17 1/2 PHASE interval 760 to 2560 m. 13 3/8 casing at ± 2550m.
  - 3.1. Drilling out with SW the cement inside 20" casing, or pretreat the system with sodium bicarbonate to avoid cement contamination and the products to adjust the following composition.
  - 3.2. Composition of gypsum surfactant system

Sea water

: 1000 1

CMC extra. HVouLV: 5 - 10 Κq

to adjust rheology reduce

filtrate

FCL

: 15 - 20 Kg

Surfactant : 10 - 15 control the normal tendency

of shales to ball up

Caustic soda

: 8 - 12 Κg

: 10 - 15 Kg to adjust ca ++

W.bentonite

: 40 - 60 Κq

cake and filtrate

barite

Gypsum

: for weight adjustment

Kg

### 3.3. Characteristics of mud

Weight

: 1.15 grading to 1.30

Solids

9 - 11 :

13.16

funnel viscosity: 55 - 60

plastic viscosity

or "N"

: 30 - 35 / 0.60 - 0.65

yield point or "K" : 25 - 22 / 0.80 - 0.85

gel 10 see

: 2 - 4

gel 10 min

: 10 - 15

filtrate 30 min : 4 - 6

pН

:9.5 - 10

alkalinity Pf

: 1 - 1.5

Ga<sup>++</sup> (filtrate) :

1 - 1.5

3.4. Provide mud

1500 m<sup>3</sup>

### 4. GENERAL DISCUSSION

It is during this stage that the "SHALES" are to be drilled. These shales are reputed as being difficult due to problems arising during penetration. They hydrate easily and eventually come apart and, if the mud is not perfectly suited to this type of formation, this may give rise to redrilling and caving.

The mud weight will be progressively increased:

20" shoe 1.10 - 1.15: 1000 m 1.16 1200 1.18 1400 1.20 1600 1.22 1800 1.24 2000 1.26 to : 1.28 - 1.30 2560

The mud used in this stage must therefore respond to the following essential criteria:

- have a specific gravity at such a level as to contain the slight tendency to creep in the shales. The specific gravity will be obtained in successive from a level of 1.15 at the start of this phase to 1.26 towards 2000m.
- possess normal rheological properties to have a good dispersed system, to ensure proper cleaning and to avoid a too high equivalent circulating density.
- have a very low filtrate.
- The solids will be eliminated on the shales shakers, desilters, desanders, mud cleaners and dilution. Run the smallest screens possible on all shakers and mud cleaners. The shakers will be constantly watched. If there is a decrease of cuttings volume, one short trip to shoe will be made immediately and circulation established.
- be in good electrochemical equilibrium with the shales to avoid hydratation phenomena. pH 9.5 - 10.
- In order to maintain this mud in a dispersed state and in good electrochemical equilibrium with the formation, constant supervision is required, together with continuous treatment with lignosulfonate, surfactant and gypsum, as well as applying all mechanical means for eliminating solids.

- 5. <u>12 1/4 PHASE</u> interval 2560 to 3260m. 9 5/8 casing at ± 3250m.
  - 5.1. Before drilling out the cement inside 13 3/8 casing, mud weight will be increased to 1.50 and pretreat the system with sodium bicarbonate to avoid cement contamination and the products to adjust the following composition.
  - 5.2. Composition of ferro chrome lignosulfonate and chrome lignite mud

Sea water : 1000 1

CMC pure HV or LV: 4 - 8 to adjust rheology and

reduce filtrate

FCL : 25 - 35 Kg

CL : 25 - 35 Kg maintain concentration

caustic soda : 8 - 12 Kg

w.bentonite : 30 - 50 Kg necessary to have a good cake

d.detergent : 1 - 2 1 more if "bit balling" or torque

barite : for weight adjustment

5.3. Characteristics to mud

Weight : 1.50 grading to 1.60

Solids : 18 - 21 21 - 24

funnel viscosity : 50 - 60

plastic viscosity

or "N" : 27 - 34 / 0.65 - 0.70

yield point or "K": 10 - 13 / 0.70 - 0.80

gel 10 sec. : 1 - 2

gel 10 min : 10 - 15

filtrate 30 min : 3 - 5

filtrate HT/HP : 10 - 12

pH :9.5 - 10

alkalinity Pf :0.3 - 0.5

- 5.4. Provide mud : 500 m<sup>3</sup>
- 5.5. The mud weight will be progressively increased:

13 3/8 shoe 1.50

2700 m 1.52

2900 1.54

3100 1.56

3200 1.58

.3250 1.60

- 8 1/2 PHASE interval 3200 to 4500 m. 7"liner if needed.
  - 6.1. Before drilling out the cement inside 9 5/8 casing dump and clean the mud pits, and made satured salt mud. Pretreat if necessary, this mud with sodium bicarbonate to avoid cement contramination and displace lignosulfonate mud by saturated salt mud.

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6.2. Composition of saturated salt mud

Sea water

1000

Salt

350 Κq

attapulgite clay: 20 - 30 Kg

starch

: 20 - 30 Kg

caustic soda

2 - 4 Kg (after drilling out cement)

barite

: for weight adjustment

CMC R110 : will be added after 4000 m if starch is

deficiency w/TO

### 6.3. Characteristics of mud

Weight

1.60 grading to 1.80

Solids

: 21 - 24

27 - 30

funnel viscosity: 50 - 55

plastic viscosity

or "N"

: 45 - 50 / 0.70 - 0.80

yield point or "K": 18 - 22 / 0.80 - 0.90

gel 10 sec.

2 - 4

gel 10 min.

: 10 - 15

filtrate 30 min

3 - 4

filtrate HT/HP

: 10 - 12

Щq

:8.5 - 9

alkalinity Pf

 $0 - 0^{+}$ 

Na Cl

: maintain maximum saturation

6.4. Provide mud

500 m<sup>3</sup>

### 6.5. The mud weight will be progressively increased:

9 5/8 shoe

1.60

3300 m

1.63

3400

1.66

3500

1.70

3600

1.72

3800

1.76

4000

1.80

to

to

TD

1.80

### 7. PERMANENT INSTRUCTIONS

7.1. To face any problem it is necessary to keep one mud pit full with heavy mud as safety, that is to say:

## ONE TANK OF MUD W/A S.G. OF 2.10 FOR THE 12 1/4 AND 8 1/2 PHASES

and necessary products needed to make this volume quickly, ready to be used immediately to control any income or blow out.

- For an immediate action, take care to test as soon as possible all the installations and facilities enabling a quick controle of any income.
- Further in the sequence of the precautions to take care, be sure to be able to make and circulate quickly a barite plug and a cement plug.
- Permanent survey of the pit level, volumes and of the characteristics of the fluid in circulation, and mainly, if a velocity break occurs. The mud ditch fluid, gas content, temperature, chloride variation, pH, should be under survey, in order to detect any show in cases of abnormal pressure, H<sup>2</sup>S etc...
- Let an homogenous and a steady fluid in well before every pulling out of hole.
- Control the permanent filling of the well, particularly during the trips (possum belly tank)
- Calibrate the densimeter often mainly for the abnormal pressured formation.
- A pressurized mud balance will be used throughtout the 12 1/4 and 8 1/2 phases to ensure accurate mud weight determination.
- For the purpose of making easier the electrical logging, the running in casings and to assume their best cementing jobs easier, it is necessary to be very careful to have the best rheological fluid characteristics during the circulation preceding these operations.
- Mud tanks for mixing water with additives will be throughly cleaned to prevent pollution by mud.

## HYDRAULIC RECOMMENDATIONS

OPTIMUM FLOW RATE	maxi	maxí maxi	3600 3900 3900 3900 4000	2000 2200 2100 2200	1200 1200 1100
FLOW RATE FOR N Re 1300		7406	3839 4097 6773 5322 3999	1612 2242 2178 2259	1811 1350 1240
MINIMAL FLOW RATE PULLING		5430	3463 3838 3841 3871 3991	1919 1821 1778 1789	738 720 742
CRITICALLARGEST FLOW ANNULUS RATE CASING MINUS D.PIPE		28 25 5	18.7	12.34	8.53
CRITICAI FLOW RATE		12313 21559	6380 6809 11262 8845 6639	2676 3726 3014 3754	3010 2242 2057
HOLE DIAMETER MINUS D.C DIAMETER	36 9.50	17.50 9.50 26 9.50	17.50 9.50	12.25 9.50	8.50 6.50
YP Or "N"		0.55	0.66 0.65 0.58 0.64 0.70	0.70 0.66 17 0.64	0,66 0.67 0.73
PV Or "K"		1.90	0.54 0.65 1.79 1.08 0.61	0.46 0.89 37 1.02	1.18 0.76 0.48
B or P		<u>Ф</u> Ф	<u>ዋዋዋዋ</u>	ው ው ወ ው	ወይል
SG MUD	1.04	1.10	1.15 1.22 1.33 1.40	1.40 1.42 1.45 1.47	1.19
VELOCITY m/min		10	15 18 20 20 20	25 25 25 25	25 25 25
CUTTING THICK X Ø		3 X 5	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	ოოოო * * * * ა ა ა ა ა	223 *** ***
рертн	150	400	900 1580 2044 2300 2486	2510 2610 2792 2892	3000 3300 3500

### 7.2. Permanent safety materials in stock - barite on the rig 200 T - barite on each supply boat (2) 100 T - w. bentonite 20 T - cement 40 T - attapulgite clay (sack) 5 T 1.5 m<sup>3</sup>- calcium chloride liquid - L.C.M. fiber 0.5 T granular (coarse and fine) 1 flake 0.5 T m<sup>3</sup> - pipe free 2 - sodium bicarbonate 1 $\mathbf{T}$ - soda ash 1 Т - sodium tetraphosphate pH 10 0.5 T - H2S scavenger 1 т

### BARITE IN STOCK THE 01.10.80

COMPANIES	BULK DUSAVIK	BULK TANANGER	SACKS STORAGE	TOTAL
ANCHOR DRIL.	370	376	55	801
MILCHEM	550	270	1100	1920
CECA	507	922		1429
DRESSER	301	171	230	702
IMCO	227	994	647	1868
NORSK PETR.	538	232		770
TOTAL m/tons	2493	2965	2032	7490

### MATERIALS QUANTITIES ESTIMATION

PHASE	MATERIALS	QUANTITIES (Kg or 1)
36"	CMC extra HV	1500
± 250 m <sup>3</sup>	caustic soda	1500
	w. bentonite	30000
	attapulgite clay(sack)	5000
	defoamer	250
	barite	50000
	lost circulation	2222
	material sodium bicarbonate	2000
	1	1000
	soda ash pipe free	1000 2000

PHASE	MATERIALS	QANTITIES (Kg or 1)
26"	CMC extra HV	3500
+ 1100 m <sup>3</sup>	CMC LV	3500
	FCL	9000
	caustic soda	8000
	w. bentonite	70000
	defoamer	500
	drilling detergent	1000
	barite	100000
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PHASE	MATERIALS	QUANTITIES (Kgorl)
17 1/2	CMC extra HV	7000
± 1500 m <sup>3</sup>	CMC LV	8000
	FCL	30000
	surfactant	23000
	caustic soda	18000
	gypsum	23000
	defoamer	1200
	barite	200000
	sodium tetraphospate	500
	H2s scavenger	1000
	drilling detergent	800

12 1/4 ± 500 m <sup>3</sup> CMC pure HV CMC pure LV 3000 FCL 18000 CL caustic soda w. bentonite defoamer barite barite sodium bicarbonate  1000 1000 1000 1000 1000 1000 1000 1	PHASE	MATERIALS	QANTITIES (Kg or 1)
FCL 18000 CL 18000 caustic soda 6000 w. bentonite 25000 drilling mud detergent 600 defoamer 400 barite 450000	12 1/4	CMC pure HV	1000
CL 18000 caustic soda 6000 w. bentonite 25000 drilling mud detergent 600 defoamer 400 barite 450000	± 500 m <sup>3</sup>	CMC pure LV	3000
caustic soda 6000  w. bentonite 25000  drilling mud detergent 600  defoamer 400  barite 450000		FCL	18000
w. bentonite 25000 drilling mud detergent 600 defoamer 400 barite 450000		CL	18000
drilling mud detergent 600 defoamer 400 barite 450000		caustic soda	6000
defoamer         400           barite         450000		w. bentonite	25000
barite 450000		drilling mud detergent	600
1		defoamer	400
sodium bicarbonate 500		barite	450000
		sodium bicarbonate	500

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### salt   180000   15

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CEMENTING AND DISPLACEMENT
CALCULATIONS

CEMENTING PROGRAM 3/7-2

FLUIDS AND CEMENT

COLENO

# CEMENTING PROGRAM

HOLE SIZE CASING LINER	WEIGHT lbs/ft	DV SHOE	CLASS	SG	INTERVAL	MUD	MAXIMUM EXPECTED PRESSURE DURING DISPLACEMENT KG/cm2	MAXIMUM ESTIMATED EQUIVALENT DENSITY DURING DISPLACEMENT
36" 30"	310	140	១	1.90	140 to 85	1.05	17	1.46
26"	133	750	LIGHT G	1.50	650 - 85 750 - 650	1.15	104	1.54
17 1/2 second . 13 3/8 stage	7.2	700	LIGHT G G	1.50	600 - 400 700 - 600			
17 1/2 first 13 3/8 stage	7 /	2550	LIGHT G	1.50	2400 - 2000 2550 - 2400	1.30		
12 1/4 9 5/8	53,5	32\$0	LIGHT G G	1.50	3100 - 2450 3280 - 3100	1.60		
8 3/8	35	4500	É	2.10	If needed	1.90		

### DEPTH RECALL

WATER DEPTH 60 m RKB SEA LEVEL 25 m RKB SEA BOTTOM 85 m

### ADDITIVES FOR CEMENT SLURRIES

### - PREHYDRATED BENTONITE

Contact time with fresh water must be sifficient to let Wyoming bentonite to be hydrated. It is generally know that this one is complete after an important mixing of about one hour.

### - CALCIUM CHLORIDE

The effect of the calcium chloride on the viscosities of slurries is without influence. The calcium chloride must be added into prehydrated bentonite just before to mixt the cement slurry.

### ADDITIVES FOR CEMENT WITH LIQUID ADDITIVES SYSTEM

Extender is incompatible with  ${\rm CaCl}_2$ , the extender mixing in water with  ${\rm CaCl}_2$  causes massive precipitation of a white insoluble residue.

FLUID LOSS REDUCER MUST BE COMPULSORY ADDED AND MIXED WITH WATER BEFORE ADDING DISPERSANT AND RETARDER.

### CLASS G AND E CEMENT

The rapid knowledge on the rig of the cement quality  ${\tt G}$  and  ${\tt E}$  will be obligatory before each operation.

Founded on two easy measures of specific gravity on rheology, it will lead to check if the cement has not been contaminated during its transport.

After been prepared the current should have their characteristics available in the hereby intervals:

MEASURES	G RATION WATER/CEMENT 44%	E RATIO WATER/CEMENT 38%
SG	1.90 - 1.92	1.96 - 1.98
600	90 - 1.20	95 - 125
300	60 - 90	50 - 80
200	50 - 75	35 - 60
100	40 - 60	20 - 45
INITIAL GEL	10 - 20	5 - 15
10 GEL	<b>&lt;</b> 100	<b>&lt;</b> 100
		·

30" CASING shoe at about	140
Hole gage 36" 656,80 Casing gage 30" (3101b)455,80/28"407.80	
Top cement up to sea bed Open hole interval	85 55
Excess in open hole 200% (hole size ± 46")	

### SLURRY VOLUME

Annula	ır 36"	X	30"	201.00	X	55	(3)	=	33165
Inside	<b>:</b>		30"	407.80	X	5		=	2039
Total	volume	G	ceme	nt				=	35204

### TIMING OF OPERATION

Mix and pump G cement 35204:700 =	51
Drop plug	20
Displacement 5" DP $\pm$ 9.27 x (140-5) = 1251	
Flow rate 700 1/min	2
Total	73
Safety factor 50%	
GRAND TOTAL min	110

### 30" SLURRY

Composition	Weight-Kg/100	Volume-1/100	$\underline{\text{By m}^3 \text{ FW}}$	For each 10 bbls
Fresh water Defoamer Accelerator G cement SG = 1.89	0.050	42 0.050 4.439 31.75 78.239	1.19 1 105.69 1	0.50 gal 44.37 "

RESULTS (\* 140m BHST 5°C - mud weight 1.05)

T.T. requested:  $2^{\text{H}}00$  T.T. recorded :  $6^{\text{H}}00$ 

24 hours compressive strenght: 94 b
rheology and gels: 175 - 138 - 121 - 102/49-114
Maximum expected pressure during displacement: 16.80 b
Maximum estimated equivalent density : 1.46

Fresh water	42 x 44995 100	18898	1
Defoamer	$\frac{0.050 \times 44995}{100}$	22.5	1
Accelerator D77	4.439 x 44995 100	1997.3	1
G cement	$\frac{100}{78.24} \times 35204$	4499.5	Kg

20" CASING shoe at about		750
Casing gage 30" (310 lb) Hoel gage 26" Casing gage 20" (133 lb)	455,80/407.80 342,50 202,96/ 18.72" 177,76	140
Top cement up to sea bed Open hole interval Casing interval Excess in open hole 125%		85 610 55

### SLURRIES VOLUMES

Lead-light 30" x 20"	204,84 x	55	=	11.266
Lead-light	139,54 x	510(2.25)	=	160.122
TOTAL VOLUME LIGHT			=	171.388
Tail - G	139,54 x	100(2.25)	=	31.396
Inside 20"	177.76 x	12	=	2.133
TOTAL VOLUME G			=	33.529

### TIMING OF OPERATION

Mix and pump light Mix and pump G Drop top plug	171388 : 900 = 19 33529 : 700 = 49	9 49
Displacement 20" 177,76  HW 4,61  Total displacement ±  Flow rate 2000 1/min	_	J 20
Total displacement <sup>±</sup> Flow rate 2000 1/min	116469 5	9 59
Total Safety factor 50%	319	9 128 64
GRAND TOTAL (min.)	478	3 192

### 20 " LEAD SLURRY

Composition	Weight-Kg/100	Volume-1/100	By m <sup>3</sup> SW	For each 100 bbls
Sea water Defoamer Extender D75 G cement SG - 1.50	109.60 0.05 6.30 100	107 0.05 4.50 31.75 143.30	0.47 l 42.05 l	0.20 gal 17.65 gal

RESULTS ( $^{\pm}$  750 m BHST 25 $^{\circ}$ C - mud weight 1.15)

T.T. requested: 8h00 T.T. recorded: 8h30 24 hours compressive strength: 35 b Rheology and gels: 50-39-35-30/18-39

Sea water	$\frac{107  x  119601}{100} = 127973$	1
Defoamer	$\frac{0.05 \times 119601}{100} = 60$	1
Extender D75	$\frac{4,50 \times 119601}{100} = 5382$	1
G cement	$\frac{100 \times 171388}{143,30} = 119601$	Kg

### 20" TAIL SLURRY

Composition	Weight-Kg/100	Volume-1/100	By m <sup>3</sup> FW	For each 10 bbls
Fresh water Defoamer G cement SG + 1.90	44 0.05 100	40 0.05 31.75 75.80	1.14 1	0.47 gal

RESULTS ( $^{\pm}$  750 BHST 25 $^{\circ}$ C - mud weight 1.15)

T.T. requested:  $\Rightarrow$  3<sup>h</sup>15 7.T. recorded: 7<sup>h</sup>00

24 hrs. compressive strenght: 122 b
Rheology and gels: 133-92-79-60/14-21
Maximum expected pressure during displacement: 104 b
Maximum estimated equivalent density : 1.54

Fresh water	$\frac{44}{100} \times \frac{44233}{100} =$	19462	1
Defoamer .	$\frac{0.05 \times 44233}{100} =$	22	1
G cement	$\frac{100  \text{x}  33529}{75.80} =$	44233	Kg

13 3/8 CASING Shoe at about	2550
Casing gage 20" (133 lb) 202,96/18,72" 177,76 DV at about	750 700
Hole gage 17 1/2 155.20	
Casing gage 13 3/8 (72 lb)90.80/12.34" 77,24	
Top cement first stage	2000
Top cement second stage	400
Open hore interval first stage	550
Casing interval second stage	300
The cement volumes are calculated according to:	

- the theoretical volume increased by 58% (hole size ± 19 1/2") for the preliminary calculations.
- and the volume given by the caliper increased by 10% for the final calculations.

### SLURRIES VOLUMES

### Fisrt stage

Lead light	17	1/2	x	13	3/8	64,40	x	400	(1,58)	=	40700
Tail G	17	1/2	x	13	3/8	64,40	x	150	(1,58)	=	15262
Inside G				13	3/8	77,24	x	24		=	1854
Total G										==	17116

### Second stage

Lead	light	20"	х	13	3/8	86,96	x	200	=	17392
Tail	G	20"	x	13	3/8	86.96	x	100	=	8696

### <u>TIMING OF OPERATION</u> (first stage)

Pump the spacer (at least 100 m of annulus) SG. Rheological and gels slightly greater than the mud one.

Mix and pur Mix and pur Drop top pur Displacement	mp G luq	17116	:	700	=	46 25 20	25 20
Flow rate	2500 l/mi 1600 l/mi				125000 70108		50 45
TOTAL Safety fac	tor 50%					186 93	140 70
GRAND TOTAL	<u>ւ</u>					279	210

Before cementing the second stage, circulate w/mud through DV necessary time for gelation of the first stage slurry (about tail slurry T.T. requested).

### TIMING OF OPERATION (second stage)

Pump the spacer (at least 100 m of annulus) SG. Rheological and gels slightly greater than the mud one.

·				
Mix and pump light Mix and pump G Drop top plug Displacement 13 3/8 ±	8696 : 700	=	20 13 20	13 20
Flow rate 2500 l/min 1600 l/min		21000 33068		9 22
TOTAL Safety factor 50%			84 42	64 32
GRAND TOTAL		1	.26	96

### 13 3/8 LEAD SLURRY (first stage)

Composition	Weight Kg/100	Volume 1/100	By m <sup>3</sup> FV	Ā	for each 10 bbls
Fresh water Defoamer Extender D75 G cement SG ± 1.50	102 0.05 7 100	102 0.05 5 31.75 138.80	0.49 49	1	0.20 gal 20.58 gal

RESULTS ( $\pm$  2550 m BHST 75 $^{\circ}$ C - mud weight 1.30)

T.T. requested:  $> 5^h00$ T.T. recorded:  $5^h15$ 24 hrs. compressive strenght: 31 b Rheology and gels: 56-36-29-20/10-29

Spacer Fresh water	± 5 102	x 6500 x 29323	=	33 29912	Kg l
Defoamer	0.0	100 100	**	15	1
Extender D75	5	x 29323 100	- =	1466	1
G cement	100	<u> </u>	- =	29323	Kg

### 13 3/8 TAIL SLURRY (first stage)

Composition	Weight-Kg/100	Volume-1/100	By m <sup>3</sup> FW	For each 10 bbls
Fresh water	42	42		
Defoamer	0.05	0.05	1.19 1	0.5 gal
F.loss reduce	D73 1.398	1.331	31.69 1	13.3 gal
Dispersant D8	0 0.546	0.444	10.57 1	4.4 gal
Retarder D81	0.446	0.355	8.45 1	3.5 gal
G cement	100	31.75		
SG <sup>±</sup> 1.90		75.93		

### RESULTS ( $^{\pm}$ 2550 m BHST 75 $^{\circ}$ C - Mud weight 1.30)

T.T. requested: >
T.T. recorded:

24 hrs. compressive strength: 262 b

Fluid loss: 92

Rheology and gels: 62-31-20-10/2-12
Maximum expected pressure during displacement: 351.50b Maximum estimated equivalent density 1.38

Fresh water	42 x 100	22542 =	9468	1
Defoamer	0.05 x 100	22542 =	12	1
F.loss reduce D73	1,331 x 100	22542 =	300	1
Dispersant D80	0.444 x 100	22542 =	100	1
Retarder D81	0.355 x 100	22542 =	80	1
G. cement	100 x 75.93	<u>17116</u> =	22542	Kg

### 13 3/8 LEAD SLURRY (second stage)

Composition	Weight-Kg/100	Volume-1/100	By m <sup>3</sup> SW	For each 10 bbls
Sea water Defoamer Extender D75 G cement SG ± 1.50	109.6 0.05 6.30 100	107 0.05 4.50 31.75 143.30	0.47 1 42 1	0.20 gal 17.65 gal

RESULTS ( $^{\pm}$  816m - BHST 30 $^{\circ}$ C - Mud weight 1.25)

T.T. requested:  $\geq 2^{h}_{h}00$ T.T. recorded:  $10^{h}$ 

24 hrs. compressive strenght: 31 b Rheology and gels: 46-35-31-21/14-20

Spacer 
$$\pm 5$$
 x 8696 = 44 Kg  
Sea water  $\frac{107}{100}$  x  $\frac{12137}{100}$  = 12987 1  
Defoamer  $\frac{0.05 \times 12137}{100}$  = 6 1  
Extender D75  $\frac{4.5 \times 12137}{100}$  = 546 1  
G cement  $\frac{100 \times 17392}{143.30}$  = 12137 Kg

### 13 3/8 TAIL SLURRY (second stage)

Composition	Weight-Kg/100	<u>Volume-1/100</u>	By m <sup>3</sup> FW	For each 10 bbls
Fresh water Defoamer	44	44 0.05	1.14 1	0.5 gal
G. cement SG <sup>±</sup> 1.90	100	31.75 75.80		

RESULTS (816 m - BHST 30°C - mud weight 1.25)

T.T. requested:  $\rightarrow$   $_{6}^{h_{00}}$ 

24 hrs. compressive strenght: 101 b
Rheology and gels: 127-87-70-53/18-22
Maximum expected pressure during displacement:
Maximum estimated equivalent density: 123 b

Fresh water 
$$\frac{44 \times 11472}{100} = 5048 1$$
Defoamer 
$$\frac{0.05 \times 11472}{100} = 61$$
G cement 
$$\frac{100 \times 8696}{75.8} = 11472 \text{ Kg}$$

9 5/8 CASING Shoe at about	የ) 3 <b>28</b> 0
Casing gage 13 3/8 72 lbs 90.80/12.34" 77.24 Hole gage 12 1/4 76.04	2550
Casing gage 9 5/8 53.5 " 47.10/ 8.53" 37.92	
Top cement 100 m above the 13 3/8 shoe	2450
Open hole interval	700
Casing interval	100

The cement volumes are calculated according to:

- The theoretical volume increased by 56% (hole size  $\pm$  13 1/2) for the preliminary calculations.
- for the preliminary calculations.

   And the volume given by the caliper increased by 10% for the final calculations.

### SLURRIES VOLUMES

Lead-light 13 3/8 x 9 5/8	30.14	x 100	=	3014
Lead-light 12 1/4 x 9 5/8	27.94	x 550	(1.56) =	24831
TOTAL VOLUME LIGHT			=	27855
Tail G	28.94	x 150	(1.56) =	6772
Inside 9 5/8	37.92	x 36	=	1365
TOTAL VOLUME G			=	8137

# TIMING OF OPERATION

Pump the spacer (at least 200 m of annulus) SG, rheology and gels slightly greater than the mud one.

Mix and pump light cement Mix and pump G " Drop top plug Displacement ± 37.92 x 3250	8137	:	32 13 20	13 20
Flow rate 1800 l/min 700 l/min maxi		700 487	49 51	49 51
Total			165	133
Safety factor 50%			83 248	67 200
GRAND TOTAL (min.)			240	200

# 9 5/8 LEAD SLURRY

Composition We	eight-Kg/100	Volume-1/100	By m <sup>3</sup> FW	For each 10 bbls
Fresh water	101	101		
Defoamer	0.05	0.05	0.50 1	0.20 gal
Extender D75	7	5	49.50 1	20.78 gal
F.loss reduce I	73 1.398	1.331	13.18 1	5.53 gal
Dispersant D80	0.546	0.444	4.40 1	1.85 gal
Retarder D81	0.223	0.177	1.75 1	0.74 gal
G cement	100	31.75		•
SG ± 1.50		139.75		

RESULTS ( $\pm$  32\$0 m BHST 100°C - Mud weight 1.60)

T.T. requested:  $\Rightarrow$  4 $^{h}_{h}$ 00 T.T. recorded : 5 $^{h}$ 30

24 hrs. compressive strength: 50 b

Fluid loss: 520

Rheology and gels: 37-26-22-18/17-24

### REQUIREMENTS

$$\frac{\pm 5}{1000} \times \frac{5800}{1000} = 29 \text{ Kg}$$
Fresh water 
$$\frac{101}{1000} \times \frac{19932}{1000} = 20131 \text{ 1}$$
Defoamer 
$$\frac{0.05 \times 19932}{1000} = 10 \text{ 1}$$
Extender D75 
$$\frac{5 \times 19932}{1000} = 997 \text{ 1}$$
F.loss reduce D73 
$$\frac{1.331 \times 19932}{1000} = 265 \text{ 1}$$
Displacement D80 
$$\frac{0.444 \times 19932}{1000} = 88.5 \text{ 1}$$
Retarder D81 
$$\frac{0.177 \times 19932}{100} = 35 \text{ 1}$$
G cement 
$$\frac{100 \times 27855}{1000} = 19932 \text{ Kg}$$

# 9 5/8 TAIL SLURRY

Composition Weigh	ht-Kg/100	Volume-1/100	By m FW	For each 10 bbls
Fresh water Defoamer F.loss reduce D73 Dispersant D80 Retarder D81 G cement SG ± 1.90	41 0.05 1.864 1.095 0.671	41 0.05 1.775 0.887 0.532 31.75 75.99	1.220 1 43.293 1 21.634 1 12.976 1	9 gal

RESULTS ( $^{\pm}$  3250 m BHST 100 $^{\circ}$ C - Mud weight 1.60)

T.T. requested: 3<sup>h</sup>30
T.T. recorded: 4<sup>h</sup>00

24 hrs. compressive strength: 250 b

Fluid loss: 19

Rheology and gels: 70-32-22-11/2-38

Maximum expected pressure during displacement: 530 b
Maximum estimated equivalent density : 1.63

# **REQUIREMENTS**

Fresh water	$\frac{41}{100} \times \frac{10707}{} =$	4390	1
Defoamer	$\frac{0.05 \times 10707}{100} =$	6	1
F.loss reduce D73	$\frac{1.775 \times 10707}{100} =$	190	1
Dispersant D80	$\frac{0.887 \times 10707}{100} =$	95	1
Retarder D81	$\frac{0.532 \times 10707}{100} =$	57	1
G cement	$\frac{100}{76}$ $\times$ 8137 =	10707	Kg

#### 7" LINER (if required) shoe at about 4300 Landing collar 4240 Casing gage 9 5/8 (53.5 lb) 47.10/ 8.53" 37.92 3250 8 3/8 Hole gage 36.61 7" 24.88/ 6 " Liner gage (35 lb) 18.27 5" 13.16/ 4.27" Drill pipe (19.5 lb) 9.16 Top liner 3150 Open hole interval 1050 Between casing-liner 100

- The cement volumes are calculated according to:
   The theoretical volume increased by 78% (hole si
- The theoretical volume increased by 78% (hole size ± 9 1/2) plus a 50% wash out factor for the preliminary calculations.
- Practically the final volumes calculations shall be based on the caliper log increased by 10% plus a 50% wash out factor (slurry SG 1.95).

### SLURRY VOLUME

Annular 9 $5/8 \times 7$			= 1304
Annular 8 3/8 x 7	10.66	$\times$ 1050(1.78)	<b>=</b> 19923
Wash out volume 50%		•	9962
Left in 7"	18.27	x 60	= 1095
Total volume E cement			32285

# TIMING OF OPERATION

Circulating the necessary time to obtain a homogeneous system and the best rheological and thixotropic characteristics of the mud. Pump the lead spacer (at least 200 m of annulus) SG, rheological and gels slightly greater than the mud one.

Mix and pump E cement 32285 : 600 = Drop pump down plug		55 20
In order to avoid contact between the mud in drill pipe and the cement above the stinger the displacement, it will be preferable to phind the plug a volume of tail spacer which the liner at least 200 m above the running tailon.	during oump be- will fill	
First bumping pressure 5" 9.16 x 3155 = Final bumping pressure 7" $18.27 \times 1080 = 70$		
· .	12000 11000 25632	12 38 52
Energizing work Inverse circulating 9.16 x 3155 (+ 10962) 1000		30 40
Total Safety factor 50% GRAND TOTAL 6 <sup>n</sup> 18		247 124 371

# 7" SLURRY

Composition	Weight-kg/100	<u>Volume-1/100</u>	weight by m3SW
salt water	45	38	
defoamer	0.05	0.05	0.5 1
f.loss reduce D 59	0.75	0.75	19.7 kg
dispersant D 45	0.20	0.15	5.3 kg
E cement	100	31.75	-
SG ± 2.06		70.7	
SG = 1.95 (for wash	out volume)		

RESULTS:  $\pm (4300 \text{ m-BHST } 140^{\circ} \text{ C} - \text{mud weight } 1.90)$ 

T.T. requested: 6:00 hrs.

T.T. recorded:

24 hrs. compressive strength:

Fluid loss:

Rheology and gels:

Maximum expected pressure during displacement: 836 b
Maximum estimated equivalent density : 1.94

# REQUIREMENTS

	±. <sub>5</sub>		27:00			
Spacer		x 1000	<u>2700</u>	=	14	kg
Fresh water	32.5	x 100	<u>45664</u>	=	14841	1
Salt	279	<u>x</u> 1000	14841	=	4141	kg
F. loss reduce D59	0.75	x 100	<u>45664</u>	=	344	kg
dispersant D45	0.20	х	45664	=	92	
E cement	100	x 70	32285 ).7	=	45664	kg

CASING CALCULATIONS

Maximal pressure at the weak point (casing shoe) Ps max is reached when gas comes to this point. To have max lower than p frac, we need:

P pore + 
$$S \leq P$$
 frac +  $\frac{\Delta z - hs}{10}$  d +  $\frac{hs}{10}$  fm (1)

Considering gas is perfect  $(\frac{PV}{ZT} = constante)$ 

(P pore + S) 
$$h_b$$
 rk = P frac x hs (2)

(1) + (2) give:

hb 
$$\langle \frac{10 \times P \text{ frac}}{(d-fm) (P \text{ pore+S})} \rangle$$
 P frac +  $\frac{\Delta Z}{10} d - P \text{ pore } - S \frac{1}{rk}$ 

# Symbols:

S: safety factor over pressure when circulating bubble

↑Z : height between bottom and casing shoe

hs : height of gas bubble at shoe

hb : height of gas bubble at bottom

d : mud specific gravity

fm : average gas density

r : ratio linear volum at bottom/linear volum at shoe

 $k : \frac{Zs \times Ts}{Zb \times Tb}$  with T = absolute temperature Z = compressibility gas factor

	A	В	С
Casing set	20"	13 3/8	9 5/8
Drilling in	17 1/2	12 1/4	8,1/2
P fracturation	at 750m: 125	g at 2550m:450kg	at 3230m 675kg
d (mud)	at 2560m: 1.3	at 32 <b>(</b> 0m: 1.60	at 4500m: 1.80
f (gas)	at 2560m: 0.2	at 32 <b>6</b> 0m: 0.30	at 4500m: 0.35
P pore	at 2560m: 3466	at 32 0m:522kg	at 4500m:810kg
S	10 kg	10 kg	10 kg
∆Z	1810 m	710 m	1250 m
K	0.95	1.0	1.0
r	0.766	0.683	0.625
<del> </del>			<u> </u>

- Nota: P frac is higher than p max used for leak off test.
  - While circulating, we keep an overpressure on bottom around 10kg/cm<sup>2</sup>. (safety factor S)
  - K = 1. We admit that T and Z remain constant between bottom and shoe except for case A where we come near the surface.

#### Calculation of r:

- case A: At bottom 17 1/2 hole, 9 1/2 DC at shoe 17 1/2 hole, 5" drill pipes  $r = \frac{109. \ 1/m}{142.5 \ 1/m} = 0.766$
- case B: at bottom 12 1/4 hole, 8" DC at shoe 12 1/4 hole, 5" drill pipes  $r = \frac{43 \ 1/m}{63 \ 1/m} = 0.683$
- case C: at bottom 8 1/2 hole, 6 1/2 DC at bottom 8 1/2 hole, 5 " drill pipes  $r = \frac{15 \text{ } 1/\text{m}}{24 \text{ } 1/\text{m}} = 0.625$

# MAXIMUM BUBBLE ALLOWABLE VOLUM :

- CASE A: hb = 59m, we can have a maximum influx of gas around 6.5 m<sup>3</sup>.
- CASE B: hb = 405m. It exceeds height of 8" DC. By considering an average r including string upper 8" DC., we have hb = 310m with an average linear volum of 51 1/m. The maximum influx of gas is then  $16m^3$ .
- CASE C: hb = 725m. same considerations give a maximum gas influx of  $10m^3$  without control problems.

string	string	string
ш 099	2460 m	3160 m
750 m	2550 m	3250 m
at	at	at
BUTT	VAM	VAM
133 lbs	72 lbs	53.51bs
K55	N80/P110	P110
20"	13 3/8	9 5/8

				i			
	BUR STANDART	BURST 1.1 SAFETY ART FACTOR	COLLAPSE SANDART F	PSE SAFETY FACTOR	TENS	TENSION 1.5. SAFETY FACTOR	WEIGHT IN AIR
20"	211	190	103	. 95	945	630	129
13 3/8 N80	371	337	184	143	738	492	259
13 3/8 P110	510	464	199	164	970	647	259
9 5/8 P110	752	684	547	427	759	506	256
	kg/cm <sup>2</sup>	kg/cm <sup>2</sup>	kg/cm <sup>2</sup>	kg/cm <sup>2</sup>	10 <sup>3</sup> da N	10 <sup>3</sup> da N 10 <sup>3</sup> da N 10 <sup>3</sup> da N	10 <sup>3</sup> da N

NB: For collapse, safety factor depends on the weight of the string in mud. But we considered weight in air, which is a greater safety factor.

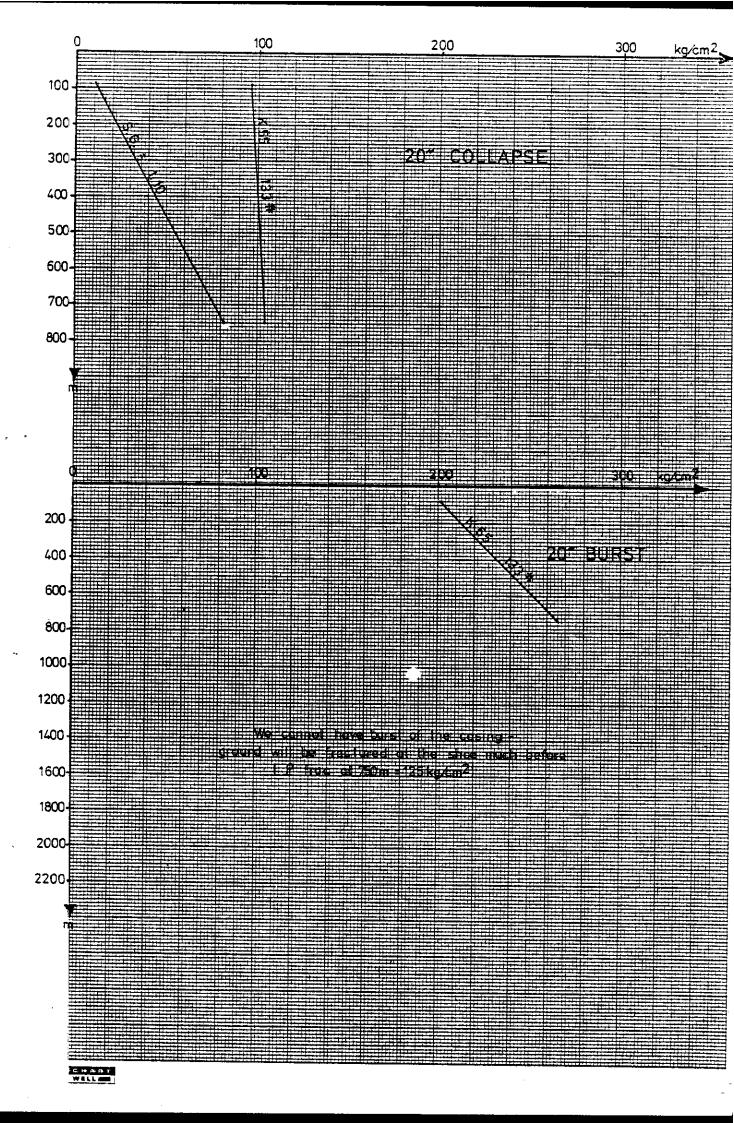
For burst, we admit that fluid behind casing is 1.0 specific gravity. For collapse, we admit that fluid behind casing has the maximum specific gravity of the mud used during the corresponding phase.

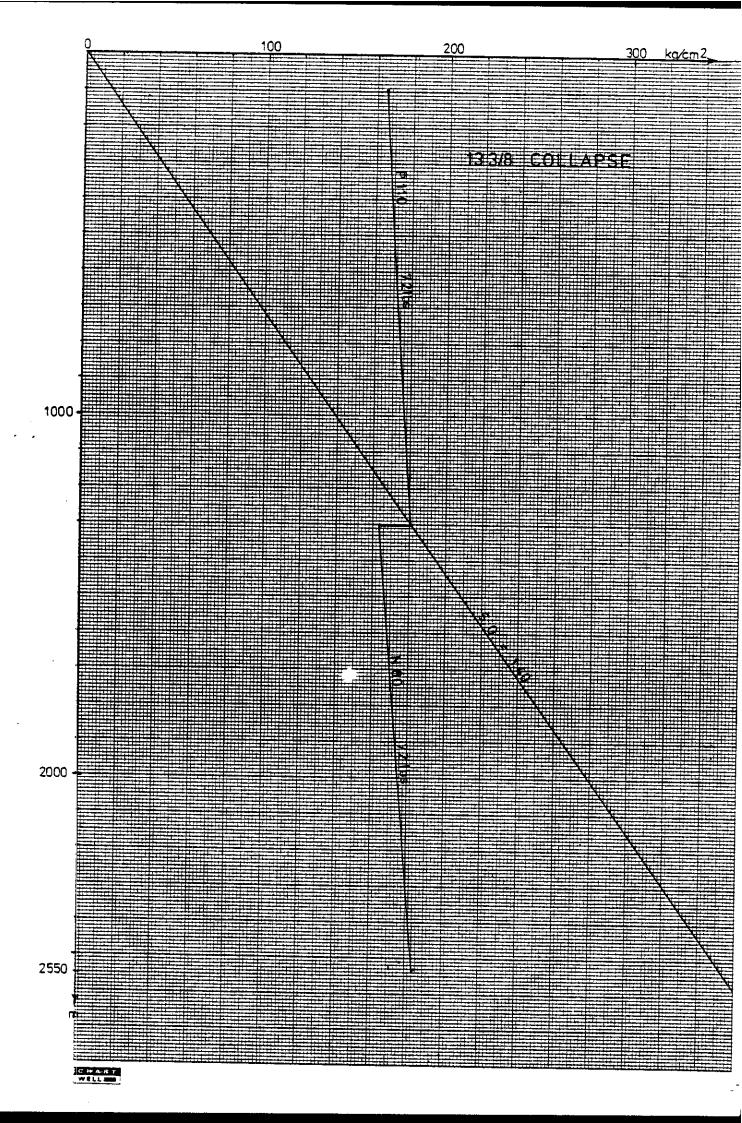
### **CONCLUSIONS:**

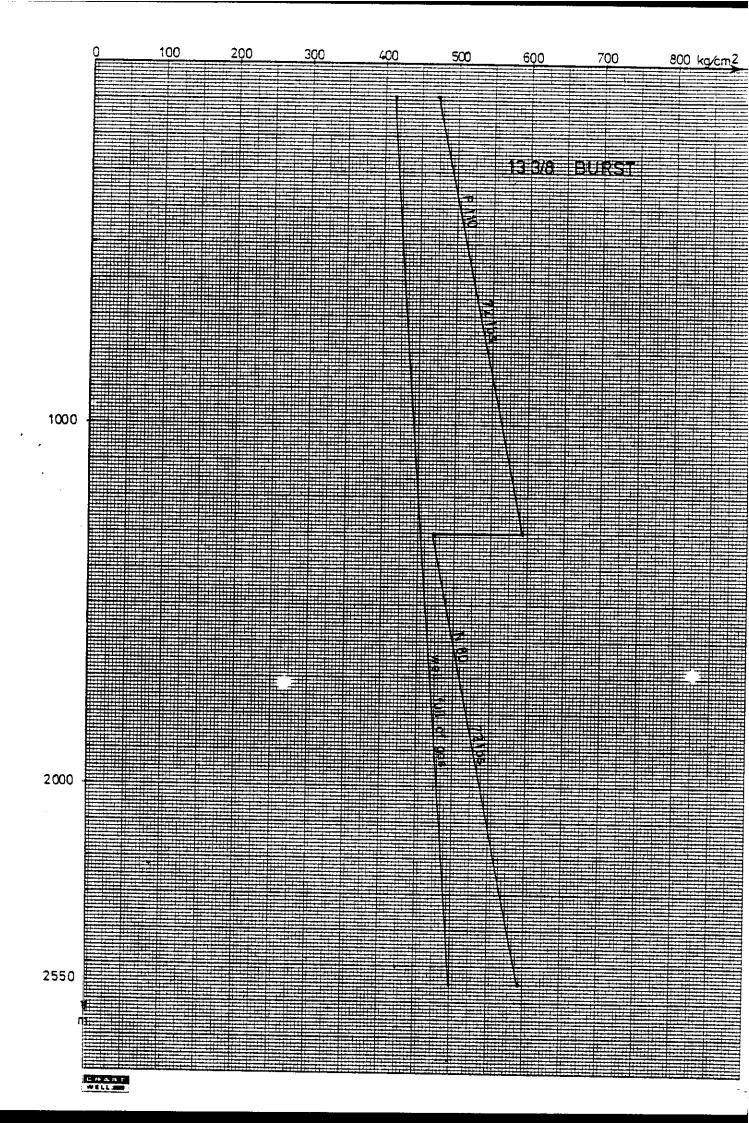
20" casing can be emptied in its full length. During 17 1/2 drilling, a  $6.5m^3$  gas bubble can be admitted. In any case, there will not be a burst from 20" casing but possibl fracturation at shoe.

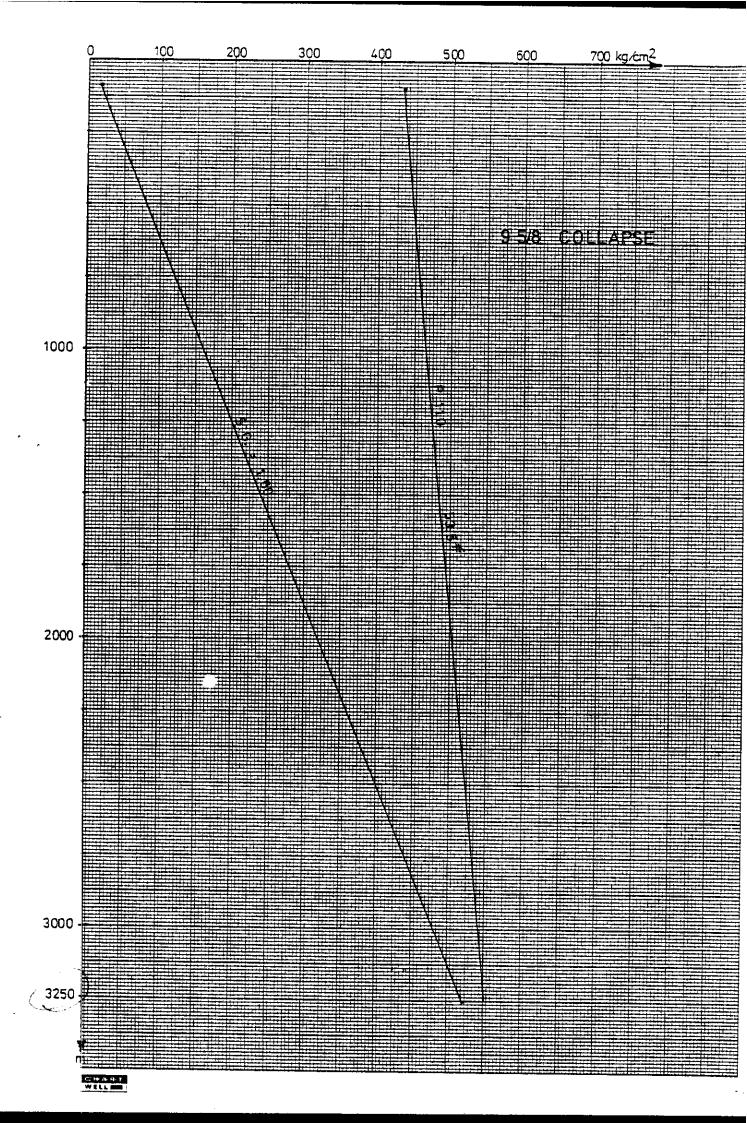
13 3/8 casing can be emptied down to 1300 m. During 12 1/4 drilling, a  $16\text{m}^3$  gas bubble can be admitted. The well can be full of gas without burst of casing. In that case, pressure at shoe will be around 500 kg/cm<sup>2</sup> (compared to an estimated P frac = 460 kg/cm<sup>2</sup>.) There will be fracturation at shoe.

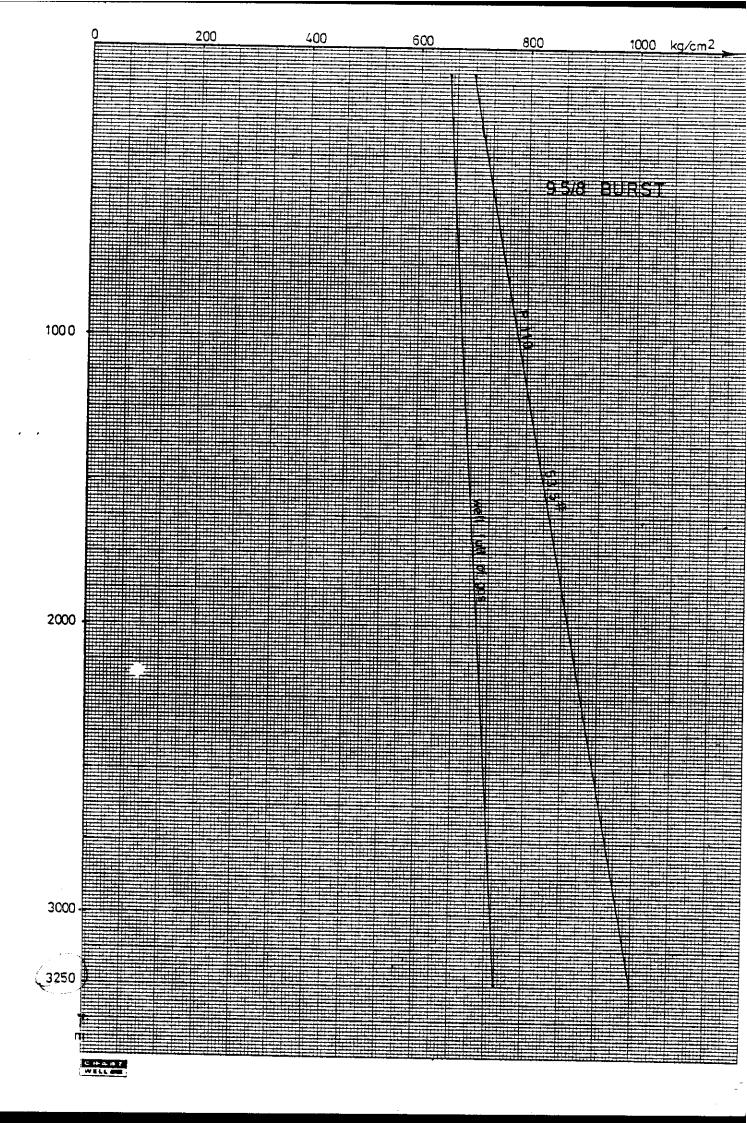
9 5/8 casing can be emptied in its full length. During 8 1/2 drilling, a  $10\text{m}^3$  gas bubble can be admitted. The well can be full of gas without burst of casing. In that case, pressure at shoe will be around 730 hg/cm<sup>2</sup> (compared to an estimated P frac = 675 hg/cm<sup>2</sup>). There will be fracturation at shoe.



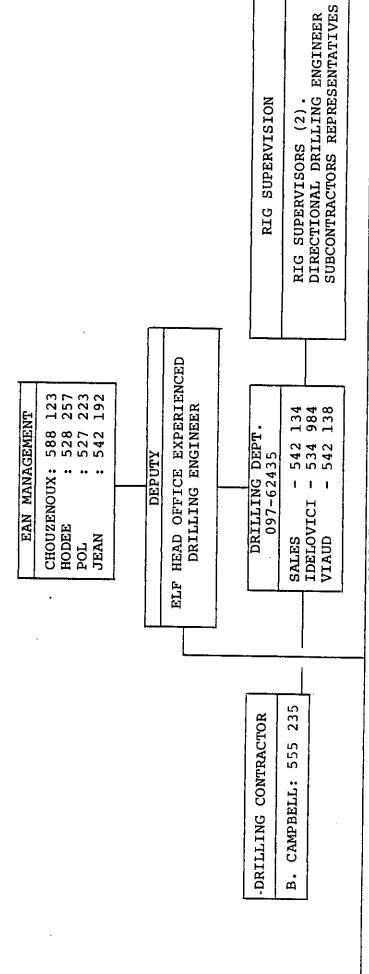








ELF AQUITAINE NORGE A/S EMERGENCY TASK FORCE



				COORDI	COORDINATION STAFF	PF				
PROD DIV. SAFETY	SAFETY	LOGISTIC	EXPL.DIV.	RESERVOIR	LOGISTIC EXPL.DIV. RESERVOIR PURCH/MAT CORPORATE PERSONNEL	CORPORATE	PERSONNEL	INFO.	TET.ECOM	
097-58116 097-64538 TRAFFIC 097-6453	097-64538	AIR TRAFFIC 097-64539	ENGI-097-64089 NEERING	ENGI- NEERING	WAREHOUSE AFFAIRS 097-47366	AFFAIRS			ENGINEER	
DE ST. PALAIS	3JØRNESTAD	BJØRNESTAD BERGSAKER J. JUNCA BATHIARD	J. JUNCA	BATHIARD	B. VAAGE	HEIDE	A.RØSSETH	1	HAUKALI E. GRUDE	
525185	560352	662572	591353	588312	585656	590311	542278	560621	74644	
MONTEIL	SELNES	JOHANNESEN	JOHANNESEN GOUYONNET	HAMRE		H. HERVIK	H. HERVIK JOHANNSEN	=-	1	
547083	696311	Gunnar 580287	597009	097-47364		556083	590198			

### CASING TESTS / LEAK OFF TESTS PROCEDURES

### PRESSURE TESTING AFTER SETTING INTERMEDIATE CASING

### I - Casing test

- 11 Casing testing requires a high pressure pumping unit with measuring tanks. Casing is tested prior to drilling out cement, first with drill string out of hole (to test total shut-in) and then with bit on cementing collar (to test shutin around the pipe if the casing has not been already tested at end of cementing operation).
- 12 Pressure should be increased gradually (50 to 200 l/min). The pressure-volume relationship should be plotted. (Record one point every 50 to 200 l depending upon the total volume to be pumped, and at least 5 points.) If there is no leakoff, the relationship is linear. See Example on Fig. l attached.
- 13 Under no circumstances should the maximum pressure exceed:
  - . The working pressure of the wellhead.
  - Ninety percent of the internal yield pressure of the most exposed casing pipe, which is not necessarily the top pipe, (mud weight in string and annulus are to be considered).

The test pressure less than the two preceeding measures may be set if justified by the maximum operating pressure anticipated at the wellhead during the subsequent drilling phase.

- 14 Pressure should be maintained for 15 minutes. The pressure test is considered positive, when the pressure drops less than 10 percent during this time. The pressure test should be recorded.
- 15 Release pressure and measure the mud returns in the measuring tank.
  Compare with the theoretical volume required, taking into account size and length of casing (see Chart 1).

# 2 - Casing seat testing.

The purpose of this testing procedure is to test the resistance of the formations immediately beneath the casing shoe and the quality of the cement sheath around the shoe, if the latter is set in a relatively imperious zone (clay, shale etc.).

# 21 - Drill out cement and drill:

- . 3 to 10 m new hole, if the formation is impervious, when testing the quality of the cement sheath;
- . not more than 50 m new hole, if a permeable zone is expected in this interval.
- 22 If necessary, circulate until mud weight is constant.
- 23 Pull the bit up to the shoe, close the B.O.P. and pump mud into the drill pipe at a speed between 50 and 100 l/min. Plot the pressure-volume relationship on a graph (one point every 50 or 100 1).
- 24 Stop the pump when one of the three following preset requirements is fulfilled:
  - Class A test: When the pressure attains a preset level, considered sufficient to cope with the problems anticipated during the subsequent drilling phase.

Class B test:

When two consecutive points are distinctly out of the normal trend set by the linear relationship of the previous points (see reference line recorded during casing test). The point of deviations from the linear relationship marks the beginning of fluid leakoff into the formation. The corresponding bottom-hole pressure is the initial squeeze pressure by the depth at which fluid leakoff takes place. See Fig. 2.

Class C test:

When after passing through a maximum, the pressure drops rapidly, the bottom-hole pressure corresponding to the maximum is the fracturing pressure "FP". The equivalent fracturing density "FD" is calculated by dividing the fracturing pressure by the depth at which fracturing takes place. See Fig. 3 and 4. In the case of Fig. 4, it has not been possible to detect the leakoff point due to excessive pump speed. This is to be avoided.

- 25 After shut down of the pump, maintain pressure of 10 minutes and record the pressure drop every minute.
- 26 Release pressure and measure the mud returns. Compare to the returns after casing test.

# 3 - Open hole pressure test.

An open hole pressure test while drilling may be justified in the following cases:

- After drilling through a permeable zone.
- Before entering into transition zone.
- Before entering into a doubtful zone.
- Before increasing mud weight significantly.

The higher the open hole, the greater the number of anomalies while increasing pressure and therefore the difficulties encountered for determining the actual leakoff point.

- 31 While pulling out, stop the bit at the shoe and proceed as set forth in para 23.
- 32 If due to leakoff, it appears that the pressure rise will last more than half-an-hour, stop the pump, release pressure, and resume operation with a higher pump speed (200 to 300 l/min.).

# 33 - Stop the pump:

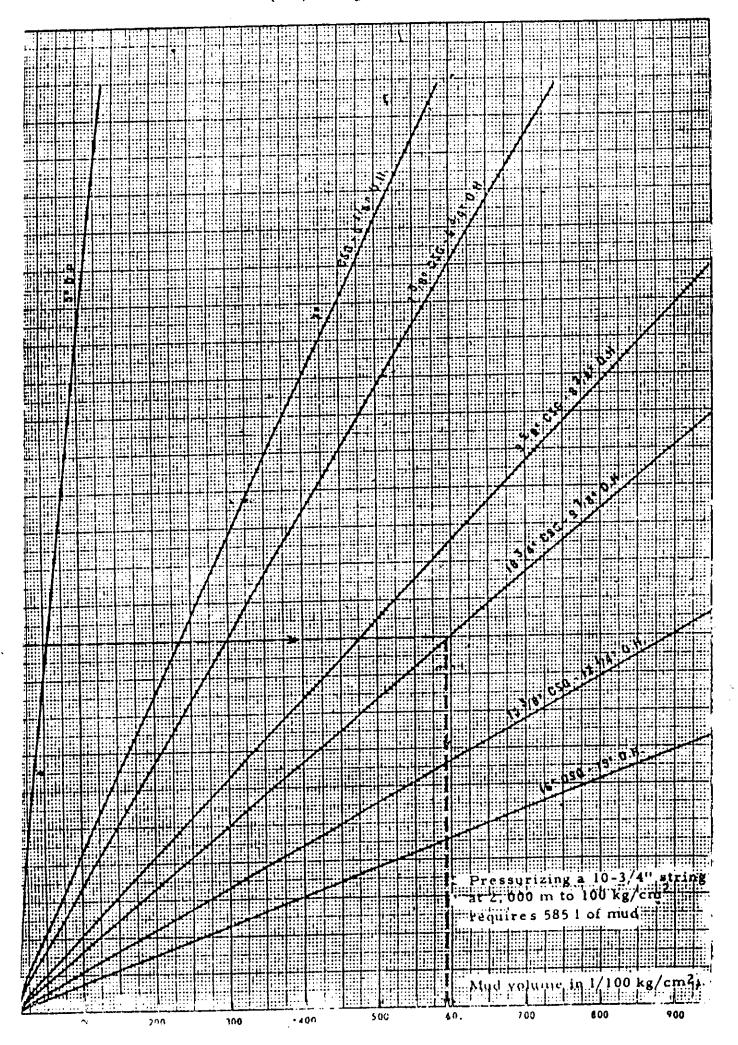
- either when the preset pressure has been attained (this pressure being for instance calculated so that the bottom-hole pressure at the shoes does not exceed the squeeze pressure determined during the first test);
- or when 3 or 4 successive points deviate from the average linear relationship in the neighbourhood of the normally expected pressure drop (see Fig. 5).
- 34 Complete the test as set forth in para. 25 and 26.

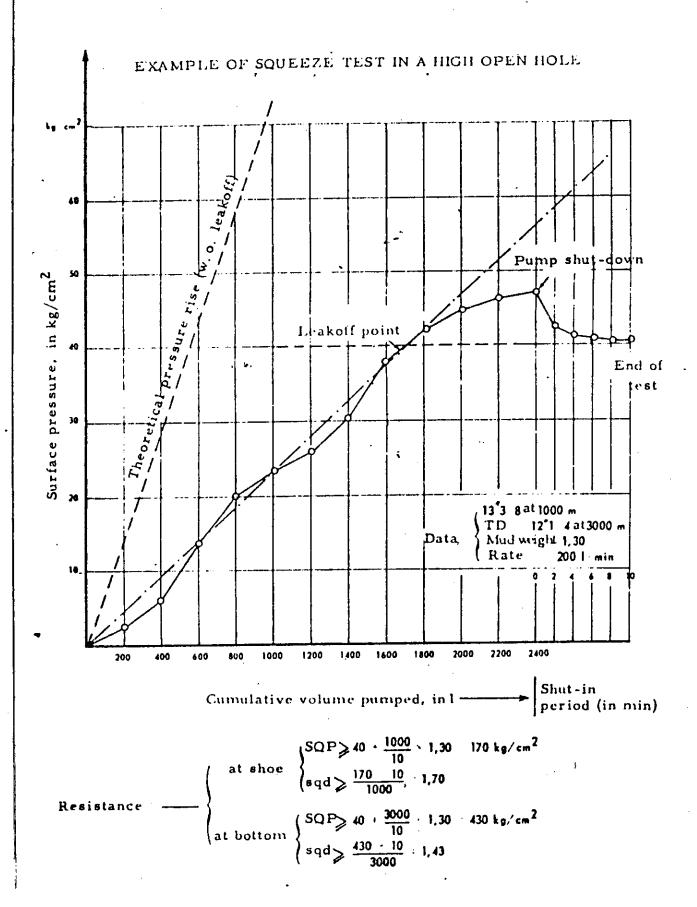
# 4 - Important note.

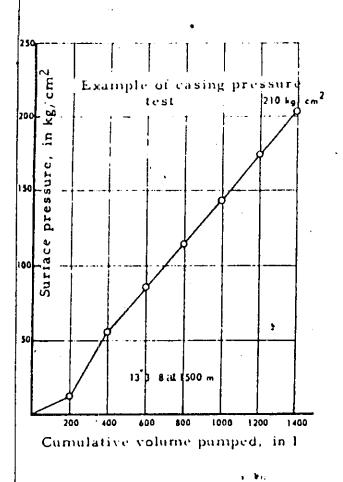
Watch annulus between strings while carrying out pressure tests and bleed pressure off if required.

When the access to this annulus is impossible carefully observe the volumes pumped during the <u>casing test</u> (prior to drilling out cement) and ascertain that they are in accordance within 10 percent with the theoretical volumes shown in Chart No 1, since monitoring the mud volumes consitutes the only means available to check if a string is leak-proof and if the pressure is not taken by preceding string. I should be noted that pressuring the annulus between two strings may burst the outer string if the annulus between the two strings is shut-o (cement top above shoe).

When pressure testing below the shoe, mud volume monitoring becomes inefficient due to fluid leakoff into the formation, such leakoff increasing with the height of the open hole (see Fig. 5). However, if the casing has been shown to be leak-proof during the casing test, it will be sufficient, during the subsequent tests in the open hole, to stay below the casing test pressure to avoid any risk of "wild" leak through the annulus.







Heakdii

point

Examply of fracturing ht shoe

Data

SQP 135

sqd

Ł b

400

latavio

7 5 8 pt 3000 m

Rate 100 1/min

615 . 10

3000

 $630 \times 10$ 

3000

volume pumped

1000

Mudweight 1,60

150 - 3000 - 1,60

2,10

1200

2,05

TD = 3030

140

130

100

80

E 120

LESEL

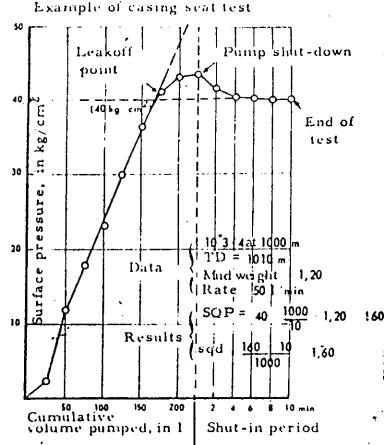


Fig. 3

down

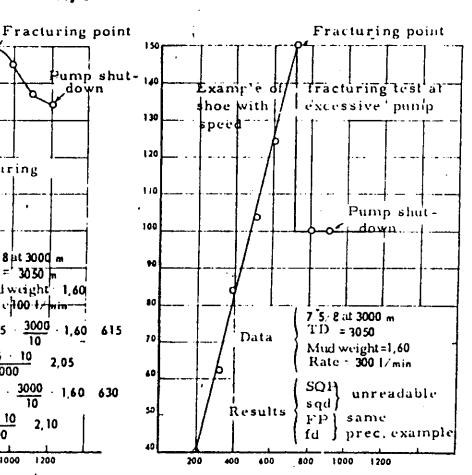


Fig. 4

ELF NORGE A/S STAVANGER

From : DRILLING SUPERINTENDANT

To : ELF NORGE A/S - SUPERVISORS

### RESPONSIBILITIES

- 1. Elf Norge A/S toolpusher on a Contractor rig is the Company representative on that rig.
- 2. No visitors are allowed when not announced by Drilling head office.
- 3. It is always the Elf Norge A/S responsibility when supervising contractor-operated rigs, regardless of the type of contract, to see that drilling operations are conducted in such manner that adequate well control is maintained at all times.

Well control starts with the planning of the well prognosis and includes an alert drilling crew having knowledge of possible hazards, mud control techniques and the mechanical facilities to detect at the earliest possible moment indications of trouble. It requires proper training of contractor personnel and continuous surveillance of the drilling operations by all concerned.

4. Elf Norge A/S is responsible for rig safety at all times; apart from being in charge during normal drilling operations, he will have the responsibility for well completion and well abandonment in connection with Drilling Department.

Sout

Drilling Superintendant (R.Sales)

Stavanger,

# WELL KILLING CONTINGENCY PLAN WHEN DRILLING EXPLORATION WELL

### 1. INTRODUCTION

The only method to kill a flowing well is to drill relief wells.

This contingency plan is established for exploration well in case of blow-out occurence after the 13"3/8 casing or 9 5/8 casing have been set.

# II. GENERAL PROGRAM

# II.I. Number of Relief Wells

The general method consists in drilling two relief wells.

- The first one to intercept the flowing well around the shoe of the last casing string set.
- The second one to intercept the well in the flowing formation (generally the bottom of the hole).

This figure permits to pump heavy mud at two different levels in the flowing well and have best results for killing wells particularly in case of blow-out.

# II.2. Positionning Drilling Platform

II.2.I. <u>Distance between flowing well and relief well</u>.

The spud location depends on depth reached and design of the rig drilling the relief well.

# II.2.2. Position of the rig

The rigs must be located following the winds and currents.

# II.3. Objectives

Around the shoe of the last casing string set and the flowing formation. The objective will be chosen according to available geological data and electrical logs in order to have the best facilities to establish communication between the wells, and deviation surveys (single shot surveys during drilling and Multishot Gyro surveys in casing strings) obtained during drilling operations.

### II.4. Well Profiles

- II.4.1. Direction and inclination will be determined exactly according to the objectives.
- II.4.2. The target should be within 30/50m from bloming well. Such accuracies should be obtained by using good survey equipment, and is sufficient to obtain communication between the wells.

# II.5. Casing Program

Setting depth of casing will be determined according to the objectives and geological results encountered during drilling.

### III. KILLING PROCEDURE

- 7" liner will be set approx. 100m above the target
- Drill to approx. 50m below liner shoe
- For fracturing and heavy mud injection a 3" DP will be run to the top of the liner (owing to high pressure loss it is not recommended to run the drill string inside the 7" liner)
- Pump heavy mud until blow out stop
- When the blow-out is controlled, squeeze cement to plug the well and the bore hole.

# IV. PARAMETERS DURING KILLING OPERATION

#### IV.1. Mud Weight

Mud weight will be adjusted according to knowledges obtained during drilling.

## IV.2. Volume of mud

The experience shows that it is necessary to have an important volume of heavy mud to succeed in a blow-out control. About three ot four times the volume of the flowing well may be required after pumping sea water in high quantities. In this case we need:

- For blow-out when drilling 12 1/4 hole:  $750m^3$ .
- For blow-out when drilling 8 \( \frac{1}{2} \) hole : 500m<sup>3</sup>.

This volume must be ready before drilling out 9"5/8 shoe in the relief well(s).

# IV.3. Pumping rate

Heavy mud must be pumped at a high rate in the flowing well as soon as the communication is established. (Pumping rate could be estimated at about  $6m^3/Min.$ ).

# V . EQUIPMENT AND MATERIALS

### Drilling Platforms

No problem is foreseen in making drilling platforms available to drill relief wells, as NSOC has pledged to maintain a current list of equipment available in an emergency situation.

## Well Heads and Casing

Sufficient well head equipment and casing is available from own stock and partners' stocks.

## Mud Chemicals and Baryte

Baryte needed is available at Tananger and Dusavik bases on a permanent basis.

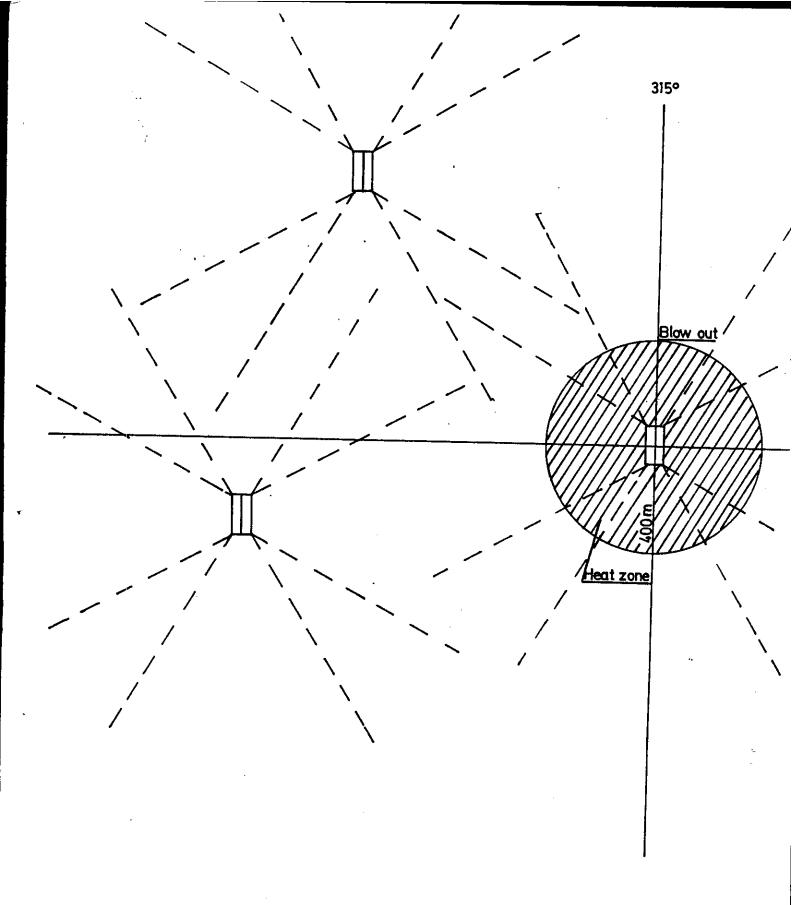
#### Pumps

Generally the North Sea drilling platforms are equipped with a pumping power of 4000 - 4800 mech.HP. The addition of three or four more pumping skids will bring the total mechanical HP up to approx. 7300 HP.

Calculations shows that the mechanical HP required on each rig when pumping at 6000 l/min. is the range of 6800 HP. So, by installing some additional pumping unit we will have sufficient power to perform the operation.

# VI . ORGANISATION

Emergency task force is shown in table No.2. Technical and personnel assistance will be given by ELF Aquitaine heat office and ELF Aquitaine subsidiaries when necessary.



ANCHOR PATTERN (SCHEMATIC)