$S_{+} = S\Sigma(\Delta hA)$ where

 S_{+} = change in storage (m^3)

S = Storage coefficient (dimensionless)

 Δh = change in head for each grid zone (m)

A = area of each grid zone (m²)

The change in storage was calculated to be $10^7 \, \mathrm{m}^3$ over the 20 year pumping period, or an average of 500 M1 per year.

2. Leakage

For the southern part of the area where significant leakage is likely to occur, the hydraulic conductivity between the aquifers was calculated to be in the range 3×10^{-3} to $3 \times 10^{-2} \text{m}^3 \text{day}^{-1} \text{m}^{-2}$ (Appendix 4). Knowing the head difference between the aquifers this allows the calculation of the rate of leakage from the confined aquifer. The head difference does not vary much seasonally, however geological consideration suggest that leakage will vary over the area, and this method is unlikely to be as reliable as the depletion method.

The thickness of clays separating the aquifers varies from zero to about 5 metres, and a value of 2.5 metres has been arbitrarily chosen for a test calculation.

The head difference varies from zero to 2 metres in the zone of leakage and a value of 1 metre has been used for the calculation.

Using Darcy's Law, the rate of leakage can be calculated:-

$$Q = K - \frac{h}{b}$$
 A, where

Q is rate of leakage through confining bed

K is vertical hydraulic conductivity of confining bed $(0.015 \text{ m}^3\text{day}^{-1}\text{m}^{-2})$

h is head difference between aquifers (1 metre)

b is thickness of confining beds (2.5 metres)

 $(\frac{h}{b}$ = hydraulic gradient across confining bed)

A is area over which leakage is calculated (100 km^2) Hence Q = 2 x 10^8m^3 year⁻¹ or 200 000 M1 per year.

This value for leakage is nearly 10 times the estimated rate of extraction from the confined aquifer. The value of head difference can vary by 100% at most and 1 metre seems a reasonable value, but the values of confining bed thickness and hydraulic conductivity can each vary by at least an order of magnitude. A minimum value of leakage of 4.5 x $10^6 \mathrm{m}^3$ year $^{-1}$ (4 500 M1 per year) can be calculated selecting extreme values of the parameters (K = 0.003, b = 25). This value represents about 20% of the amount extracted from the confined aquifer. The main uncertainty is the value of K, because only one aquifer test is available for its determination, however the true value of leakage is unlikely to be less than $4.5 \times 10^6 \mathrm{m}^3$ year $^{-1}$.

3. Discussion

Depletion of the upper aquifer is being nearly balanced by natural recharge from the rivers, and re-circulation of excess irrigation water. This means that the observed change in storage in the unconfined aquifer is not a measure of leakage, but of the disparity between leakage from the aquifer and recharge to it. The low transmissivity of the aquifer between recharge zones along the rivers and the flanking irrigation areas requires that a steep hydraulic gradient be developed towards the areas where leakage is occurring.

RECALCULATION OF AQUIFER TEST DATA (J. Sinclair)

RE-EVALUATION OF 1969 AQUIFER TESTS

INTRODUCTION

Partially complete tests were carried out on DM4, located beside the Bremer River, near Langhorne Creek. These may have been flooded out, so arrangements were made to move to another site further from the river. Successful aquifer tests were then performed at two sites, one near Langhorne Creek, and one near the Bremer River in the south of the area (Figure 47).

METHOD

The transmissivity and storage coefficients of a leaky artesian aquifer with fully penetrating wells, without water being released from storage in the aquitard and constant discharge conditions, and the hydraulic conductivity of the overlying aquitard were determined from aquifer test data by following the type curve graphical method.

The family of curves $(\frac{r}{B})$ used were NON STEADY STATE LEAKY ARTESIAN TYPE CURVES, which plot W(u, $\frac{r}{B}$) against $\frac{1}{u}$.

The drawdown data are plotted against the corresponding values of time on double logarithmic paper. A comparison with the WALTON family of type curves shows that the plotted points fall along a curve for $\frac{r}{B}$.

A point where $W(u, \frac{r}{B})$ and $\frac{1}{u}$ are simple is chosen as the match point. Then the co-ordinates of this point are read from the observed data sheet. The appropriate numerical values are introduced into the following equations, allowing the calculations of the aquifer parameters.

$$T = \frac{Q}{4\pi s} W(u, \frac{r}{B})$$

$$S = \frac{4Tut}{r^2}$$

$$K' = \frac{\left(\frac{\mathbf{r}}{\mathbf{B}}\right)^2 \mathbf{T.b'}}{\mathbf{r}^2}$$

leakage coefficient $(\frac{1}{c}) = \frac{K'}{b}$

where $T = transmissivity (m^3/day/m)$

Q = pumping rate $(m^3/day, originally gallons per hour)$

s = drawdown (m)

 $W(u, \frac{r}{B})$ = well function of u, $\frac{r}{B}$

S = storage coefficient (dimensionless)

t = time (days)

r = distance to observation well from pumped bore (m)

K' = vertical hydraulic conductivity of confining layer $(m^3day^{-1}m^{-2})$

b' = thickness of confining layer (m)

RESULTS

Results are shown in the tables below.

curve match points

	WELL	DRAWDOWN s (feet)	TIME x(min)	$\frac{\mathbf{r}}{\mathbf{B}}$	$W(u,\frac{r}{B})$	$\frac{1}{u}$	
1000 min Q=30,810 GPH	DM1* DM2 DM3	0.40 0.62	4.4 4.9	0.1 0.05	1 1	1 10	
1450 min Q=26,216 GPH	DM3* DM1 DM2	0.625 1.0	1.15 6.0	0.01	1 1	1 1	
1300 min Q=26,300 GPH	DM5* DM6	1.35	3.6	0.000	1	10	

^{*}indicates production well

Meter readings were not recorded for the pump test on DM5, so a visual estimate of pumping rate was made from the discharge rate curve.

SUMMARY OF RESULTS AQUIFER PARAMETERS

Pumped Well, time	well	Transmissivity m ⁵ /day/m	Storage Coefficient
DM3			
1450 min (24hr 10 min)	DM2 DM1	750 1 200	8.41×10^{-4} 1.02×10^{-3}
DM1			
1000 min (16hr 40 min)	DM3	1 400 1 300	4.71 x 10 ⁻⁴ (5.76 x 10 ⁻⁴) (Roberts, 1972)
	DM2	2 200 2 250	7.95 x 10 ⁻⁴ (8.4 x 10 ⁻⁴) (Roberts, 1972)
DM5			
1300 min (21hr 40 min		560 490	1.675 x 10 ⁻⁴ (2.33 x 10 ⁻⁴) (Roberts, 1972)

CONFINING BED DATA

Observation Bore	Depth	thick (fe	mess et)	hydraulic conductivity m day m
DM1	23'- 38' 38'- 58'	clay dry sand	15' 20'	3.1 x 10 ⁻²
DM2	62'- 77'	clays	15'	$9.61 \times 10^{-3}_{-3}$ DM3 pumped 2.97×10^{-3} DM1 pumped
DM3	50' - 60' 60' - 63'	clay dry sand	10' 3; 13 ft	3.43×10^{-3}
DM6	50'-112'	clay	62'	0.00 (no leakage detected)

Aquifer thickness (feet)

Bore	Depth Range	Thickness					
DM1	93'-265'	172 ft	52 m				
DM2	92†-240†	148 ft	45 m				
DM3	85'-250'	165 ft	50 m				
DM5	127'-214' + ?	88 ft ?	27 m				
DM6	122'-215' + ?	93 ft ?	28 m				

(iv)

<u>Leakage Coefficient</u> is measured in day⁻¹. It may be defined as the rate at which water will leak from a unit area of the confining layer per unit drawdown in the aquifer proper. It is the inverse of hydraulic resistance.

i.e. leakage coefficient
$$(\frac{1}{c}) = \frac{p}{m}$$

p' = vertical hydraulic conductivity

m' = aquitard thickness

c = hydraulic resistance

DM1 Leakage coefficient =
$$\frac{3.10 \times 10^{-2}}{35 \times 0.3048}$$
 = 2.9 x 10⁻³ days⁻¹

$$p' = 3.10 \times 10^{-3}$$

$$m' = 35 \times 0.3048$$

DM2 Leakage coefficient =
$$\frac{2.97 \times 10^{-3}}{15 \times 0.3048}$$
 = 6.50 x 10⁻⁴ days⁻¹

$$p' = 2.97 \times 10^{-3} \text{ m}^3/\text{day/m}$$

$$m' = 15 \times 0.3048 \text{ m}$$

DM3 Leakage coefficient =
$$\frac{3.43 \times 10^{-3}}{13 \times 0.3048}$$
 = 8.66 x 10⁻⁴ days⁻¹

$$p' = 3.43 \times 10^{-3} \text{ m}^3/\text{day/m}$$

$$m' = 13 \times 0.3048 m$$

DM6 Leakage coefficient = 0.00

$$p' = 0.00$$

$$m' = 62 \times 0.3048$$

DISCUSSION

Results calculated compare well with those obtained by Roberts (1972). There is one anomaly however; the transmissivity values for DM2 obtained from tests at both DM1 and DM3 differ considerably.

To determine the thickness of the upper confining bed, well logs were consulted, and deductions made according to water level cut information and stratigraphy. The upper sequence is complex, and values used are approximate.

CONCLUSIONS

An average for the transmissivity of the confined aquifer in the central southern part of the basin is approximately 1 500 m 3 /day/m. Storage coefficient is about 6.5 x 10^{-4} .

Further to the north, transmissivity decreases markedly, to $500 \text{ m}^3/\text{day/m}$, and storage coefficient is about 2 x 10^{-4} . The aquifer is thinner at this location.

No leakage would seem to occur through the upper confining layer in the northern part of the area. Although the leakage coefficient varies for the 3 wells in the southern part of the basin, the range is consistent $(2.9-0.07 \times 10^{-3} \text{ days}^{-1})$.

ESTIMATION OF WATER CONSUMPTION OF EVAPORATED CROPS FROM AERIAL PHOTOGRAPHS

Colour aerial photographs flown in March, 1976 were used to measure the area of irrigated crop for the 1975-76 season. A fairly good differentiation of land use could be determined upon inspection of the photographs. Uncertain areas (whether or not irrigated, and the type of land use) were clarified on field trips.

The irrigated paddocks were traced from the photographs directly onto a transparent map of corresponding scale (1:20 000). They were traced again onto good quality paper, then cut out, the weight of paper representing an area of land. By weighing several sheets of paper of known area, to obtain an average, it was possible to determine the area of irrigated land from the weight ratio.

Some error occurred when tracing the areas from the photographs to the map. Aerial photography becomes distorted at the edges due to lens aberrations so the fit of photographs at their edges was not perfect. One could expect an under-estimation of irrigated area in the results obtained.

Results are tabulated below. Figure 15 shows the distribution of irrigated land in March 1976.

TABLE
Areas of Irrigated Land

LAND USE	AREA (km²)
Bremer River area : lucerne Angas River area : lucerne Mosquito Creek area : lucerne	18.3 4.9 0.9
TOTAL : Groundwater irrigated lucerne	24.1
Lake Alexandrina irrigated lucerne Vineyards (mainly river flooding) Orchards (mainly river flooding)	1.4 4.7 1.7
TOTAL AREA IRRIGATED	31.9

ESTIMATE OF VOLUME OF WATER WITHDRAWN

The total lucerne area irrigated with water from the Tertiary aquifer is 24.1 $\mbox{km}^2.$

The value used for the water requirement of lucerne is 1 042 mm (Holmes and Watson, 1969). The average annual rainfall at Langhorne Creek is about 375 mm, and the balance (670 mm) is provided by irrigation.

Water consumption = irrigation x irrigated area $= 0.675 \times 24.1 \times 10^{6}$ $= 16.2 \times 10^{6} \text{m}^{3}/\text{year}$ = 16.000 Ml/year

The value of 16 000 M1/year is the estimate of water provided for lucerne evapotranspiration in an average year. The amount of water applied will be greater because it is virtually impossible to apply exactly the correct amount of water, and an excess must be applied to leach salt from the plant root zone. The actual amount of water applied to lucerne in the area has never been measured, but it is probably in the range 800 to 1 200 mm, or 20 000 to 28 000 M1 year over the entire irrigation area.

A figure of 25 000 M1 year $^{-1}$ is used here; an error of $^{\pm}$ 30% is highly probable for the estimate of extraction from the confined aquifer until better methods are used.

DISCUSSION OF THE GROUNDWATER SYSTEM PRIOR TO THE COMMENCEMENT OF MAJOR IRRIGATION

It is instructive to consider the probable groundwater system prior to the large scale irrigation withdrawals (i.e. before about 1960).

The cones of depression in both aquifers south of
Langhorne Creek can reasonably be attributed to irrigation
pumping, and prior to their formation groundwater flow would
have been in a southerly direction throughout the area.

One well near the lake is known to have flowed until 1963, and a component of decreasing upward leakage from the confined into the unconfined aquifer is inferred for that period. Information from local farmers suggests that water levels in the confined aquifer near the lake were 1.5 metres above ground level before irrigation commenced.

1. Confined Aquifer

Recharge would have occurred mainly in the north and north-west, from the rivers, with a small component of intake from the unconfined aquifer in the north where the head difference was appropriate. Outflow would have been beneath the lake, with some upward leakage into the unconfined aquifer in the southern part of the area.

2. Unconfined aquifer

Recharge would have occurred along the northern sections of the rivers, and along Mosquito Creek. The southern sections of the rivers may have been responsible for less recharge than at present, because water levels are known to have fallen several metres recently creating greater storage capacity. Another source of recharge would have been upward leakage from the confined aquifer.

Discharge along the lake margin could have been expected (and local farmers report that springs did exist).

Higher groundwater levels than at present would have allowed a larger area for evaporative discharge.

With relatively saline water recharging the aquifer, and some evaporative discharge it is understandable that very high salinities could occur, and it may be that modern high salinity zones are an indication of areas of maximum evaporative discharge from the aquifer prior to 1960.

The probable system is shown diagrammatically on Figure 38, together with the modern situation for comparison.

STATISTICAL TREATMENT OF SALINITY DATA

SALINITY DATA

Time/salinity data was treated statistically using the following relationship:-

Salinity = $(a_0 + s_0) + (a_1 + s_1)$ (year - 1970), where

 a_0 : salinity, 1970 (estimated from regression line)

 ${\bf a}_1$: yearly change, positive increasing (estimated from regression line)

 ${\rm r}^2$: "goodness of fit" between regression line and field data

Syx: one standard deviation from regression line, in salinity

 s_{o} : one standard deviation from a_{o} estimate, in salinity

 \mathbf{s}_1 : one standard deviation from \mathbf{a}_1 estimate, in salinity

The results are presented overleaf, followed by the field data.

Well	readings taken	Simulated Salinity 1970	yearly change	regression coefficient	standard error along	standard error on a	standard error on a ₁
	(No. of years)	(mg/1)	(mg/1)		salinity line s=a _o +a ₁ x	0	1
					(co-vari-		
				r^2	ance)		_
		a _o	^a 1	r	Syx	s _o	s ₁
BRM101	5	3160	160	0.39	365.60	600.75	115.61
102	7	3250	35.45	0.01	910	687	154.1 63.9
103	5	3170 3070	-92.6 51.5	0.41 0.15	246 360	326 287	61.8
104 105	6 6	5390	-131.4	0.13	373	430	89.3
103	4	3820	41.0	0.06	249	626	111.5
107	6	3740	104.4	0.33	438	349	75.1
108	4	5610	20.0	0	780	1618	349
109	5	5084	81.0	0.37	294	246	61.1
110	6	4417	14.3	0.02	226	211	54.1
111	6 3 5	3980	121.4	0.79	134	359	61.9
112		2290	118.3	0.99 0.92	28.4 81.7	31.1 236	6.2 57.7
113 114	3 2 7	1730 2266	200.0 83.33	0.94	01./	230	3/•/
115	7	3490	62.0	0.24	292	221	49.5
116	4	2540	-5.0	0	168	423	75.3
117	4	1000	657.1	0.96	293	540	99.0
118	7	5015	16.4	0.05	211	138	32.7
119	3	5330	50.0	0.11	204	874	144.3
120	6	4180	-165.7	0.38	445	512 2541	106.4 463.9
121 122	3 7	2960 2910	139.3 26.4	0.08 0.34	1002 106	69	16.4
123	7	4930	275.0	0.13	1662	1405	314.1
124	2	1575	+175	0.13	1002	1100	22.112
125	4	3590	-45.0	0.12	189	475	84.7
126	1	2350					
127	3 7	2730	100.0	0.92	40	175	28.9
128		2250	46.4	0.49	111	76	21.0
129 130	7	3800 3350	81.9	0.53	224	146	34.6
131	1 4	3935	30.0	0.01	543	1364	243.0
132	i	6600	50.0	0.01			
133	4	4865	70.0	0.11	309	775	138.0
134	2	450	500				
135	4	3975	0.0	0.0	237	595	196.1
136	4	5315	-55.0	0.15	208	523	93.1
137	4	3265 4275	-5.0 -50.0	0.0 0.09	189 248	475 621	84.7 110.7
138 139	4 4	4273 3700	-29.0	0.04	213	534	95.1
140	7	3080	51.4	0.26	244	171	38.9
141	7	4530	-60.0	0.17	209	523	93.3
142	7 3 3 4	5225	-125.0	0.25	306	1097	216.5
144	3	4775	-175.0	0.64	184	658	129.9
145		3930	15.0	0.01	275	689	122.8
146	8	3830	26.2 75.0	0.10 0.96	211 20	136 87	32.5 14.4
147 148	3 2	1930 3800	100	0.90	40	0/	74.4

Well	readings taken (No. of years)	Simulated Salinity 1970 (mg/1)	yearly change (mg/1)	regression coefficient	standard error along salinity line s=a ₀ +a ₁ x	standard error on a	standard error on ^a 1
		a _o	^a 1	r^2	(co-vari- ance) Syx	s _o	s ₁
FRL101 102 103 104 105 106 107 108 109 110 111 112 113 114 115 116 117 118 119 120 121 122 123 124 125 126 127 128 129 130 131 132 133 134 135 136 137	3548885886857564788248657478475783622	5430 1640 3365 3340 3660 1946 1937 2090 3180 5090 2820 4290 6160 4960 6600 7850 5960 7390 4820 3350 5250 2730 4840 3430 4140 3910 5790 2970 3660 3480 5080	200.0 152.0 -25.0 128.1 151 51.2 17.0 41.7 133.6 49.8 61.6 82.4 224.6 -11.9 -43.6 -400.0 103.6 41.7 20.8 -250 -50.0 43.5 33.4 55.0 336.9 -16.0 78.6 23.1 180.0 31.8 150.0 187.5 336.9 -100.0 54.0 -250 0	0.32 0.59 0.30 0.42 0.45 0.52 0.03 0.22 0.71 0.06 0.21 0.16 0.79 0.00 0.01 0.90 0.45 0.09 0.06 0.10 0.36 0.00 0.12 0.29 0.01 0.32 0.13 0.55 0.83 0.09 0.58 0.74 0.05 0.68	408 233 124 397 445 129 184 210 226 542 318 419 302 505 990 212 271 343 210 240 153 1378 268 1376 208 269 158 255 37 849 378 523 612 96	898 313 114 257 287 83 303 136 146 354 205 358 228 385 794 532 184 221 135 602 99 1283 359 1038 522 183 102 529 28 28 28 385 85 85 86 87 87 88 88 88 88 88 88 88 88	289 73.7 27.2 61.3 68.7 19.9 58.3 32.5 34.9 97.7 49.1 109.0 51.2 97.5 204.9 94.9 51.1 52.9 32.4 107.2 23.6 329.5 84.6 233.0 93.0 50.8 24.3 114.2 6.33 268.5 71.5 80.7 433.0 18.5
STY101 102 103 104 105 106 107	4 6 5 6 5 2 6	7389 1680 3750 2210 4390 1830 2850	-741.4 44.3 -26.7 2.86 68.7 256.0 37.4	0.83 0.48 0.08 0.0 0.50	695 95.9 213 148 205	1058 110 245 119 156	235.1 22.9 51.3 30.7 39.5

WATER SAMPLE CONDUCTIVITY VALUES

1977	44750 2600 3700 4800 4500 4500	4,490 7,490 1,200	ed 2350	3450 430 0 4300 ing 5400 3050 5000 3250
1976	4200 2950 2950 2900 4900 4650 7600 7600	7,400 7,000 7,000 7,000 7,000 7,000 7,000	2700 2700 5850 N.sampled 3300	3500 2600 4650 NW 4200 not working 5400 5450 5000 5000
1975	4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4	23 23 75 75 75 75 75 75 75 75 75 75 75 75 75	00000000000000000000000000000000000000	
1974	4 4 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	2800 2800 2800 2850 2650	4100 4050 5650 3550	2600 3350 4450 6600 5350 3150 3150
1973	5500 (Box 03) 5900 (Box 01) 5200 4100	2300 3400 2800 4700	3500 2900 4500	2300 3800
1972	4500 (Box 01) 3000 3000 5000 3500 5100	3900	3600 3000 4300	2300 4000
1971	4500	2400	2900 7600 (Box 02)	2300 3900
1970	2700 3100 4000 5100	3500	3000	2300
State No.	263007202 263279602 263355802 263278201 263276701 263277102 263277108 263277108	26320002 263207503 263207503 2632004903 263200402 263280002 263280001	263207703 263281002 263281003 263053803 263004605 263004302	263280103 263280107 263281102 263281202 263281204 263094201 263205602 263282801
Observation No.	20000000000000000000000000000000000000	- - - - - - - - - - - - - - - - - - -	-01010101010101010101010101010101010101	C C C C C C C C C C C C C C C C C C C

	1977	4100 3620 3430 4200 repair	2450 4500 4500		4080 (Box 01) 4400 2350	7600 4110 7400	5410	7800 4800	5100	7700 5000	4800 3000	7200 3930	3220
	1976	4 + 4 20 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	2320 2400 4400	2460	74400 7500 7500 7500 7500 7500	2200 4000	3300	7400	5500 5500	7800 5000	5200 3150 3354	6800 3700	6400 3050
	1975	2000 2000 2000 2000 2000 2000 2000 200		2300	23000	2200 2600 3600	2600 4500	7000 NM	5950 5600 6100	7200 4700 700 700 700	2700 5800 5800	3670 3650 3650	5800
	1974	4250 4450 4450 4450 4450 4450	t 0	6400 2600	4 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	/ 0 4 m - 0 0 c g - 0 0 c g	7700 7700 7700	7600 4800	6900 6400	8400 8400 8700 8400	10005	4000 4000 000	6500 3350
IVITY VALUES	1973	3000	0000	2700 2100 4400	44300000000000000000000000000000000000	7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7	2700 4200	9600	7600	7300 4600	2800	5500	6000 2900
SAMPLE CONDUCTIVITY	1972	O C C c x	0000	6000 180 0	3300 4100 2000	2000 2300 2300	4 4 4 4 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	6700 4400	7200	7300	2800 5500	2500 3500	5800
WATER	1971	3100	000		4000 3400 2000	1900 3200	2900		7	7200 4800	2700 3500		5800 3000
	1970	3200	0066	0022	2000 2000 2000 2000	2400	2900 4400	6100 5000	5500	7700	2800	5000	5900 3000
	State No.	265004103 265004101 265004001 265282601 265055702 265005902 265005002	263282001 263282001 263003701	360012402 260009904 360009907	360013102 360002203 360354602	360355502 360002002 360355604 360358003	360358201 360358201	360014501 360014502	360003601 360015301 360015301	360002401 360002401 360003701 360357301	360015801 360015801 360358001	260225501 360325501 360358104	360004801 360360902
	Observation No.	BRW. 138 139 140 142 143 144 145	747 148	FRL. 101 102	707. 707. 700.	7 7 7 7 7 0 0 8 7	- -	2 1 2 1 2 4	277 277 201	-	27777 27007 27007 27007	127 727 725	127 128

		1977	3700 ore	0	0	000%	4000		1990 3600	4800	3000
		1976	4800 3700 3800 before	6000	6700	6400 6100	4000	3250	3500 1950 3500	2300 4800	3510
		1975	4350 3600	5400	0009	7400 5450	N.H. 2750	\ -	3350 1850 NM	2100 NE	DK1 2850
		1974	4650 3650	5700	6200	5500 6300	3800 3000	3250	3800 2000 3850	2400 4950	2850 2850
	ITY VALUES	1973	4100 3600	6700	5400	4800	3800 (Box 03)		1700 3400	2100 4400 6400	2600
-5-	WATER SAMPLE CONDUCTIVITY VALUES	1972	3500	4600	5200	00/.4	3900 (Box 03)		6300 1800 3800	2100 4500	2800
	WATER S	1971		5500	5600	4600	3600 (Box 03)				
		1970	3500		5600	2000				2300 4400	3000
		State No.	360336802 360360701	360337801 360355403	360338301	560 <i>55</i> 7802 56000640 1	360337206 360360201	360360202	639354804 639354806 639263903	639354809 639263801	639563501
		Observation No.	FRL. 129 130	131	132	133	<u>た</u> でが での	137	STY. 101 102 103	40C	106 107

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HYDROCHEMISTRY OF SURFACE WATER AND GROUNDWATER (FROM WILLIAMS, 1975)

HYDROCHEMISTRY

Water samples from various depths in each bore were submitted to AMDEL for full analysis. The results are set out in the table at the rear of this Appendix. Most samples were collected from the Quaternary, Pliocene and Miocene aquifers with two from Cambrian aquifers (DM14, DM21) one from an Eocene aquifer (DM26) and four from the Bremer River at stream gauge sites in July, 1973. Results of analyses were plotted as Stiff diagrams (Fig. 42) and on portions of a Piper trilinear diagram (Fig. 43) and a comparison made between sulphate and chloride proportions (Fig. 44).

a. Stiff Diagrams (Fig. 42)

- (i) Surface Water Bremer River
 - A distinct pattern emerges for this water. It must be emphasised that the surface water varies greatly in total dissolved solids with time and presumably also in the proportion of different ions present. The high sulphate: bicarbonate ratio may be a result of pollution from the upstream Nairne pyrite mine. A detailed sampling programme would be necessary to obtain characteristic patterns for the Bremer River.
- (ii) Quaternary aquifer
 Results for this aquifer are varied. Noticeable in many analyses is the high magnesium: sodium ratio. No distinct pattern emerges. Total salt content is generally highest in water from this aquifer.
- (iii) Pliocene aquifer
 Results fall in a group, but cannot always be distinguished from those of different aquifers.

(iv) Miocene aquifer

Similar patterns are again noted. Some are almost identical with those of (iii) which suggests natural hydraulic connection between the two aquifers or poor sampling.

(v) and (vi) Eocene and Cambrian aquifers
The samples are too few to show any characteristic pattern.

b. Piper Diagrams (Fig. 43)

The cation diagram shows a random distribution for all waters except Bremer River samples. The mixed anion-cation diagram shows a similar distribution although plots are more dispersed. There is a slight tendency for the Miocene aquifer waters to have a greater bicarbonate proportion which might be expected considering aquifer chemistry.

The anion diagram is the most useful of the three. It shows a distinct grouping of the surface waters and waters from the Cambrian aquifer (although only two samples). The Miocene aquifer waters are generally lower in chloride and sulphate and higher in bicarbonate compared with the Pliocene and Quaternary aquifer waters.

Where waters from each aquifer intersected in the bore were analysed (DM25, DM27) each showed a different chemistry. In DM25, calcium, magnesium and bicarbonate ion percentages increased with depth, sodium and sulphate decreased while chloride varied only slightly. In contrast, with DM27, ion percentages showed random variation.

c. Sulphate vs chloride proportions (Fig. 44)

A plot of the above ion proportions shows that the surface and Quaternary aquifers form a distinct group having a greater abundance of these two ions. Waters from the Pliocene and Miocene aquifers plot together suggesting hydraulic connection in part. The sulphate-chloride ratios are all similar for pre-Quaternary aquifer waters. The Quaternary aquifer waters show a spread of high and low values. Surface waters have a distinctly higher ratio.

Other plots e.g. calcium vs magnesium and sodium plus potassium vs chloride show only an interspersion of ion proportions and are of no use in distinguishing different aquifer waters.

It is clear that far more analyses are required to detect any significant patterns. It is also suggested that samples obtained during drilling may be mixtures from two or more aquifers if sufficient care is not taken. In future, any hydrochemical analyses should be carried out on samples collected, using a portable pump, from bores which obtain their supply from a single aquifer.

TABLE 3 FULL ANALYSIS DETAILS - MILANG BASIN

RECHARGE INVESTIGATIONS

Ca++ Mg++ Na+

Anions(m. equi v/1

Na/Total cation 1 fm.equiv7

!!	ଞ	29	6	27	26	25	24	23	22	23	8	19	18	17	16	215	=	13	12	=	5	ø	•	7	•	v	•	u	~	-	
Gauge4 Brener	Gauge 2	:	Bremer	27	27	27	26	26	26	25	25	z	25	24	24	23	23	z	22	23, -	. 21	8	8	8	, 19	19	:	17	ï	11-14	
639355101	639BX6101		639266101	:	:	6398K6603	:	:	639BK6902	:	•	:	639007703	1	6398K6703		639BK6205		:	639264301	263275903	:		263277705	:	263277402	263050401	263276205	:	C39 BK 6004	
;	:	.:	Surface	38	8	13	70	25.	13	6	36	×	ø	25	G	21	14	38	2	11.	25.3	35	24	18	8	20	క	42.3	2	7.6	
٠		١	•	Niocene Lst.	Pliocene	Quaternary	Eocene	Quaternary	Quaternary	Miocene Lst.	Miocene Lst. 2677/74	Pliocene	Quaternary	Pliocene	Quaternary	Quaternary	Quaternary	Miocene Lst.	Quaternary	Quaternary	Quaternary	Miocene Lst. 1702/74	Quaternary	Quaternary	Miocene Lst.	Quaternary	Cambrian- Kama.	Pliocene	Cambrian- Kanm.	Quaternary	
117/73	118/73	177/3	116/73	u	3105/74	3104/74	3108/74	2673/74	2672/74	2678/74	2677/74	2676/74	2675/74	20 27/74	2026/74	1921/74	1922/74	1701/74	1700/74	1699/74	1533/74	1702/74	1381/74	1380/74	1142/74	1141/74	921/74	760/74	4660/73	3406/73	
\$/7/73	5/7/73	19/7/73	5/~/73	12/6/74	5/6/~4	4/6/74	11/6/74	15/5/74	14/5/~4	7/5/74	8/5/74	6/5/74	3/5/74	29/4/74	23/4/~4	18/4/74	17/4/74	4/4/74	4/1/74	3/4/74	18/3/74	1/4/74	8/3/74	5/3/74	28/2/74	28/2/74	18/2/74	7/2/74	1/11/73	21/8/73	
1222	1437	1212	1111	10.39	1779	2932	2239	2588	3397	1534	1486	1626	2208	1182	2060	2043	1570	1933	3979	4400	1910	1254	3962	2168	1216	1459	824	1751	737	2042	
7.5	7.4	4.9	6.9	7.8	7.4	7.3	7.6	8.0	7.8	7_3	7.2	7.3	8.0	8.0	7.2	7.4	7.6	7.9	8.0	99 . 4	6.9	7.7	7.9	7.4	7.3	7.5	7.1	7.7	7.1	7.1	
3.1	3.7	3.2	3.7	2.6	5.8	5.5	8.9	8.8	2.7	6.4	9.8	:	3.00	:3 60	5.1		2.6	3.5 \S	9.1	6.1	7.2	5.0	14.5	5.9	6.2	4.3	2.3	5.2	1.4	5.7	
3.8		4.4	4.7	2.7	6.	8 .	8.2	10.8	5.3	5.8	5.4	6.5	6.2	4.0	7.1	5.8	4.3	4.1	15.8	12.1	7.5	3.1	19.9		4.2	4.3	3.0	S. 8	2.1	7.4	
13.6	15.1	12.1	14.8	12.8	3 3 3	36.8	22.1	25.3	49.6	1	14.6	17.3	27.6	13,8	23.5	24.8	19.1	25.7	44.6	\$7.7	19.1	13.5	35.9	23.4	11.2	16.3		19.8	9.3	22.1	
0.3	0.4	0.3	0.3	0.2	0.4	0.5	0.4	0.5	0.6	0.3	0.3	0.4	0.3	0.4	0.5	0.5	0.5	0.3	0.6	0.7	0.3	0.3	0.3	0.6	0.4	0	0.3	0.4	0.3	0.6	
 8	1.2	0.7	0.8	4.3	2.7	4.6	4.2	3.5	6.2	4 .5	3.6	2.9	.	2.6	3.6	4.2	2.9	3.9	2.	3.0	J. 3	1.1	2.7	2.6	6.0	3.6	3.2	4.5	2.8	3.6	
:	6. 8	7.2		1.4	1.2	1	3.2	4.7	6.2	2.0	2.0	2.4	5.5	1.3	6.1	4.6	5.2	2.3	6.0	6.8	2.5	1.7	2.6	1.3	1.0	. .	0.9	2.5	0.	5.7	
14.8	16.3	12.3	15.3	12.8	27.9	42.0	32.5	37.9	45.8	21.1	21.1	23.6	28.3	16.3	25.6	26.9	18.8	27.7.	61.4	65.7	28.0	19.2	8.7	35.2	15.2	20.7	10.8	23.8	9.5	26.1	
65.4	63.2	60.4	62.8	69.	59.4	71.9	55.7	55.8	85.3	54.1	\$5.6	60.6	72.9	65.8	2 2	70.5	72.0	76.4	63.7	75.4	\$5.9	61.6	SO.5	61.2	51.0	2	59.6	63.4	70.0	61.7	

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