

Fig. 9-21 Energy Generation of Olur Project

Table 9-21 Preliminary Cost Estimation of Olur Project

	Unit: Million TL
Relocation Road	14,000
Camp Facilities	5,000
Land Acquisition	51,752
Civil Work	295,542
Diversion	8,496
Care of River	9,857
Dam	107,510
Spillway	38,011
Outlet Works	4,356
Intake	3,216
Headrace Tunnel	108,479
Surge Tank	5,429
Penstock	2,617
Power House	6,800
Switchyard	771
Hydraulic Equipment	30,051
Electro-Mechanical Equipment	75,564
Contingency	36,735
Engineering and Administration Cost	42,016
Total	<u>550,660</u>
Interest During Construction	120,713
Grand Total	<u>671,373</u>
	(156.1 Million \$)

Table 9-22 Summary of Operation Study on Ayvali Reservoir

Unit: 10⁶ m³

YEAR	INFLOW	POWER DISCHARGE	SPIII
1940	1082.08	1115.76	0.39
1941	1317.81	1256.24	31.60
1942	773.55	775.63	0.0
1943	1051.77	1028.94	0.0
1944	848.43	849.02	0.0
1945	1086.98	1080.58	0.0
1946	805.47	797.09	0.0
1947	1007.92	988.59	0.0
1948	810.01	821.36	0.15
1949	890.59	879.95	0.0
1950	921.85	925.14	0.0
1951	1183.25	1153.51	0.0
1952	996.32	991.03	0.0
1953	1260.45	1248.49	0.0
1954	708.58	730.59	0.0
1955	688.24	663.13	0.0
1956	849.97	843.32	0.29
1957	706.29	699.60	0.0
1958	827.92	819.66	0.0
1959	982.02	961.99	0.00
1960	438.50	491.39	0.0
1961	479.24	518.37	0.0
1962	1183.65	1074.61	10.92
1963	1095.44	1096.85	0.0
1964	675.70	667.98	0.0
1965	720.90	723.24	0.0
1966	712.94	689.60	0.0
1967	1467.16	1216.24	250.83
1968	891.76	894.50	0.0
1969	522.21	524.13	0.0
1970	551.57	540.16	0.0
1971	672.37	661.15	0.0
1972	695.18	684.07	0.50
1973	518.94	527.82	0.0
1974	462.12	518.21	0.0
1975	800.14	709.47	0.0
1976	736.88	746.07	0.68
1977	790.94	786.78	0.31
1978	666.49	646.74	0.0
1979	727.95	746.12	0.42
1980	602.56	581.63	0.30
1981	604.96	580.07	0.00
1982	435.09	518.14	0.0
1983	662.15	574.28	0.0
1984	569.04	581.53	0.0
1985	665.10	642.17	1.04
1986	922.51	908.57	0.04
1987	992.15	978.57	0.0
1988	532.92	536.43	0.0
1989	647.93	594.78	0.06
AVERAGE	804.88	792.18	5.95

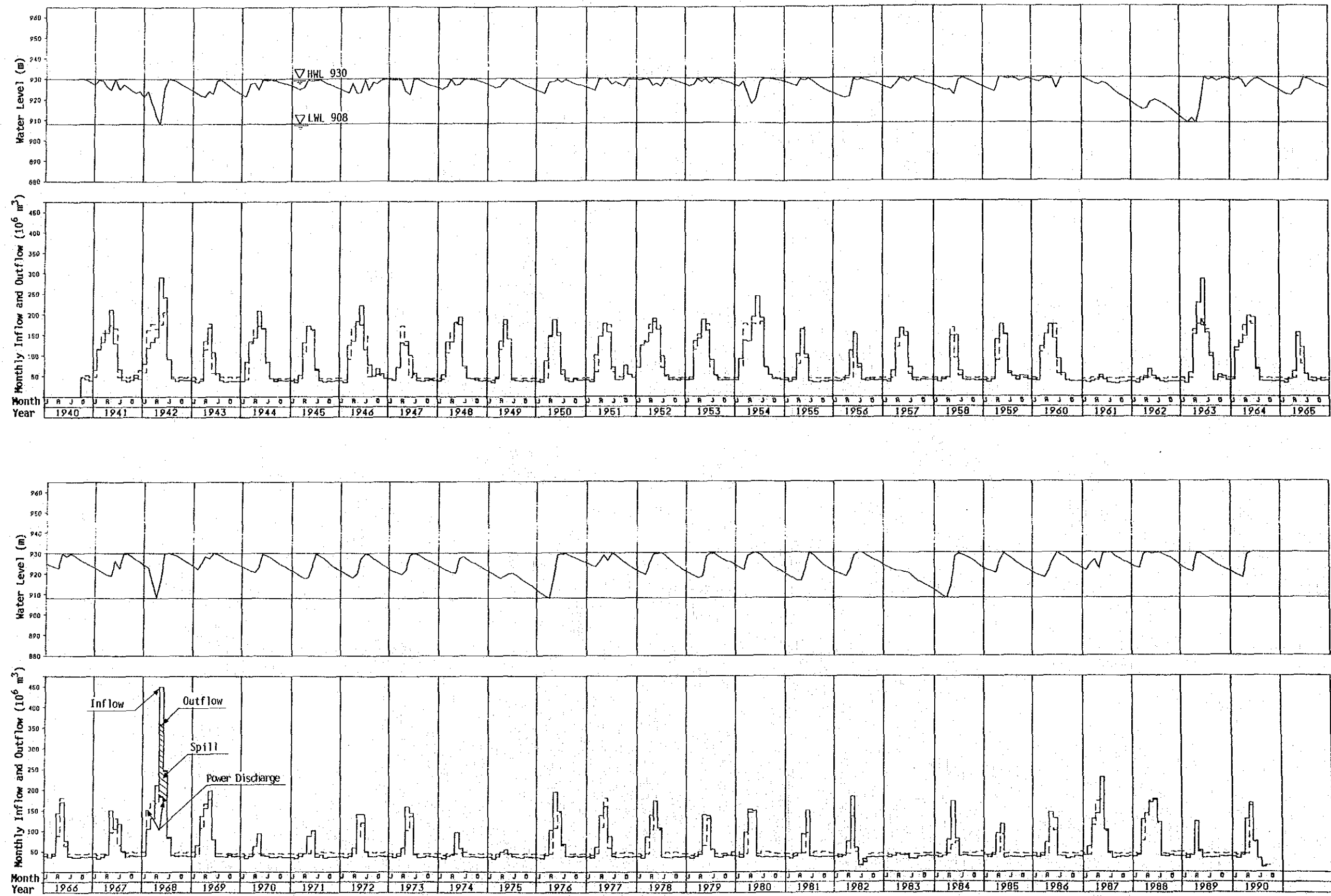


Fig. 9-22 Ayvalı Reservoir Operation

Table 9-23 Total Energy Generation of Ayvali Project

Unit: GWh

NO.	Year	Month	< OCT >	< NOV >	< DEC >	< JAN >	< FEB >	< MAR >	< APR >	< MAY >	< JUN >	< JUL >	< AUG >	< SEP >	< TOTAL >
1	1940		24.99	26.87	24.99	24.99	62.14	82.55	87.26	91.71	85.97	24.99	24.99	24.18	585.59
2	1941		24.99	24.18	33.31	30.54	83.11	87.33	79.47	86.84	90.00	47.19	24.99	24.18	636.13
3	1942		24.99	24.99	24.99	22.99	22.57	22.57	59.10	91.61	29.55	27.76	24.99	24.18	401.22
4	1943		24.99	24.18	24.99	22.76	21.78	48.26	87.56	90.83	85.97	41.64	22.70	21.99	537.64
5	1944		22.83	22.22	23.12	23.32	21.10	23.37	55.01	89.31	83.28	33.31	22.78	22.18	441.83
6	1945		23.10	22.55	23.38	23.38	20.95	47.84	89.01	93.00	90.00	77.73	24.99	26.87	562.66
7	1946		27.76	26.87	23.10	23.34	21.26	36.65	88.66	69.40	24.18	27.76	22.95	22.38	414.31
8	1947		23.31	22.67	23.46	23.54	21.90	56.98	82.78	91.61	90.00	36.09	22.85	22.30	517.32
9	1948		23.21	22.64	23.62	23.54	21.18	23.42	60.39	91.69	72.54	22.77	22.89	22.32	430.21
10	1949		23.26	22.65	23.51	23.51	20.88	23.47	77.23	91.75	80.60	22.76	22.72	22.15	458.89
11	1950		23.06	22.44	23.37	23.36	21.03	31.17	77.77	91.77	90.00	30.54	24.99	24.18	483.66
12	1951		24.99	26.87	23.12	36.42	62.99	69.49	90.00	93.00	88.66	36.09	24.99	22.23	599.03
13	1952		23.17	22.64	23.59	23.52	21.24	59.89	80.30	91.80	90.00	36.09	24.99	22.14	519.37
14	1953		23.01	22.45	23.58	23.48	21.42	92.91	87.58	86.71	90.00	93.00	36.09	26.87	643.87
15	1954		22.99	22.47	23.45	23.66	21.29	23.49	41.28	88.18	48.36	22.83	23.02	22.48	383.50
16	1955		23.25	22.49	23.13	23.02	21.42	22.81	22.07	23.36	81.13	36.09	22.80	22.24	343.91
17	1956		23.21	22.71	23.49	23.40	21.06	23.41	63.44	86.88	80.60	27.76	22.80	22.21	440.77
18	1957		23.16	22.63	23.49	23.41	21.07	21.25	22.48	86.06	88.36	24.99	22.84	22.25	363.99
19	1958		23.17	22.61	23.44	23.35	21.01	23.18	46.51	89.40	77.91	27.76	24.99	24.18	427.50
20	1959		22.92	22.59	23.32	23.37	22.14	57.10	80.41	91.08	90.00	27.76	25.69	19.52	505.69
21	1960		18.72	18.74	19.71	22.22	21.44	23.59	22.55	22.82	21.24	21.71	21.89	21.56	256.16
22	1961		22.67	21.56	22.57	22.40	20.07	22.05	21.16	21.16	20.19	20.75	20.77	20.32	256.08
23	1962		21.34	20.94	21.69	21.48	19.22	21.33	77.37	85.34	90.00	83.28	49.97	24.18	536.15
24	1963		22.92	22.48	23.46	23.67	23.17	71.17	89.64	93.00	90.00	33.31	22.71	22.07	573.32
25	1964		22.93	22.35	23.20	23.17	20.81	22.95	22.14	74.96	40.30	22.76	22.74	22.12	345.40
26	1965		22.97	22.30	23.13	23.21	20.95	23.11	45.98	93.00	32.24	22.69	22.65	21.97	374.18
27	1966		22.84	22.30	23.16	23.03	20.67	22.73	21.91	49.97	67.14	33.31	24.99	24.18	356.25
28	1967		24.99	24.18	24.99	24.99	78.10	84.45	81.49	93.00	90.00	44.42	22.66	24.18	618.43
29	1968		24.99	24.18	24.99	24.99	22.57	55.91	88.14	92.31	40.30	22.68	22.54	21.81	465.39
30	1969		22.54	21.85	22.71	22.88	20.77	22.83	21.61	22.33	21.94	22.53	22.50	21.83	266.40
31	1970		22.63	21.98	22.83	22.85	20.49	22.53	21.69	21.91	26.87	22.70	24.99	24.18	275.62
32	1971		24.99	24.18	22.72	22.84	21.23	22.55	21.55	44.42	61.79	24.99	24.99	24.18	340.41
33	1972		24.99	24.18	24.99	22.85	20.63	22.74	22.07	52.75	69.85	22.70	22.70	22.09	352.53
34	1973		22.44	22.11	23.27	23.15	20.78	22.88	22.07	22.62	24.18	24.34	21.84	21.56	271.90
35	1974		22.45	22.12	22.96	22.82	20.47	22.49	21.57	21.37	20.35	20.85	20.89	20.47	253.80
36	1975		21.51	21.13	21.81	21.60	20.02	21.22	21.10	55.52	75.22	33.31	22.74	22.06	357.25
37	1976		22.86	22.20	23.11	23.29	20.95	23.26	57.53	93.00	29.55	22.66	24.99	24.18	387.57
38	1977		24.99	24.18	22.82	22.85	20.53	23.02	54.90	89.29	53.73	22.75	24.99	24.18	408.22
39	1978		24.99	24.18	22.74	22.77	20.45	22.51	21.73	33.31	67.14	27.76	22.70	22.03	332.33
40	1979		22.90	22.11	24.99	24.99	21.34	23.36	77.55	75.25	21.97	24.99	24.99	24.18	388.60
41	1980		22.50	21.87	22.71	22.65	20.32	22.34	21.55	22.18	48.36	22.68	24.99	24.18	296.32
42	1981		24.99	24.18	22.67	22.80	20.49	22.57	22.04	69.40	24.30	7.27	15.84	21.65	298.40
43	1982		22.84	22.48	23.30	23.17	20.82	22.91	21.93	22.48	20.71	20.77	20.74	20.34	262.48
44	1983		21.41	20.85	21.48	21.46	20.04	21.24	20.86	33.04	37.61	22.70	22.57	21.75	285.01
45	1984		22.47	24.18	24.99	22.19	20.14	22.35	24.18	47.19	21.88	22.63	22.89	22.39	297.48
46	1985		23.15	22.77	22.88	22.73	20.43	22.51	21.95	55.52	51.04	22.75	22.96	22.44	330.63
47	1986		23.13	22.38	24.99	22.71	20.89	59.10	89.32	91.69	51.04	22.91	23.11	22.47	473.54
48	1987		23.31	22.58	23.27	23.15	21.59	39.47	72.17	91.81	90.00	81.07	22.75	21.94	513.13
49	1988		24.99	24.18	24.99	24.99	19.92	22.16	29.55	24.99	22.06	23.07	23.39	22.53	286.80
50	1989		23.15	22.29	22.91	22.77	20.43	22.50	22.42	86.04	37.47	15.96	5.96	7.36	309.26
TOTAL			1170.56	1145.63	1178.05	1180.95	1304.97	1772.87	2558.01	3417.47	2885.83	1551.19	1184.39	1121.19	20471.06
AVE			23.41	22.91	23.56	23.62	26.10	35.46	51.16	68.35	57.72	31.02	23.69	22.42	409.42
MAX			27.76	26.87	33.31	36.42	83.11	92.91	90.00	93.00	90.00	93.00	49.97	26.87	643.87
MIN			18.72	18.74	19.71	21.46	19.22	21.22	20.86	21.16	20.19	7.27	5.96	7.36	256.08

Table 9-24 Firm Energy Generation of Ayvali Project

Unit: GWh

NO.	Year	Month	< OCT >	< NOV >	< DEC >	< JAN >	< FEB >	< MAR >	< APR >	< MAY >	< JUN >	< JUL >	< AUG >	< SEP >	< TOTAL >
1	1940		23.25	22.50	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	23.25	22.50	273.75
2	1941		23.25	22.50	23.25	23.25	20.78	21.85	20.24	21.72	22.50	23.25	23.25	22.50	268.34
3	1942		23.25	22.50	23.25	23.25	21.00	22.19	22.50	23.25	22.50	23.25	23.25	22.50	272.69
4	1943		23.25	22.50	23.25	23.25	21.75	23.25	22.50	23.25	22.50	23.25	22.70	21.99	272.96
5	1944		22.83	22.22	23.12	23.25	21.00	23.25	22.50	23.25	22.50	23.25	22.78	22.18	272.11
6	1945		23.10	22.50	23.25	23.25	20.95	23.25	22.50	23.25	22.50	23.25	23.25	22.50	273.54
7	1946		23.25	22.50	23.10	23.25	21.00	23.25	22.50	23.25	22.50	23.25	22.95	22.58	273.19
8	1947		23.25	22.50	23.25	23.25	21.75	23.25	22.50	23.25	22.50	23.25	22.85	22.50	273.90
9	1948		23.25	22.50	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	22.77	22.32	272.70
10	1949		23.25	22.50	23.25	23.25	20.88	23.25	22.50	23.25	22.50	23.25	22.72	22.15	272.70
11	1950		23.25	22.44	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	23.25	22.50	273.50
12	1951		23.25	22.50	23.12	23.25	21.75	23.25	22.50	23.25	22.50	23.25	23.25	22.23	274.10
13	1952		23.17	22.50	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	23.25	22.14	273.31
14	1953		23.01	22.43	23.25	23.25	21.00	23.25	21.00	21.69	22.50	23.25	23.25	22.50	271.28
15	1954		22.99	22.47	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	23.02	22.48	272.79
16	1955		23.25	22.49	23.13	23.25	21.42	23.25	22.50	23.25	22.50	23.25	22.80	22.24	272.23
17	1956		23.21	22.50	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	22.84	22.21	272.97
18	1957		23.16	22.50	23.25	23.25	21.00	23.25	22.48	23.25	22.50	23.25	22.84	22.25	272.98
19	1958		23.17	22.50	23.25	23.25	21.00	23.18	22.50	23.25	22.50	23.25	23.25	22.50	273.59
20	1959		22.92	22.39	23.25	23.25	21.75	23.25	22.50	23.25	22.50	23.25	23.25	19.52	271.08
21	1960		18.72	18.74	19.71	22.22	21.00	23.25	22.50	22.82	21.24	21.71	21.89	21.56	255.35
22	1961		22.67	21.96	22.54	22.50	19.91	21.84	21.11	20.99	20.12	20.75	20.77	20.26	255.22
23	1962		21.01	21.29	21.29	23.25	18.73	20.79	20.12	21.35	22.50	23.25	23.25	22.50	256.50
24	1963		22.92	22.48	23.25	23.25	21.75	23.25	22.50	23.25	22.50	23.25	22.71	22.07	273.18
25	1964		22.93	22.33	23.20	23.17	20.81	22.95	22.44	23.25	22.50	23.25	22.74	22.12	271.39
26	1965		22.97	22.50	23.13	23.21	20.95	23.11	22.50	23.25	22.50	23.25	22.65	21.97	271.21
27	1966		22.44	22.30	23.16	23.03	20.67	22.73	21.90	23.25	22.50	23.25	23.25	22.50	271.38
28	1967		23.25	22.50	23.25	23.25	21.21	21.38	20.70	23.25	22.50	23.25	22.66	22.50	269.70
29	1968		23.25	22.50	23.25	23.25	21.00	23.25	22.50	23.25	22.50	23.25	22.54	21.81	271.78
30	1969		22.44	21.85	22.71	22.88	20.77	22.88	21.48	22.33	21.96	22.68	22.53	21.83	266.28
31	1970		22.63	21.98	22.83	22.85	20.49	22.47	21.60	21.60	22.50	22.70	23.25	22.50	267.39
32	1971		23.25	22.50	22.72	22.84	21.23	22.51	21.59	23.25	22.50	23.25	23.25	22.50	271.19
33	1972		23.25	22.50	23.25	22.85	20.63	22.74	22.07	23.25	22.50	23.25	22.70	22.09	270.53
34	1973		23.25	22.41	23.25	23.15	20.78	22.88	22.07	22.62	22.50	23.25	21.84	21.36	269.11
35	1974		22.45	22.12	22.96	22.82	20.47	22.45	21.57	21.31	20.35	20.85	20.89	20.47	258.71
36	1975		21.24	20.86	21.47	21.18	19.55	20.68	20.76	23.25	22.50	23.25	22.74	22.06	259.64
37	1976		22.46	22.20	23.11	22.85	20.95	23.25	22.50	23.25	22.50	23.25	23.25	22.50	272.28
38	1977		23.25	22.50	22.82	22.85	20.53	23.02	22.50	23.25	22.50	23.25	23.25	22.50	271.72
39	1978		23.25	22.50	22.74	22.77	20.45	22.45	21.65	23.25	22.50	23.25	22.70	22.03	269.54
40	1979		22.00	22.11	23.25	23.25	21.34	23.25	22.50	23.25	21.97	23.25	23.25	22.50	272.82
41	1980		22.00	21.85	22.71	22.65	20.26	22.21	21.39	21.99	22.50	22.68	23.25	22.50	266.50
42	1981		23.25	22.50	22.67	22.60	20.49	22.54	22.04	23.25	22.50	22.68	23.25	21.65	246.80
43	1982		22.84	22.48	23.25	23.17	20.82	22.91	21.93	22.68	20.71	20.77	20.73	20.27	262.36
44	1983		21.20	20.48	21.00	20.96	19.58	20.70	20.43	22.68	21.88	22.70	22.57	21.75	256.86
45	1984		22.47	22.50	23.25	23.25	20.00	22.25	22.50	23.25	21.88	22.63	22.89	22.39	268.00
46	1985		23.15	22.27	22.88	22.73	20.42	22.45	21.95	23.25	22.50	22.75	22.96	22.44	269.75
47	1986		23.13	22.18	23.25	22.71	20.84	23.25	22.50	23.25	22.50	22.91	23.11	22.47	272.15
48	1987		23.25	22.50	23.25	23.15	21.59	23.25	22.50	23.25	22.50	23.25	22.75	21.94	273.18
49	1988		23.25	22.50	23.25	23.25	19.70	21.96	22.50	23.25	22.06	23.07	23.25	22.50	270.54
50	1989		23.25	22.29	22.91	22.77	20.42	22.44	22.42	23.25	22.50	15.96	5.96	7.36	231.44
TOTAL			1142.64	1110.21	1146.79	1145.80	1040.40	1137.10	1102.92	1147.40	1115.27	1122.58	1116.22	1088.79	13416.12
AVE			22.85	22.20	22.94	22.92	20.81	22.74	22.06	22.95	22.31	22.45	22.45	21.78	268.32
MAX			23.25	22.50	23.25	23.25	21.75	23.25	22.50	23.25	22.50	23.25	23.25	22.50	274.10
MIN			18.72	18.74	19.71	20.98	18.73	20.68	20.12	20.99	20.12	7.27	5.96	7.36	231.44

Table 9-25 Monthly Peak Power of Ayvali Project

Unit: MW

NO. Year	Month	< OCT >	< NOV >	< DEC >	< JAN >	< FEB >	< MAR >	< APR >	< MAY >	< JUN >	< JUL >	< AUG >	< SEP >	< TOTAL >
1	1940	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1500.00
2	1941	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1470.40
3	1942	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1494.28
4	1943	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1491.62
5	1944	122.73	123.43	124.28	125.00	125.00	125.00	125.00	125.00	125.00	125.00	122.46	123.20	1491.09
6	1945	124.18	125.00	125.00	125.00	124.67	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1498.86
7	1946	125.00	125.00	124.21	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.38	124.36	1496.95
8	1947	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.87	123.87	1496.74
9	1948	124.80	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	122.43	123.06	124.02	1494.31
10	1949	125.00	125.00	125.00	125.00	124.26	125.00	125.00	125.00	125.00	125.00	122.14	123.05	1494.25
11	1950	123.96	124.67	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1498.62
12	1951	125.00	125.00	124.29	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	123.50	1497.79
13	1952	124.58	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	123.01	1497.59
14	1953	123.70	124.60	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1486.60
15	1954	123.61	124.82	125.00	125.00	125.00	125.00	125.00	125.00	125.00	122.73	123.76	124.90	1494.83
16	1955	125.00	124.92	124.36	125.00	123.13	122.62	125.00	125.00	125.00	125.00	122.58	123.56	1487.54
17	1956	124.79	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	122.37	123.38	1495.74
18	1957	124.51	125.00	125.00	125.00	125.00	124.41	125.00	125.00	125.00	125.00	122.80	123.61	1495.81
19	1958	124.56	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1499.16
20	1959	123.24	124.38	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	108.42	1481.04
21	1960	100.62	104.11	105.99	119.44	125.00	125.00	125.00	125.00	118.02	116.70	117.67	119.75	1400.02
22	1961	121.86	121.98	121.20	119.87	118.53	117.42	117.28	112.87	111.75	111.56	111.68	112.57	1398.57
23	1962	113.45	114.45	114.47	112.95	111.47	111.78	111.78	114.81	125.00	125.00	125.00	125.00	1405.17
24	1963	123.23	124.89	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	122.11	122.59	1492.81
25	1964	123.27	124.07	124.72	124.57	123.90	123.38	122.99	125.00	125.00	125.00	122.24	122.86	1487.02
26	1965	123.47	123.87	124.35	124.76	124.68	124.44	125.00	125.00	125.00	121.98	121.76	122.06	1486.17
27	1966	122.81	123.90	124.52	125.00	123.01	122.21	121.67	125.00	125.00	125.00	125.00	125.00	1486.92
28	1967	125.00	125.00	125.00	125.00	121.89	114.95	115.00	125.00	125.00	125.00	121.95	125.00	1473.88
29	1968	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	121.93	121.17	121.18	1459.28
30	1969	121.16	121.42	123.12	123.02	123.65	123.02	119.36	120.07	125.00	121.11	120.97	121.28	1459.17
31	1970	121.67	122.12	122.73	123.83	121.94	120.83	119.98	116.15	125.00	125.00	125.00	125.00	1465.27
32	1971	125.00	125.00	122.15	122.79	122.00	121.02	118.84	125.00	125.00	125.00	125.00	125.00	1481.81
33	1972	125.00	125.00	125.00	125.00	122.82	122.26	122.62	125.00	125.00	122.02	122.06	122.74	1482.35
34	1973	123.75	124.50	125.00	124.44	123.71	123.00	122.59	121.59	125.00	125.00	117.40	118.66	1474.63
35	1974	120.70	122.86	123.45	123.67	121.85	120.72	119.81	114.56	113.05	113.11	112.33	113.74	1417.86
36	1975	114.75	115.89	115.41	115.85	112.33	111.17	115.36	125.00	125.00	125.00	122.28	122.56	1418.59
37	1976	122.92	123.36	124.27	125.00	124.68	125.00	125.00	125.00	125.00	121.81	125.00	125.00	1492.03
38	1977	125.00	125.00	123.70	123.84	122.18	123.78	125.00	125.00	125.00	122.32	125.00	125.00	1488.82
39	1978	125.00	125.00	122.27	123.44	121.71	120.73	120.28	125.00	125.00	125.00	122.02	122.39	1476.83
40	1979	123.10	122.83	123.00	125.00	122.63	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1490.63
41	1980	120.97	121.38	123.10	121.77	120.59	119.43	118.85	118.23	125.00	121.95	125.00	125.00	1460.27
42	1981	125.00	125.00	121.89	123.57	121.98	121.21	122.45	125.00	125.00	39.07	85.14	120.30	1354.60
43	1982	122.81	124.86	125.00	124.56	123.93	123.18	121.84	120.88	115.03	111.66	111.47	112.64	1337.84
44	1983	113.93	113.76	112.93	124.80	112.52	111.31	113.50	123.56	125.00	122.02	121.32	120.81	1403.47
45	1984	120.82	125.00	125.00	118.30	119.07	119.52	125.00	125.00	121.55	121.65	123.07	124.41	1468.39
46	1985	124.46	123.72	122.99	123.23	121.55	120.68	121.97	125.00	125.00	122.31	123.43	124.66	1478.00
47	1986	124.37	123.25	125.00	122.12	124.32	125.00	125.00	125.00	125.00	123.19	124.23	124.82	1491.29
48	1987	125.00	125.00	125.00	124.46	124.06	125.00	125.00	125.00	125.00	125.00	122.35	121.91	1492.76
49	1988	125.00	125.00	125.00	125.00	117.29	118.05	125.00	125.00	124.53	124.03	125.00	125.00	1481.90
50	1989	124.47	123.86	123.18	122.42	121.55	120.62	124.58	125.00	125.00	85.81	32.06	40.87	1369.41
TOTAL		6143.23	6167.87	6165.57	6160.23	6140.57	6113.44	6127.37	6168.84	6195.98	6035.37	6001.21	6048.83	73468.50
AVE		122.86	123.36	123.31	123.20	122.81	122.27	122.55	123.38	123.92	120.71	120.02	120.98	1469.37
MAX		125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	125.00	1500.00
MIN		100.62	104.11	105.99	112.80	111.47	111.17	111.78	112.87	111.75	39.07	32.06	40.87	1269.41

Table 9-26 Peak Power Duration of Ayyali Project

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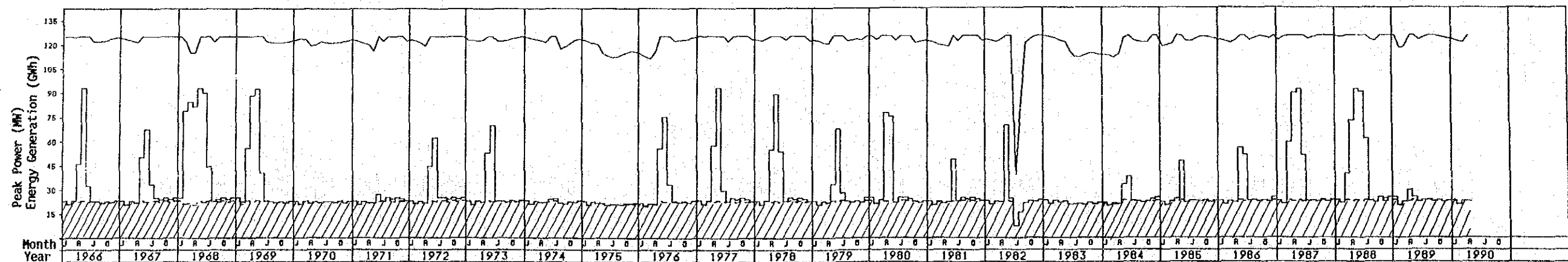
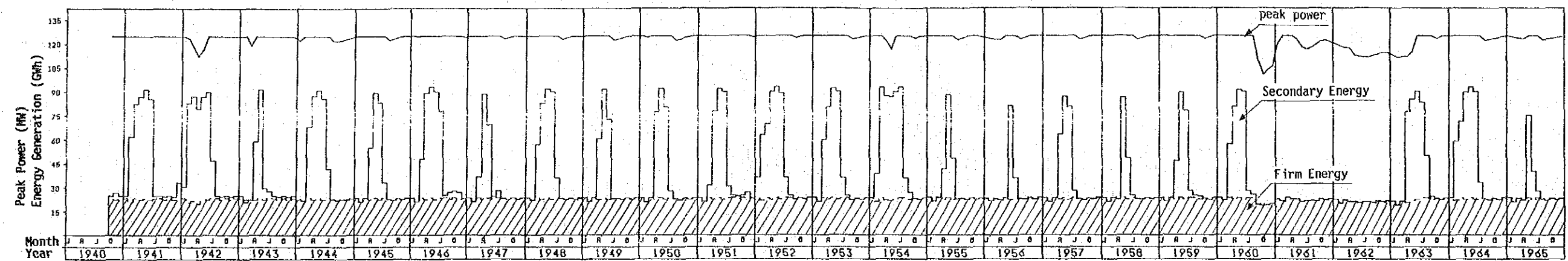
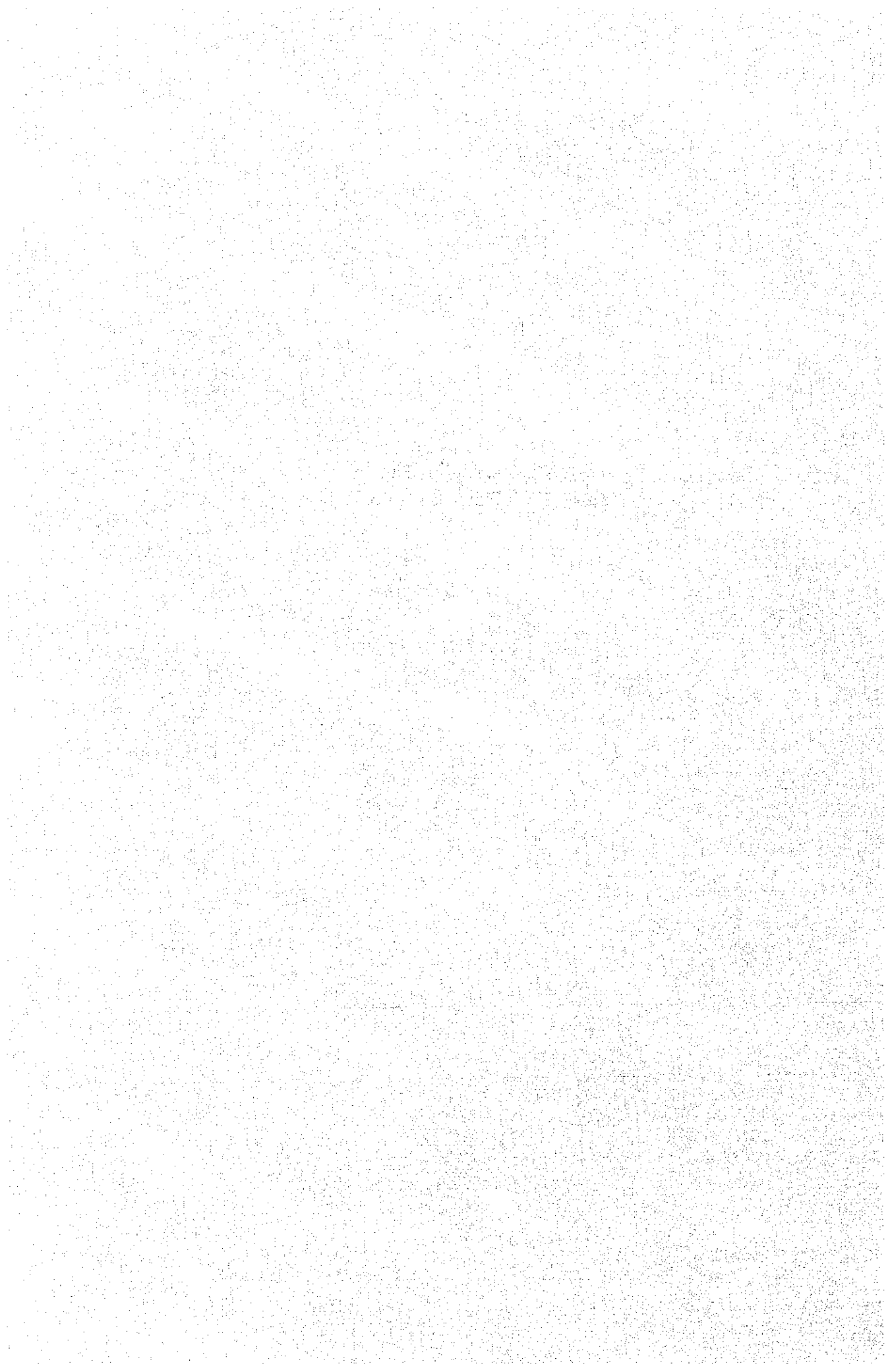


Fig. 9-23 Energy Generation of Ayvali Project

Table 9-27 Preliminary Cost Estimation of Ayvalı Project

	Unit: Million TL
Relocation Road	28,000
Camp Facilities	5,000
Land Acquisition	34,106
Civil Work	486,819
Diversión	9,742
Care of River	14,147
Dam	242,823
Spillway	63,358
Outlet Works	4,748
Intake	3,238
Penstock	2,333
Power House	21,340
Tailrace	123,914
Switchyard	1,176
Hydraulic Equipment	24,883
Electro-Mechanical Equipment	96,516
Contingency	58,052
Engineering and Administration Cost	64,122
Total	<u>797,497</u>
Interest During Construction	190,508
Grand Total	<u>988,005</u>
	(230 Million \$)

Chapter 10 TRANSMISSION SYSTEM PLANNING AND SYSTEM STABILITY



Chapter 10

TRANSMISSION SYSTEM PLANNING AND SYSTEM STABILITY

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Chapter 10 TRANSMISSION SYSTEM PLANNING AND SYSTEM STABILITY

10.1 TEK's Transmission System and Oltu Project

Turkish Electricity industry has been mainly managed and operated by Turkish Electricity Authority (TEK). The power system in Turkey consists of 380 kV for main grid to be connected between big power plants and load centers, and 154 kV for subsidiary grids to be connected among 380 kV grid, medium/small scale power plants and local load centers.

Power Development Program covers only the detail power generation plants by 1995, but the power generation program after 1995 is not clear enough for the system planning. Fortunately, TEK has a transmission system plan idea for 380 kV system plan of 2010 (see Fig. 10-1), and then we discussed far future based on the TEK's idea. We had considered that the power produced by the Oltu Project be used for some load centers located in the eastern region, but we found that the load in eastern regions is not so prospected since the power in the region will be supplied by local power plants through the existing and planned 154 kV transmission lines. Therefore, the power produced by the Oltu Project should be transmitted to the 380 kV main grid through Yusufeli Switching Station which is to be a part of Yusufeli Hydropower Project.

The transmission system of the River Hydroelectric Power Development Project is shown in Fig. 10-2.

10.2 Transmission Line Route Survey

In accordance with the original schedule, the survey for the 154 kV transmission line routes from Olur Hydropower Plant to the eastern region and to the Yusufeli Switching Station through Ayvali Hydropower Plant was performed. The route from Olur

Hydropower Plant to the eastern region was not selected for the transmission line in this Project because the power produced by this Project will be transmitted to the 380 kV main grid through Yusufeli Switching Station based on the meeting among EIE, TEK and JICA Mission. The route from Olur Hydropower Plant to Yusufeli Switching Station by 154 kV transmission line is technically feasible, however, the cost will be much more than the cost of ordinary transmission line due to the bad access to the location of towers in the steep/wild slopes. At the stage of detail design of the transmission lines, the exact location of the towers shall be decided based on the geological profiles of the route.

10.3 Site Survey of Yusufeli Switching Station

For sending the electric power produced by the Oltu Project to the 380 kV main grid, two circuits of incoming facilities for 154 kV transmission from the Ayvalı Hydropower Plant and a step-up transforming facilities from 154 kV to 380 kV in the Yusufeli Switching Station to be constructed as a part of the Yusufeli Hydropower Project are necessary. According to the site survey, it has been confirmed that the site is wide enough for installing the necessary facilities for the Oltu Hydropower Project. At the present, the Yusufeli Switching Station is designed as a conventional outdoor switching station, but the design may be changed to that of Gas-insulated Switching (GIS) in the future. If so, the space requirement for the facilities will be small, then the space for the facilities for Olur Project has no problem at all.

10.4 Transmission System Plan for Oltu Hydropower Project

10.4.1 Conditions for Planning

- (1) The electric power shall be transmitted to the demands area economically
- (2) The proposed plan shall be in accordance with TEK's power system plan
- (3) The transmission line design shall be in accordance with TEK's standard design criteria
- (4) The proposed system shall be stable even though system disturbance occurring on the system

10.4.2 Demand Areas and System Interconnecting Point

The electric power is to be sent to the demand area in whole Turkey through 380 kV, therefore, the Olur and Ayvalı Hydropower Plants are to be connected to the 380 kV system by 154 kV transmission lines at the Yusufeli Switching Station.

10.4.3 Transmission Line Voltage

The transmission voltage to be considered for this project may be 154 kV and 380 kV. Taking short distance from the power plants and the capacity of each power plant into consideration, 154 kV transmission lines are considered to be suitable for the Project. Judging from the transmission capacities, the number of circuits of each transmission line will be as follows:

- (1) Olur - Ayvalı 154 kV Transmission line: Single circuit
- (2) Ayvalı - Yusufeli 154 kV Transmission line: Double circuit

10.4.4 Distance and Conductor Size of Transmission Line

According to the survey on topographic maps, the transmission line distance of each transmission line are as follows:

- (1) Olur - Ayvalı 154 kV Transmission line: approx. 13 km
- (2) Ayvalı - Yusufeli 154 kV Transmission line: approx. 20 km

In accordance with the TEK's practice, the conductor of 154 kV transmission lines has been decided as follows:

- 636 MCM Aluminum Conductor Steel Reinforced (ACSR)

10.4.5 Interconnection with 380 kV Transmission System

The nearest terminal of 380 kV system from the Project is 380 kV Switching Station for Yusufeli Hydropower Plant to be completed before commissioning of the Oltu Project. The switchgear for the Oltu Project is of conventional outdoor type and to be installed in Yusufeli Switching Station since Yusufeli Switching Station is to be of a conventional outdoor switching station.

For stepping up the transmission voltage from 154 kV to 380 kV, it is necessary to install a step-up transformer at the Yusufeli Switching Station.

10.4.6 Power System Stability after Completion of the Project

Before completion of the Project, Yusufeli Hydropower Plant are to be completed and to be connected to the 380 kV transmission System. Yusufeli Power Plant has three units of 180 MW generators. During feasibility study of the Yusufeli Power Project, it has been confirmed that the 380 kV system is stable in even when the system disturbance occurred in the system during full load (540 MW) operation of Yusufeli Power Plant.

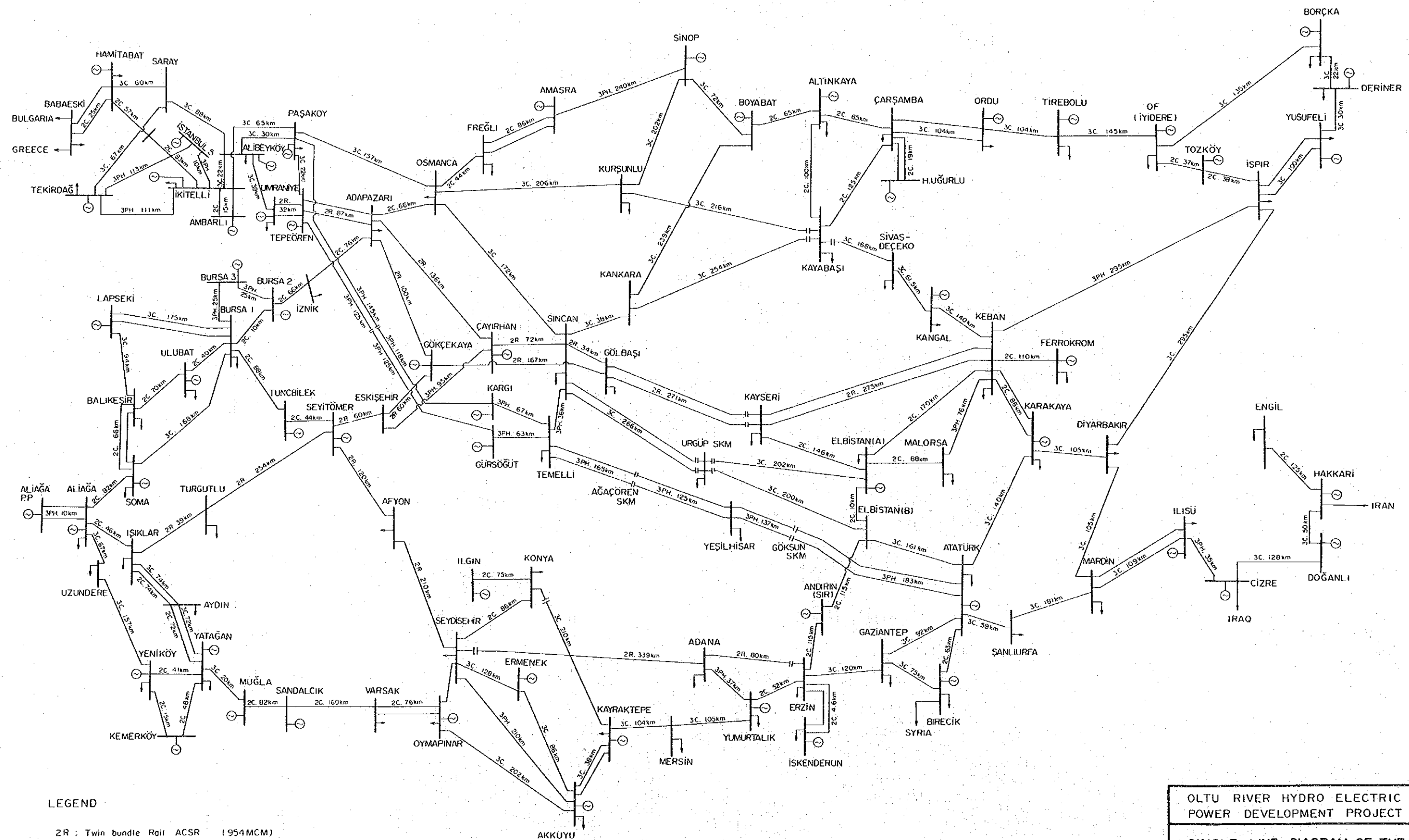
The power plants of the Oltu Project is to be connected to the 380 kV system at the Yusufeli Switching Station. Since the total capacity of two power plants (190 MW) is less than that of Yusufeli Power Plant, any disturbance on the 154 kV system of Oltu Project will not affect to 380 kV system seriously. Therefore, for the Oltu Project, the system analysis was made on the 154 kV system only.

10.4.7 Result of Power System Analysis

Short circuit current is the maximum at the time when three phase short circuit at the 154 kV bus of the Ayvalı Power Plant and the value is not more than 50 kA. Judging from the value, it is confirmed that the standard series of the circuit breakers is applicable for the Project.

For the steady state system stability, it should be confirmed that the Phase angle between the transmission line terminals, say sending end and receiving end, shall be within 60 degrees even if the line sending full power on the line. According to the analysis, the phase angle is within the criteria stated above and it is confirmed that the system is stable in the steady state operation.

For transient stability of the system, it shall be confirmed that the system is transiently stable under the most serious condition. The most serious condition will be the case that one circuit of the line between Ayvalı and Yusufeli Switchyard has three-phase ground fault near Ayvalı Power Plant during full load operation of two power plants and the fault be cleared by disconnecting the fault circuit by mean of the transmission line protective relaying system. As the result of the analysis, it is confirmed that the system is stable transiently and the both power plant can continue the operation.



LEGEND

- 2R : Twin bundle Rail ACSR (954MCM)
- 2C : Twin bundle Cardinal ACSR (954MCM)
- 3C : Triplet bundle Cardinal ACSR (954MCM)
- 3PH : Triplet bundle Pheasant ACSR (1272MCM)

OLTU RIVER HYDRO ELECTRIC
POWER DEVELOPMENT PROJECT

SINGLE-LINE DIAGRAM OF THE
380KV NETWORK
- LONG TERM EXPANSION
YEAR 2010

Fig. 10-1

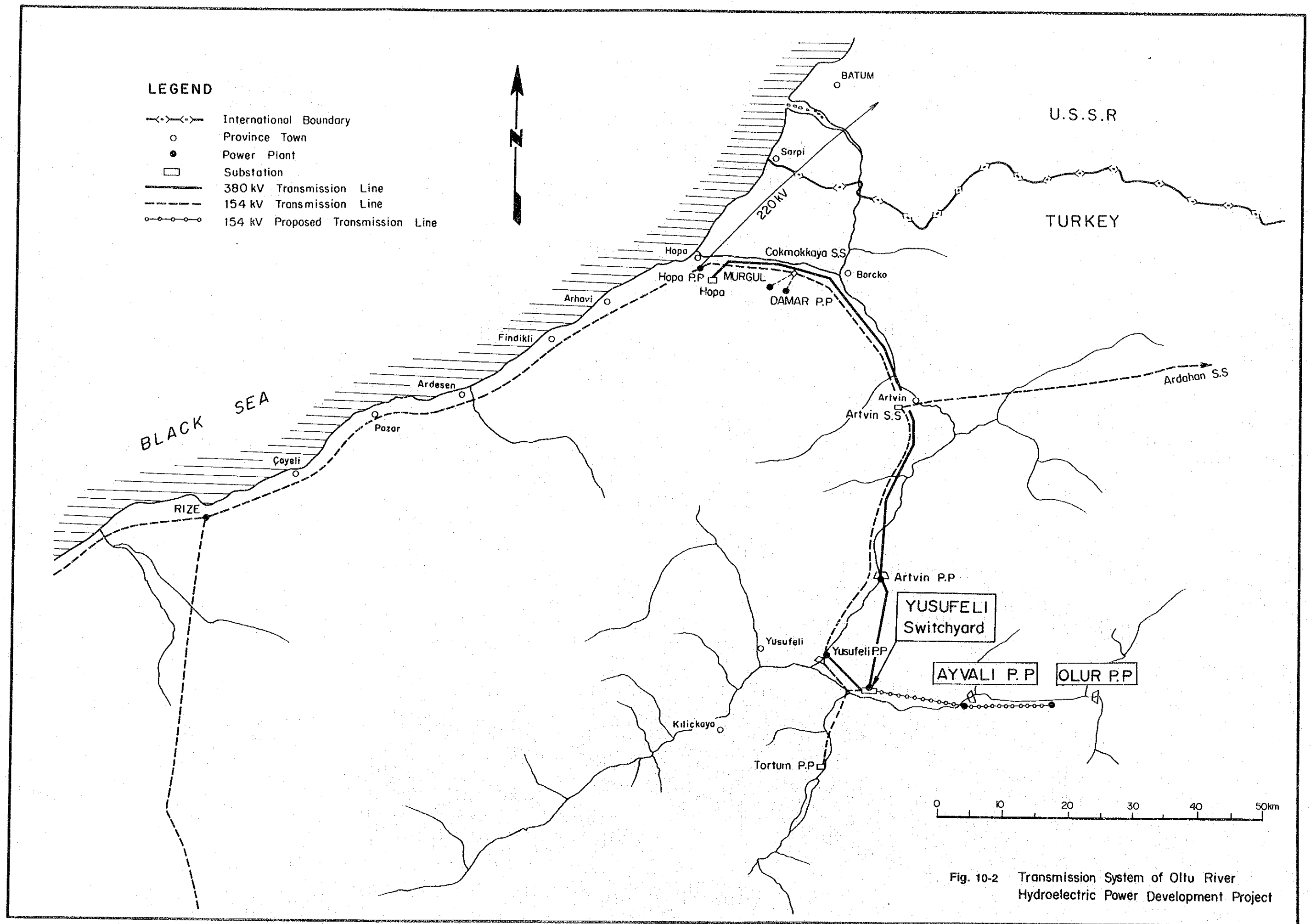
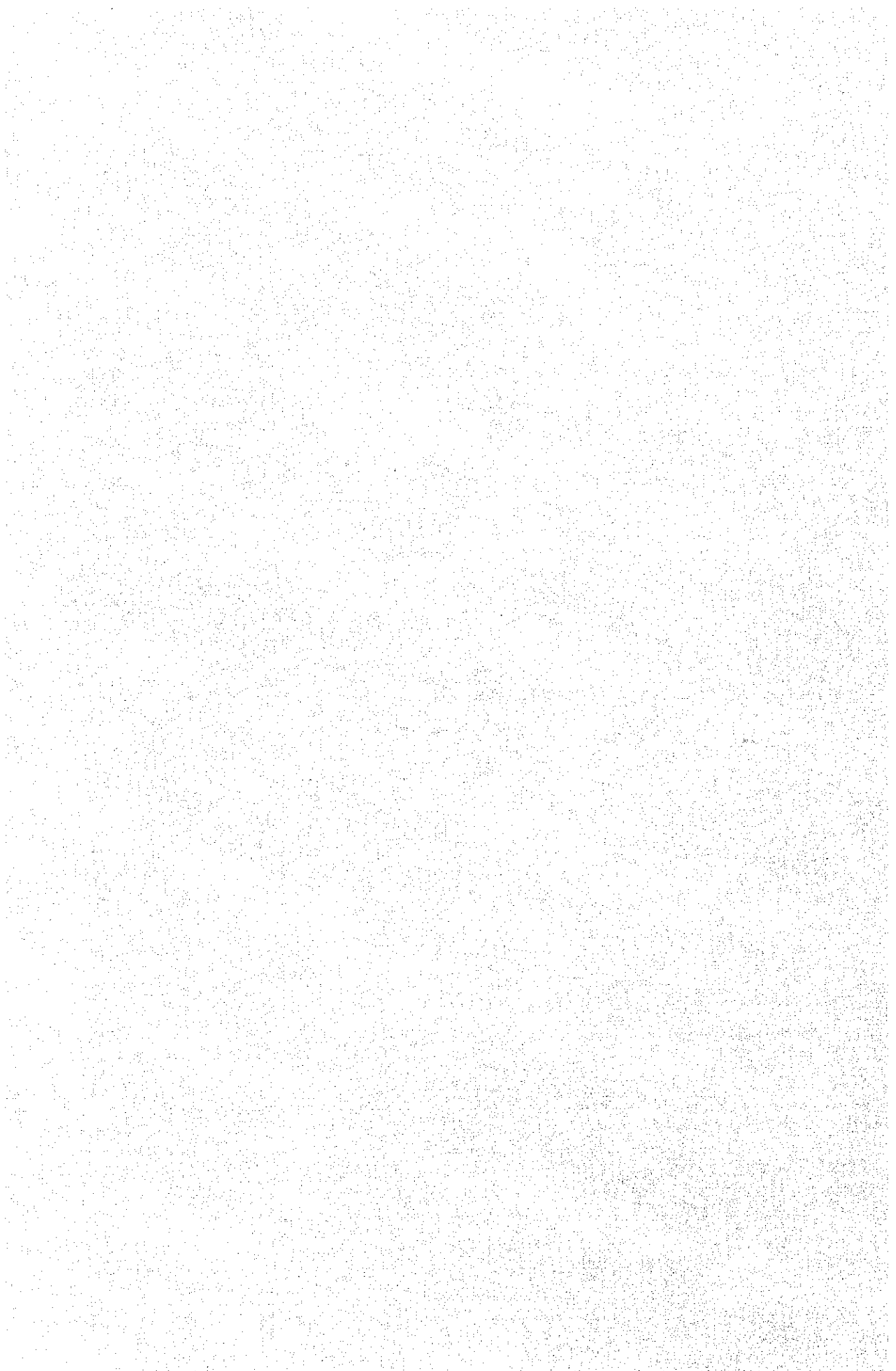


Fig. 10-2 Transmission System of Oltu River Hydroelectric Power Development Project

Chapter 11 FEASIBILITY DESIGN



Chapter 11

FEASIBILITY DESIGN

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Chapter 11 FEASIBILITY DESIGN

This Chapter describes the feasibility designs of civil structures, electro-mechanical equipments, transmission line, and temporary facilities for construction of the Olur and Ayvalı projects, respectively.

The results of studies of the designs of permanent facilities such as dam, spillway, intake, headrace tunnel, surge tank, penstock, powerhouse and tailrace tunnel which are situated as the main structures for the power plant are described including comparative designs of principal structures.

11.1 Olur Project

11.1.1 Dam and Auxiliary Structures

(1) Location and Outline

The projected damsite is located at the upstream part of the Oltu River, a tributary of the Çoruh River at the east side. Approximately 10 km downstream of the site for the Olur dam there is the Ayvalı dam which is included in this Oltu Project, and next downstream, the Yusufeli Project.

The topography at the site of Olur Dam is that of a V-shaped valley more or less symmetric right and left with slope gradients of about 30 to 35 deg. Regarding the dam axis, there is besides the present dam site a site of comparatively good quality as a dam axis from topographical and geological standpoints approximately 500 m downstream, and an investigation of this site had been carried out. Accordingly, an economic comparison was made of the two dam axes, upstream and downstream. As a result, the dam volume for the upstream proposal was $3.7 \times 10^3 \text{ m}^3$, and that for the

downstream proposal $5.3 \times 10^3 \text{ m}^3$, so that the downstream proposal will have a volume 40% larger than the upstream proposal. The lengths of the diversion tunnels would be roughly equal. On the other hand, the excavation volume for the spillway would be 16% larger for the downstream proposal compared with the upstream proposal. Further, the distance between the two dams would be 500 m and it was found there would be hardly any difference in the amount of water storage, and it was judged reasonable to select the upstream proposal for which the construction cost would be cheaper.

The riverbed at the damsite has sand-gravel layers deposited to a thickness of approximately 45 m. It has been confirmed that gravel of maximum size 30 cm and fine-particled sand are deposited in stratified form in these sand-gravel layers. The stratigraphical sequence consists of a sand-gravel layer of 25 m at the top, a sand layer of 25 m, and then a sand-gravel layer of 5 m to reach the foundation rock.

The ground surface of both abutments has almost no top soil or trees, and weathered rock is exposed at the whole surfaces. The depths of weathering of the foundation rocks at both abutments are about 3 to 8 m. The foundation rocks at both abutments and the river bed are hard and large faults have not been confirmed. Therefore, the topographical and geological conditions make it possible to construct even a concrete dam as well as a rockfill dam.

(2) Selection of Dam Type

Various types are conceivable for the dam, but generally, a fill dam and a concrete dam can be cited. The main factors in dam design are the two points that the dam body including the foundation is stable and that the construction cost is economical.

The quality and quantities of materials available in the vicinity of the damsite have a strong influences on the selection of the dam type, and these are important factors from the standpoint of the economics also. The results of investigations and tests of the geology and materials in the surroundings of the damsite were examined taking these into account for Olur Dam.

The great feature of the surroundings of the Olur damsite of the abundantly existing river-bed sand-gravel and talus deposits was focused on, and utilization of these was studied.

According to the site investigations, the gradation of the river deposit materials is fair. If the oversized material can be eliminated, it may be judged that 75% of the materials can be amply used as the concrete aggregates and the filter materials which are on the coarse side.

When looking upon the river-bed sand-gravel as material, both quality and quantity are adequate, and from the standpoint of gradation, although there would be a necessity for consideration to be given in design and construction depending on the use and the range of the fluctuation of the grain size, it is thought possible for the selection of dam type to be made over a broad scope.

Factors to be considered in selection of the dam type, as shown in Table 11-1, directly are topography, geology, meteorology, hydrology, seismicity, dam scale and availability of dam material quality, quantity and place. Indirectly, there are also the influences of conditions of purpose of construction, construction period, technological capability, labor force, construction machinery, etc. And, the fundamental matter is for the most economical type to be selected upon securing the required degree of safety.

In the following, the features in type selection of concrete gravity dam and fill dam which are widely selected will be described.

- 1) A concrete gravity dam is the most general type, and whatever type may be finally selected, it is made an object of comparison study.
- 2) Hollow gravity dam and buttress dam are few in material-savings effects compared with their complexities in construction, and their advantages are becoming lessened.
- 3) In case of a fill dam, it is capable of adapting even where the geology is comparatively adverse. A fill dam is extremely weak against overtopping by flood on the other hand, and where the catchment area is large, it is necessary for a special study to be made of the method of controlling of flood.
 - A concrete or asphalt facing type fill dam has a weakness in the facing against settlement and earthquake motion, and it would be inadvisable to select this type unless there are conditions allowing the water level of the reservoir to be lowered for inspection and repair of the facing.
 - A zone-type fill dam is comparatively stable against settling compared with a homogeneous type fill dam so that the majority of dams are of this type. There are center core type and inclined core type depending on the location of the impervious core. Recently, the center core type has been adopted more often.

Table 11-1 General Factors for Dam Type Selection

Factor/Type	Concrete Dam	Rockfill Dam
Dam-site topography	A place where there is no extreme drop from upstream to downstream or no scraggly ridge. A narrow V-shaped valley in case of an arch dam.	Nothing in particular for the topography. The space of spillway should be considered.
Dam-site geology	Strength of foundation rock high, especially river bed and both abutments.	Selected often at sand-gravel foundations and rock foundations.
Meteorological conditions	No limitations in particular.	Limitations in core constructibility.
Dam height	No limitations in particular.	No limitations in particular.
Material conditions	When concrete aggregates obtainable.	When both core and rock materials obtainable.
Construction conditions	Depends on topography and scale but rapid construction possible.	Limitations in constructibility of core.

With the above taken into consideration, the dam type is considered capable of conforming with the siting characteristics of the Olur Dam and the recent technological trends are to be selected.

Firstly, with regard to fill dams, an interior core type is conceivable from the various conditions above. Concerning core materials, there is a deposit along the road at Kaledibi site on the left bank,

approximately 3 km upstream from the dam, which can be utilized as core material in large quantity. As coarse-particled sizes are somewhat high in content, it is thought to be effective for this to be used in blending with some amount of fine-grained material which is confirmed at the Yolboyu site. But there will be no problem in securing a quantity of about 600,000 m³.

As for rock materials, there is a quarry site, approximately 1 km upstream in a gully on the right-bank side immediately downstream of the dam, from which large-scale collection will be possible.

Regarding both abutments at the damsite, it has been confirmed as a result of investigations that fresh foundation rock can be obtained by excavation of 3 to 8 m. The river bed consists of deposits of thickness as much as 45 m, and methods such as removal by excavation and by grouting or cut-off slurry walling method are conceivable. But since the dam height exceeds 130 m, the safest method is removal by excavation of the core portion alluvium.

Regarding the foundation of the dam, it is normally considered that there will be risk of liquefaction in the following cases:

- When a sand layer exists within 15 m from the ground surface
- When N-value in the standard penetration test is under 15. (In case of fines content 15 to 35%)
- When purely a sand layer consisting of medium-grained sand of uniform particle size and silt and clay content is less than 10%.

- When a sandy soil of content of fines (74 μ and under) is less than 35%.

Regarding the sand layer existing at the foundation of Olur Dam, according to the results of investigations made so far, since it has been found that the thickness from the ground surface is approximately 25 m, and N-value from additional boring was higher than 15, it is considered that liquefaction of the dam foundation will not occur. In the aspect of design, waiting zones were provided at the cofferdams upstream and downstream of the dam giving consideration to stability of the tops of the slopes. However, in further detailed designing, it will be necessary for confirmations to be made of the distribution of sand layers over a wide area and of grain-size distributions through additional investigations.

In case of selecting a rockfill dam, the spillway is to be made a chute type of width 46.5 m at the right-bank side because of the topography, in order to safely release the design flood discharge of 4,750 m³/s (PMF). The energy dissipating device at the bottom part is to be made a flip bucket type.

On the other hand, a concrete dam would require further investigations to be made such as of bedrock strength. But a straight gravity dam of a general type of cross section is to be considered.

In case of a concrete dam there would be advantageous aspects compared with a fill dam with respect to flood disposal and the liquefaction of the foundation mentioned previously. Further, when considering this type of dam, it would be necessary for investigation to be made of items such as foundation rock strength, but a concrete arch dam which is possible to an extent

was also made an object of comparison. Further, in case of a concrete dam, a large quantity of good-quality cement would be required, and Kars which is comparatively close by, may be considered as the candidate production site. In this case, the transportation distance would be approximately 170 km, while road conditions are favorable.

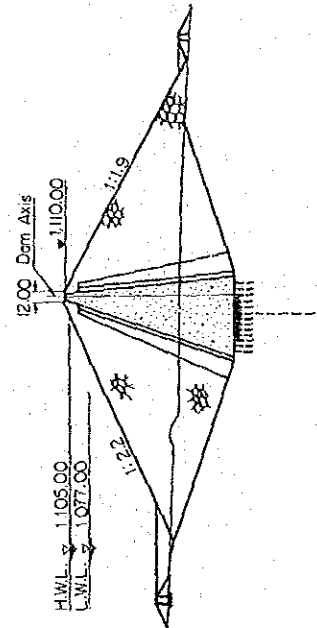
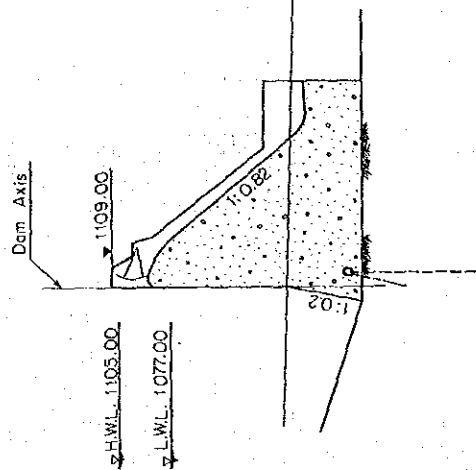
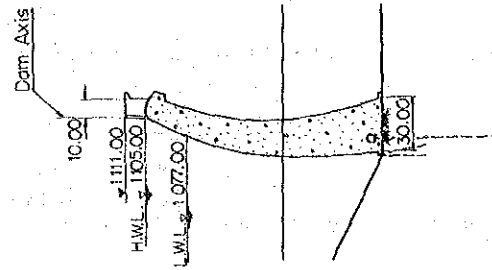
With the above as the objects, basic designs were made of the individual types, and the results of comparing the economic natures are as shown in Table 11-2 and Table 11-3.

From the above results, a rockfill dam may be considered to be advantageous with respect to the economics, and for the Olur site, although depending to an extent on the results of further investigations, a rockfill dam is to be adopted in this study.

Table 11-2 Comparison of Dam Type, Olur (1)

Specification/Type	Rockfill	Concrete Gravity	Concrete Arch
High water level (m)	1,105	1,105	1,105
Dam height (m)	136	137	146
Spillway type	Chute	Center overflow	Center free fall weir
Gate	Radial	Radial	---
W x H (m)	13.5 x 16.5	13.5 x 16.5	---
Specifications			
Diversion tunnel length (m)	530	502	696
Inside diameter (m)	6.0	6.0	6.0
Dam excavation (10^3 m^3)	651	1,396	1,686
Dam volume (10^3 m^3)	3,818	1,045	874
Spillway excavation (10^3 m^3)	546	---	---

Table 11-3 Comparison of Dam Type, Olur (2)

Item	Rockfill Type	Concrete Gravity Type	Concrete Arch Type
Typical Section			
Slope	Upstream Downstream	(1 : 0.2) 1 : 0.82	- -
Dam Volume	3,817,900 m ³	1,045,100 m ³	874,100 m ³
Construction Cost	182,102 x 10 ⁶ TL	276,568 x 10 ⁶ TL	260,753 x 10 ⁶ TL
Ratio of Construction Cost	100	151.9	143.2

(3) Designs of Structures

1) Dam

As stated up to the preceding clause, a rockfill type is selected for the Olur Dam. The topography and geology of the damsite are as previously described, and with nothing to pose a problem, this is a site suited to this type of dam. Regarding the type of the impervious core, recent high rockfill dams (more than 100 m) in the world have almost all been center core type, and since there is no topographical condition at this site for an inclined type to be adopted in particular, a center core type was selected. Concerning the width of the core, a core having a thickness of 30 to 50% of water pressure is considered to be safe even under fairly adverse conditions regardless of the soil and dam height. The core material for Olur Dam is available at the left bank 3 km upstream, and this has a high content of coarse particles, but quantity can be secured with comparative ease. Because of this, the core was made some-what thick and about 50% of water pressure from the viewpoints of permeability and securing of quantity. Regarding dam configuration, a horizontal seismic coefficient of 0.15 was adopted, the examples of similar sites in Turkey were referred to, and the upstream slope was made 1:2.2 and the downstream slope 1:1.9. Regarding stability of the dam, the result that there will be no special problem has been obtained also in stability calculations including the dam foundation. (See Table 11-4)

Regarding dam materials such as for the filter and rock zones, the abundant river-bed deposits and muck from excavation for structures are to be diverted as much as possible, and a zone type is to be adopted.

A quarry is to be provided at the right bank downstream of the dam.

Table 11-4 Stability Analysis of Olur Dam

Condition	Hydrostatic Pressure		Pore Pressure	Seismic coefficient	Minimum Safety Factor	
	Upstream Side	Downstream Side			Up-st. Side	Dn-st. Side
End of Construction • Normal	Water Level 1,026.00	Water Level 1,026.00	Value at the end of Construction	---	1.9	2.0
End of Construction • Earthquake	Water Level 1,026.00	Water Level 1,026.00	Value at the end of construction	k=0.10	1.3	1.5
Steady Seepage • Normal H.W.L.	H.W.L. 1,105.00	Water Level 1,026.00	Based on steady seepage at H.W.L.	---	2.2	2.0
Steady Seepage • Earthquake H.W.L.	H.W.L. 1,105.00	Water Level 1,026.00	Based on steady seepage at H.W.L.	k=0.15	1.6	1.3

2) Spillway

It will be necessary for the spillway to be capable of safely releasing the design flood discharge of $4,750 \text{ m}^3/\text{s}$ (PMF). From the standpoints of topography and geology, it would be possible for the spillway to be provided at either bank, right or left, but when the plan configuration such as the flow direction and excavation volume are considered, the right-bank side would be suitable. Accordingly, a chute-type spillway of width 46.5 m with 3 gates is to be provided on the right bank. The energy dissipating device at the

bottom part is to be made a flip bucket type since the right bank on the downstream side happens to be the outlet of a large gully, the fan deposit is deep, and provision of a large-capacity stilling basin is difficult topographically. Moreover, there are no houses in the vicinity.

3) Care of River

Care of river for the dam would consist of upstream cofferdam, diversion tunnel, and downstream cofferdam. The upstream cofferdam is for the purpose of preventing inflow of river water to the work area during construction of the main dam, and diverting the river water to the diversion tunnel.

The location of the upstream cofferdam was arranged further upstream from the toe of the main dam in order to facilitate excavation of the thick river-bed deposit at the core section and to provide a extra banking for the counterweight between the main dam and the cofferdam taking into account the fact that the foundation of the rock zone contains sand and silt layers. The height of the cofferdam was set for a design flood discharge ($332 \text{ m}^3/\text{s}$) of 25 years return period flood. But to increase safety, it was designed so that when the water storage effect was considered there would be no overtopping even with a 50 year return period flood ($376 \text{ m}^3/\text{s}$).

The height of the cofferdam is 22 m and is to be such that overtopping will not occur in the flood season and the cofferdam is to be completed during the non-flood season. The diversion tunnel is to have a discharge cross section ($D = 6.0 \text{ m}$, $L = 530 \text{ m}$) with which it would be possible for the design flood discharge to be released completely downstream without

overtopping the cofferdam in the flood season, and in a manner that the construction cost including the cofferdam would be cheapest. The height of downstream cofferdam was determined as 6.5 m that the downstream water level would not overtop, during the flood season.

4) Outlet Works

Outlet Works are to be provided for the purposes of irrigation downstream at the time of initial water impoundment and emergency dewatering. The outlet works is to be provided by the conversion of the diversion tunnel to a valve chamber; but since the diversion is planned with a single tunnel, it is to be of a construction possible for branching to be done at the section of the diversion tunnel for plugging of the diversion tunnel.

The outlet works would consist of a high pressure valve and a high-pressure slide gate for emergency purposes.

The inner diameter and the length of the outlet works are 4.0 m and 148 m respectively, and the diameter of the high pressure valve is 1.5 m. Outlet Works is consisted of a main valve and an emergency high pressure gate.

The inlet structure of the outlet works is to be of reinforced concrete and provided at the diversion tunnel inlet, the elevation being set at the vicinity of the planned sedimentation level of the reservoir. The capacity is set for it to be possible for a design discharge of 44 m³/s to be released at high water level of the reservoir. Furthermore, besides the emergency

outlet valve, a facility for irrigation is to be provided.

11.1.2 Waterway and Powerhouse

(1) Location and Outline

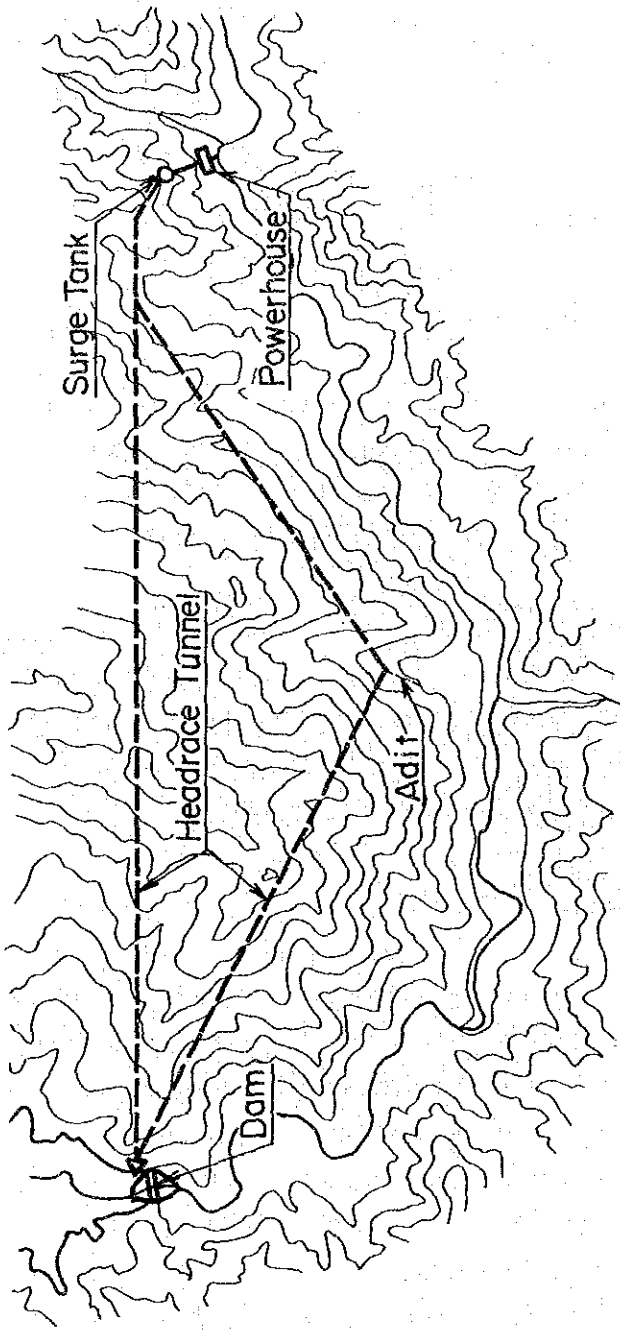
The headrace of the Olur Project is to consist of a tunnel of approximately 10 km from an intake provided at the left bank of the reservoir and taking advantage of the meandering of the river to obtain a head of 180 m.

The geology of the route of the tunnel is as described in 7.3.1(3), and although faults and sheared zones to be especially problematic geologically in tunnel excavation have not been recognized, there is a part near the end of the headrace where deposits are thick.

(2) Study of Waterway Route

For the headrace route, what is first conceivable is the shortest path connecting the intake and powerhouse. In this proposal, the waterway length is approximately 9,660 m, while it is impossible to provide work adits along the way because of the topographical conditions. In tunnel excavation in general, in case of the conventional method, it is effective to provide a work adit at about every 3,000 to 5,000 m when considering work efficiency. Therefore, comparison studies were made of the three alternatives of a case of excavation of the shortest route by the conventional method, a case of excavation by TBM (Tunnel Boring Machine), and a case of excavation by the conventional method but providing a work adit at an intermediate point as shown in Table 11-5. As a result, it was found that the excavation of the shortest route by the conventional method would be the cheapest, but

Table 11-5 Comparison of Waterway, Olur

Item	Straight by NATM	Straight by TBM	Alternative by NATM
General Plan			
Total Length of Headrace Tunnel	9,660 m	9,660 m	11,000 m
Construction Cost of Headrace Tunnel	121,407 x 10 ⁶ TL	166,325 x 10 ⁶ TL	137,875 x 10 ⁶ TL
Ratio of Construction Cost	100	137.0	113.6
Construction Period	70 months	34 months	48 months

approximately 70 months would be required for the construction period. The TBM method would shorten the construction period to 34 months but the construction cost would be increased approximately 37%. In the case of providing a work adit, it would shorten the construction period to 48 months but the construction cost would be approximately 14% higher while there also would be loss in kWh production due to increased tunnel length. Accordingly, the construction period in case of the shortest route by the conventional method would be problematic. But when the dam construction period including care of river works such as the diversion tunnel are taken into consideration, it could be judged that the overall construction schedule would not be impeded if tunnel excavation were to be commenced at an early stage. The original plan for the conventional method with the construction cost the lowest was adopted.

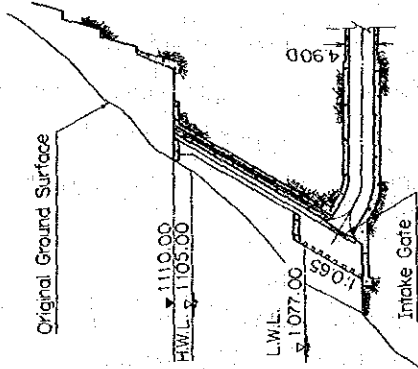
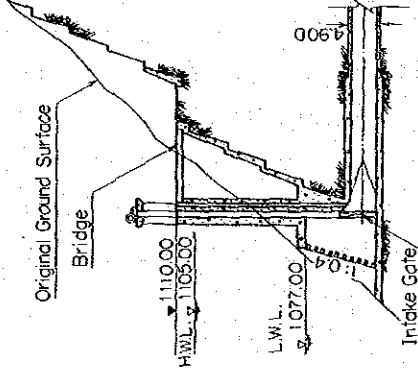
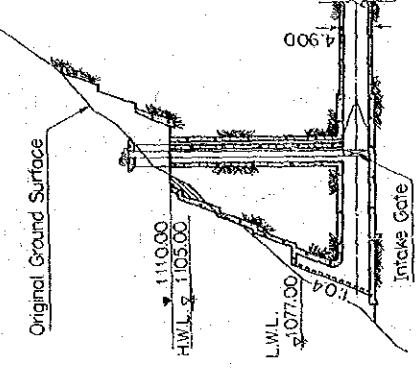
(3) Designs of Structures

1) Intake

The intake is to be provided at a gully at the left bank approximately 170 m upstream from the dam axis in consideration of the tunnel route, the topographical and geological conditions, and the plan for an access road.

The topography and geology of this site are as described in 7.3.1(2). There is no problem regarding the foundation for the intake structure. However, the topography is that of a steep slope of approximately 40 deg. and if open excavation for the intake structure were to be done excessively, it is thought it would be disadvantageous both economically and in the aspect of slope stability during construction. The type of intake structure was considered making a

Table 11-6 Comparison of Power Intake Type, Olur

Item	Inclined Type	Intake Tower	Gate Shaft
Typical Section			
Excavation Volume	26,300 m ³	12,200 m ³	12,500 m ³
Concrete Volume	5,300 m ³	7,200 m ³	4,600 m ³
Construction	Good	Fair	Good
Ratio of Construction Cost	100	113.8	115.2

comparison study of the construction costs for the three conceivable types. As a result, an inclined type was judged to be optimum with respect to the economics and constructibility. The alternatives are shown in Table 11-6.

The elevation of the basement of the intake structure was decided at elevation 1,067 m, to take into consideration the future sedimentation, and also the elevation can avoid the vortex for the conduit system.

2) Headrace

The topography and geology of the headrace tunnel route are as described in 7.3.1(3), and there are no serious problems posed in particular. Of the tunnel route, there is a part of thin overburden near the end of the tunnel. This has been confirmed by results of seismic prospecting, so it was considered to install the steel lining about 250 m of this part.

The elevation of the center of the headrace tunnel was decided at elevation 1,056.5 m to take into consideration passing the thin overburden part, and the down surging of the sudden load reduction of the power.

The inner diameter of the tunnel, as a result of the economic comparisons shown in Fig. 11-1, was decided to be made 4.9 m. Considering the fact that the geological conditions of the tunnel route are generally favorable, it is thought there would be ample safety against the anticipated internal and external pressures with a lining thickness of 0.4 m.

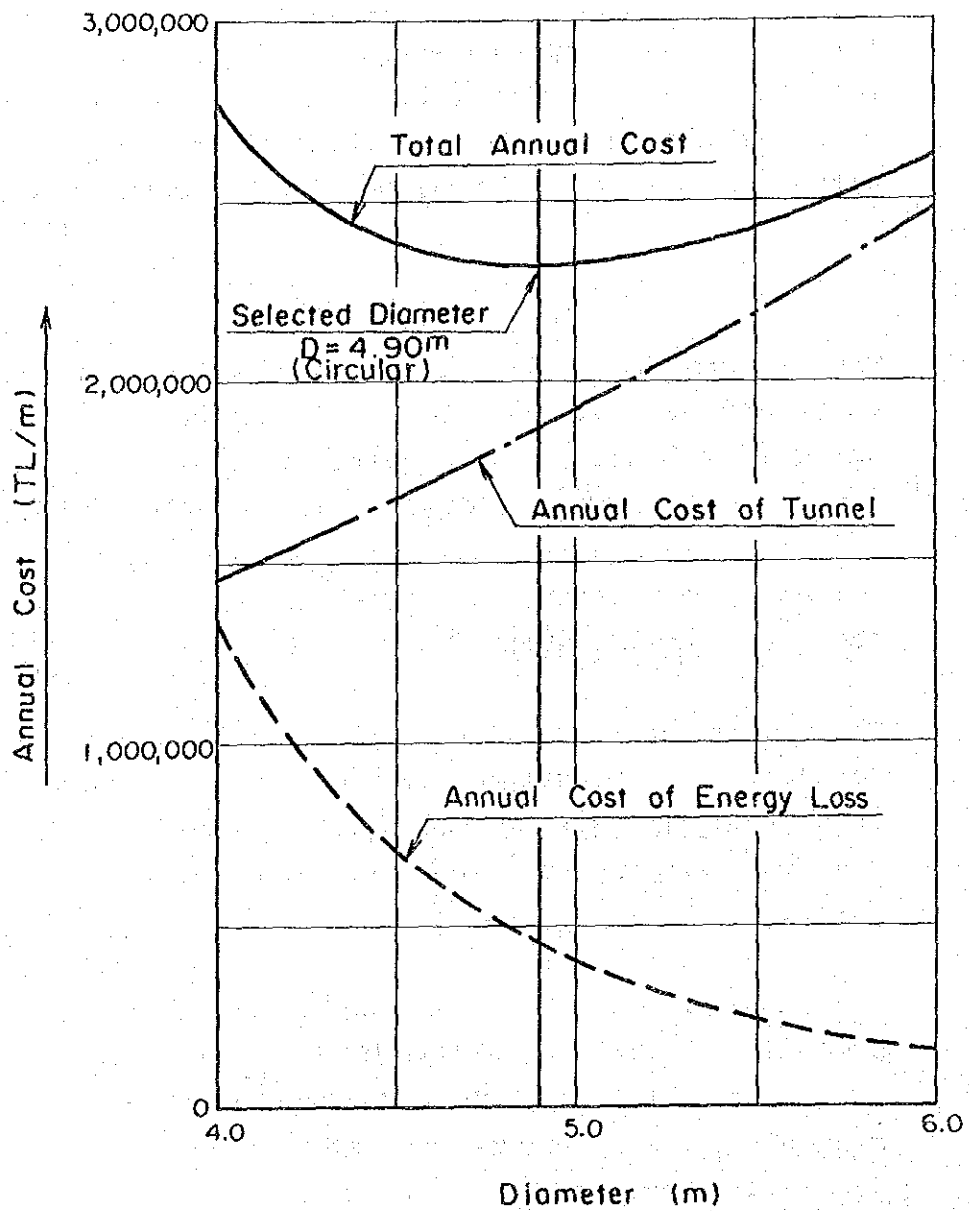


Fig. 11-1 Estimation of Optimum Diameter of Headrace Tunnel, Olur

3) Surge Tank

The length of the headrace in the Olur Project is long at approximately 10 km, and a surge tank will be needed in the headrace to adjust available water in accordance with sudden increase in load at the power station and to absorb water-hammer action when the sudden load reduction occurs. The surge tank is to be provided at the upper part of a ridge at the left bank immediately downstream of the village. The topography of the surroundings is complex with almost no topsoil, and slopes are steep and rock masses are outcropped. But an access road can be provided with comparative ease from the upstream side. No problem geologically has been pointed out in particular.

The two types of restricted orifice type and water chamber type were examined for the type of the surge tank in view of the topography. The results of the examinations are shown in Table 11-7. As a result, it was found that the restricted orifice type excelled with respect to the economics, and that it would also be superior in constructibility. The inside diameter of the vertical shaft was decided on at 12 m based on the conditions for static and dynamic stability of the water surface.

4) Penstock

The penstock is to be located roughly down the middle of the ridge connecting the powerhouse to be provided at the river-bed, immediately downstream of the village and the surge tank, and has a plan configuration of a more or less straight line from the surge tank to the powerhouse (OPK Alternative).

Table 11-7 Comparison of Surging Handling, Olur

Item	Orifice Type	Chamber Type
Typical Section		
Excavation Volume	Open Shaft, Tunnel 10,000 m ³ 14,300 m ³ Total 24,300 m ³	Open Shaft, Tunnel 21,800 m ³ 27,600 m ³ Total 49,400 m ³
Concrete Volume	4,200 m ³	5,100 m ³
Ratio of Construction Cost	100	133.2

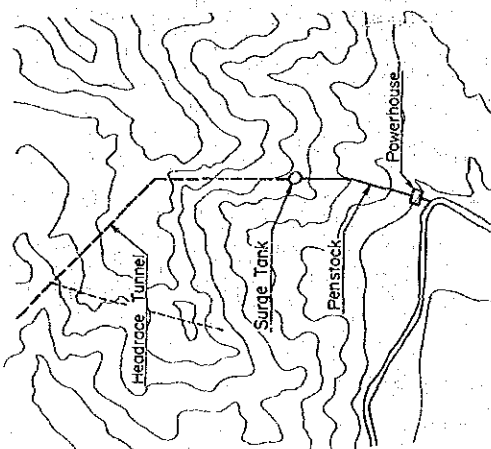
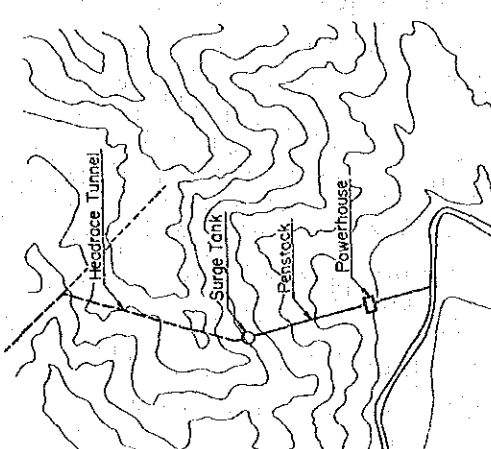
This site is slightly rugged in the vicinity of the surge tank, but the geology is favorable, as indicated in 7.3.1(4), and it is suitable for a surface-type penstock which would be easier to construct.

Further, for an alternative penstock, several upstream locations were studied for rough economic comparisons, and field reconnaissances were made, and as a result, a site approximately 300 m upstream (OPT Alternative) was selected. In the event this alternative is adopted, the thalweg of the river would become distanced from the headrace, and since there are river deposits in terraced form in the vicinity of the river bed, a layout providing the powerhouse immediately below the penstock and leading to the river with a culvert tailrace would be provided. Concerning the foundations of the powerhouse and tailrace in OPT Proposal, the depths of the basement rocks are 60 m which have been confirmed through exploratory drilling and seismic prospecting.

The results of studies on the two cases above are given in Table 11-8. As a result, it was judged that the OPK Proposal is superior in the economics and hydraulic characteristics would be advantageous.

The penstock would have a length of 436 m and a maximum static head of 181 m, and design and construction of steel pipe can be amply done based on domestic procurement in Turkey. The average diameter would be 3.6 m, fairly large-diameter for a surface penstock. Therefore, the turbine center elevation was studied so that even when there is load variation at the turbine when the dam water level is lowered, negative pressure will not occur in vicinity of the upper bend, and EL. 1,046 m was adopted.

Table 11-8 Comparison of Penstock Layout, Olur

Item	OPK	OPT
General Plan		
Length of Waterway and Powerhouse	Headrace 9,661 m, Penstock 436 m, Powerhouse and Outlet 44 m	Headrace 9,361 m, Penstock 341 m, Powerhouse and Outlet 44 m, Tailrace Culvert 283 m
Construction Cost of Waterway and Powerhouse (Excluding Dam)	152,229 x 10 ⁶ TL	162,976 x 10 ⁶ TL
Ratio of Construction Cost	100	107.1
Total Head of Power Generator	166.67 m	165.67 m

5) Powerhouse and Switchyard

The powerhouse would be located at the left bank where the Oltu River, near a village 7.5 km downstream from the damsite, bends roughly at a right angle. The topography and geology are as described in 7.3.1(4). The ridge at the left bank, to be the penstock route, forms a steep cliff near the river edge. Immediately below the penstock, there is a part where taluses from the ridge are deposited so that the penstock route will be located shifting slightly to the upstream side where outcropping of basement rock has been confirmed.

It is possible for a ground surface type powerhouse to be provided at this site from a topographical standpoint and a tailrace would be unnecessary, so that a comparatively compact power station can be provided, while an access road and an equipment delivery road can be readily connected.

The elevation of the powerhouse at the ground surface is selected at EL. 934 m from the standpoint of equipment layout. The design flood discharge at the powerhouse site is taken to be 500 m³/s (200 year return period flood), and the water level in this case would be at elevation 930 m.

Consequently, this powerhouse would have a reasonable height allowance even during flood, and it may be said there would be no risk of inundation.

Regarding the switchyard, there is a flat river terrace area at a point 100 m upstream and it would be possible for the switchyard to be provided with only a small amount of excavation. On the other hand, as for an alternative, a site approximately 300 m upstream (OPT Alternative) is conceivable as mentioned

under the clause on the penstock. This powerhouse site would be provided more toward the mountain than the riverbank terrace due to the topography, and therefore, would require a tailrace of approximately 250 m.

11.1.3 Electro-mechanical Equipment

(1) Selection of Number of Units

The installed capacities of both Olur and Ayvalı projects are 65 MW and 125 MW, respectively.

Keeping the installed capacity as the same, there are many conceivable combinations of number of units and unit capacity, but the fewer the number of units is, the more economical the construction cost becomes.

The relatively high turbine efficiency will be obtained in the case of a one unit compared with a two unit case, and thus result in the increase of annual energy production. The following hydroelectric power generation plans in the downstream area of Çoruh river are scheduled to be constructed and connected to 154 kV or 380 kV network, to which both Olur and Ayvalı power stations are also incorporated.

Yusufeli	180	MW x 3 units
Artvin	160	MW x 2 units
Deriner	167.5	MW x 4 units
Borçka	150	MW x 2 units
Muratlı	57.5	MW x 2 units

On account of operation of those power stations as a group, selection of any operable unit in the group including the Olur and Ayvalı power stations does not raise any objection

as each unit has a similar unit capacity or is interchangeable, i.e. when one is selected for the number of units for both Olur and Ayvalı power stations.

Even in the case of unit shut down of the Olur or Ayvalı power station by failure, there is no undesirable affect on the power system stability in one unit case.

The transportation road up to the Yusufeli site will be constructed in advance of this project, and the upstream portion from the Yusufeli site to this project site is to be extended during construction of this project.

The downstream portion of the transportation road is sufficient to transport bulky items of this project as the unit dimensions and weight of this project are less than those of Yusufeli project, and also obvious increase of construction cost for the upstream portion will not occur, even in the one unit case for both power stations.

Taking all those above mentioned, the number of units for both Olur and Ayvalı power stations will be one each.

(2) Type and Ratings of Major Equipment

From the maximum discharge and effective head, vertical shaft Francis is judged as appropriate.

The generator is directly coupled to the turbine shaft, vertical shaft, three phase, alternating current, synchronous generator.

Generator voltage is stepped up to the transmission voltage by main transformer.

The type of main transformer will be three phase, oil-immersed, as the capacity is not so large, and weight and

dimensions less than transportation limitation, accordingly.

154 kV transmission line will be terminated at outdoor switchyard to send power generated at Olur power station to Ayvali switchyard.

The type of switchyard is that of Aluminum pipe bus as of TEK's standard for the time being, but GIS type may be considered as an alternative to catch up with technological aspects.

The ratings of major electro-mechanical equipment are as follows;

Water Turbine

Type	Vertical shaft, Francis
Number of units	1
Normal effective head	157.2 m
Maximum discharge	48 m ³ /s
Turbine output	66500 kW
Revolving speed	333 rpm

Generator

Type	Three phase, Alternating current, synchronous
Number of units	1
Output	74000 kVA
Power factor	0.9 lagging
Voltage	11 kV
Frequency	50 Hz
Revolving speed	333 rpm

Main Transformer

Type	Outdoor, three phase
Number of units	1
Capacity	74000 kVA

Voltage primary: 11 kV
secondary: 154 kV

Outdoor Switchyard

Bus system single bus + transfer Bus
 (future)
Bus Aluminum pipe
Number of transmission
lines connected 154 kV 1 for Ayvali
 34.5 kV 1

(3) Main Circuit Equipment

As the powerhouse is a ground surface type, the main transformer is equipped at outdoor switchyard near the powerhouse and generator lead and main transformer are connected by 11kV power cable or segregated phase bus.

A parallel-in circuit breaker is equipped at low tension side of main transformer and used for synchronizing generator to the power system.

The number of 154 kV transmission lines which take off from the outdoor switchyard is only one for the Ayvali switchyard at the time of commissioning, and the single bus system is applied, accordingly.

Another one 154 kV transmission line is planned for future extension, and the space for the outdoor switchyard will be such that one transfer bus can be added. It is TEK's practice to provide a transfer bus to enable the inspection of circuit breaker for transmission line with the line alive and by-passed through the transfer bus.

One feeder is branched from the single bus to a step-down transformer and connected to one 34.5 kV transmission line and one 6.6 kV station service circuit.

To secure station service power in any failure of transmission line or switchyard equipment, emergency power source of a diesel engine-generator set will be equipped in this power station.

Fig. 11-11 and Fig. 11-12 indicate the single line diagram of the power station and switchyard plan, respectively.

(4) Telecommunication Equipment

Power line carrier system (earth return) is provided for composing telecommunication circuits for power generation and dam operation.

11.2 Ayvalı Project

11.2.1 Dam and Auxiliary Structures

(1) Location and Outline

The projected damsite is located approximately 20 km downstream of the Olur Dam, and the end of the backwater of the projected Yusufeli Reservoir would be reached approximately 10 km further downstream.

The topography at the damsite is that of a V-shaped valley of gradients of the right and left banks approximately 40 deg. and roughly symmetrical, and although the slopes of the two banks are slightly steeper compared with the upstream of the Olur damsite, the valley width is larger. The projected damsite is a location topographically advantageous since the length of the diversion tunnel can be made short by utilizing the meandering of the river. From topographical and geological viewpoints, there is no favorable damsite seen in the vicinity which can serve as an alternative. The river-bed at the project site is

covered by alluvium of a thickness of 60 m, but according to the results of investigations by drilling up to now, thick sand layers have not been confirmed. Further, since N-values in standard penetration tests on the river-bed deposits are 17 to 50 and higher, it is thought there is little possibility of liquefaction occurring at the Ayvalı site.

The geology of the damsite, as described in the part on geology, has almost no tree growth or topsoil, and weathering has occurred to about 3 to 5 m. The basement rock is hard and large faults have not been confirmed to exist.

(2) Selection of Dam Type

Regarding the procedure followed in selection of the dam type, it has been described in detail in the part on the Olur dam so it will be omitted here.

The Ayvalı damsite is a comparatively good location where it will be possible topographically and geologically for either a fill-type dam or a concrete dam to be constructed. However, the valley width is large at approximately 500 m, while dam height would be 180 m or more from the foundation rock, and especially, the right-bank side is a rock mass having numerous cracks. Because of this, a hard and dense bedrock required as the foundation for an arch dam cannot be hoped for, and it is thought difficult from the standpoints of safety and economy for an arch dam to be adopted. Therefore, comparative designs were made of the two cases of rockfill dam and concrete gravity dam.

The materials available at the damsite, similarly to the case of the Olur Dam, were the abundantly existing river-bed sand-gravel and talus deposits in the vicinity of the damsite, and test results were examined to judge whether

they would be usable. The particle-size distribution of the river-bed sand-gravel is fairly good. So, if can be selected, the over sized gravel can be selected, it may be judged that more than 70% of the river deposits are amply usable as concrete aggregates and also as filter materials for the fill type dam.

On the other hand, considered from the aspect of materials in case of adopting a fill-type dam, it is considered possible to borrow core materials from deposits at both banks in the Bulanik Valley 8 km downstream of the dam and the Tavusker Valley 8 km upstream. There is a considerable amount of scatter in gradations at the Bulanik candidate site. The material of the Tavusker candidate site does not have as much scatter as that of the Bulanik candidate site, while the gradation is fine so that it is quite possible for this to be used. The two candidate sites in connection with Ayvalı Dam do not necessarily have large quantities, but the problems of quality and quantity can be solved by using the materials of the two sites independently or combining the materials of the two.

As mentioned before for filter material, it is thought that the river-bed sand-gravel in the surroundings can be amply used both in quality and quantity. Regarding rock materials, it has been ascertained that good-quality material exists in large quantity at the left bank upstream of the dam. As stated up to this point, it is possible for either a concrete dam or a fill dam to be constructed as seen from the aspects of topography, geology, and materials, and therefore, designing was done for the respective types considering materials and work execution and a comparison study was made of the economics.

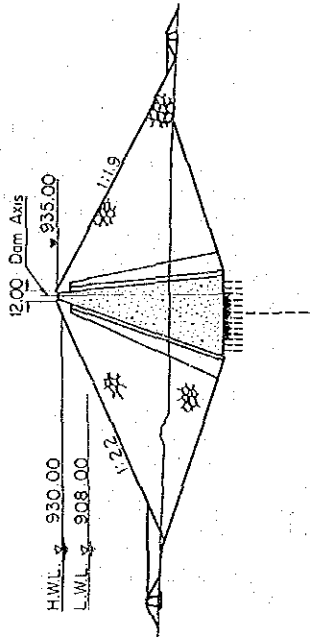
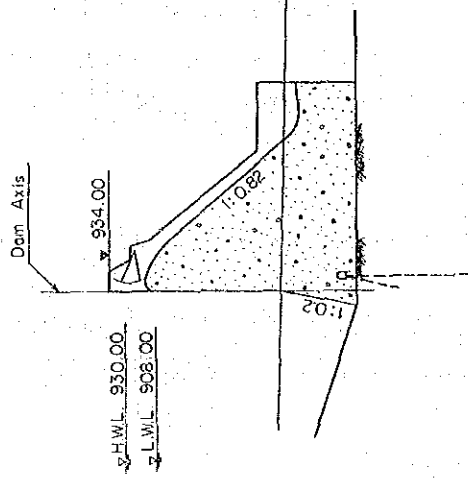
The results of studies are as shown in Table 11-9 and Table 11-10.

According to the above results, it is considered that a rockfill dam would be advantageous with regard to economics in comparison with a concrete gravity dam, and it is thought the rockfill dam proposal would be optimum at the Ayvalı site.

Table 11-9 Comparison of Dam Type, Ayvalı (1)

Specification/Type	Rockfill	Gravity
High water level (m)	930	930
Dam height (m)	175	177
Spillway type	Chute	Central free fall weir
Gate	Radial	Radial
W x H (m)	13.5 x 17.5	13.5 x 17.5
Specifications		
Diversion tunnel length (m)	659	536
Inside diameter (m)	6.0	6.0
Dam excavation (10^3 m^3)	1,112	2,392
Dam volume (10^3 m^3)	9,268	2,423
Spillway excavation (10^3 m^3)	1,105	---

Table 11-10 Comparison of Dam Type, Ayvali (2)

Item	Rockfill Type	Concrete Gravity Type
Typical Section		
Slope	Upstream	(1 : 0.2)
	Downstream	1 : 0.82
Dam Volume	9,268,100 m ³	2,423,100 m ³
Construction Cost	344,970 x 10 ⁶ TL	581,169 x 10 ⁶ TL
Ratio of Construction Cost	100	168.5

(3) Designs of Structures

1) Dam

With regard to the dam type, as a result of thorough field investigations of topography, geology, and dam materials, and study of the economics, a fill-type dam with center core type was selected, the same as for the Olur Dam. But because of the problem concerning available quantity of core material, the core was made slightly thinner than for the Olur Dam and about 45% of the maximum water pressure was made the measure. The dam configuration, because the embankment materials would be about the same as for the Olur Dam, and the design conditions regarding geology and earthquake were also the same, the upstream slope gradient was made 1:2.2 and the downstream slope gradient 1:1.9. Regarding stability of the dam, as shown in Table 11-11, the result that there will be no problem in particular has been obtained in stability calculations including the dam foundation. With respect to dam materials such as filter and rock out of the embankment materials, the river deposits, abundantly available in the vicinity, and material from a quarry at the left bank immediately upstream are to be used. For core materials, the deposits in the Bulanik Valley 8 km downstream and the Tavusker Valley 8 km upstream are to be used independently or upon blending together. And, the excavation muck from the spillway, underground powerhouse, and other structures are to be diverted to rock material as much as possible.

Table 11-11 Stability Analysis of Ayvali Dam

Condition	Hydrostatic Pressure		Pore Pressure	Seismic coefficient	Minimum Safety Factor	
	Upstream Side	Downstream Side			Up-st. Side	Dn-st. Side
End of Construction • Normal	Water Level 810.00	Water Level 810.00	Value at the end of Construction	---	2.2	2.0
End of Construction • Earthquake	Water Level 810.00	Water Level 810.00	Value at the end of construction	k=0.10	1.6	1.3
Steady Seepage • Normal H.W.L.	H.W.L. 930.00	Water Level 810.00	Based on steady seepage at H.W.L.	---	2.2	2.0
Steady Seepage • Earthquake H.W.L.	H.W.L. 930.00	Water Level 810.00	Based on steady seepage at H.W.L.	k=0.15	1.1	1.3

2) Spillway

The spillway is to be provided at the right bank of the dam in consideration of topography, geology, and downstream water-flow direction. The design flood discharge is 5,270 m³/s (PMF). To safely release this at maximum water level and during a flood, a spillway 46.5 m in width is to be provided and three radial gates of height 17.5 m and width 13.5 m are to be installed. The spillway is to be a chute type and energy dissipation downstream is to be by flip bucket.

3) Care of River

The river is to be diverted prior to the work on the dam proper providing cofferdams at upstream and downstream of the work area, and excavating a diversion tunnel. Regarding the locations of the cofferdams, they are to be 400 m upstream and downstream of the dam axis since sufficient distance is required in order not to hinder excavation of the thick river-bed sand-gravel. The height of the upstream cofferdam (24.5 m) was set based on a design flood discharge at the damsite of a 25 year return period flood ($376 \text{ m}^3/\text{s}$), but the height was made one at which a 50 year return period flood ($425 \text{ m}^3/\text{s}$) would not overtop the dam considering the water storage effect upstream of the cofferdam. And in a manner that the construction cost of the diversion tunnel ($D = 6.0 \text{ m}$, $L = 659 \text{ m}$) including the cofferdam would be cheapest. Further, the height of the downstream cofferdam (4.5 m) was decided so that the downstream water level at the time of flood would not overtop the cofferdam.

4) Outlet Works

For initial water impoundment of the dam and for emergency dewatering, the outlet works are to be provided utilizing the diversion tunnel. Since the diversion tunnel would be a single one, the construction is to be for branching at the section of the outlet valve in order to be able to carry out plugging of the diversion tunnel. The inner diameter and the length of the outlet works are 4.0 m and 156 m respectively, and the diameter of the high pressure valve is 1.5 m.

The inlet of the outlet valve is to be of reinforced concrete and provided at the inlet structure of the diversion tunnel, and the elevation made near the planned sedimentation level of the reservoir. The design capacity of the outlet valve made one with which it would be possible to discharge 53 m³/s at high water level of the reservoir. Furthermore, besides the emergency outlet value, a facility for irrigation is to be provided.

11.2.2 Waterway and Powerhouse

(1) Location and Outline

The Ayvalı Project is a dam-and-waterway type power generation project utilizing the head between the Olur Project at the middle stretch of the Olur River and the backwater end of the Yusufeli Reservoir planned downstream of the Olur Project. The course of the Olur River downstream of the planned Olur Power Station meanders in a westerly direction and merges with the Çoruh River. It has been confirmed in studies of the project made up to the present that it will be advantageous to utilize to the maximum the head of 230 m between the Ayvalı Dam located 20 km downstream of the Olur Dam, conducting water by a tunnel at the left-bank side to the Yusufeli Reservoir, taking advantage of the meandering of the river.

The topography and geology of the waterway route are as described in 7.3.2(3).

(2) Study of Waterway Route

The waterway route for the Ayvalı Project needs to pass Anzav Gully at the left-bank side near the Yusufeli Reservoir, so that it is difficult from the standpoint of

the topography close to the end of the Yusufeli Reservoir to conduct directly to the reservoir at a high elevation, and it is also economically disadvantageous.

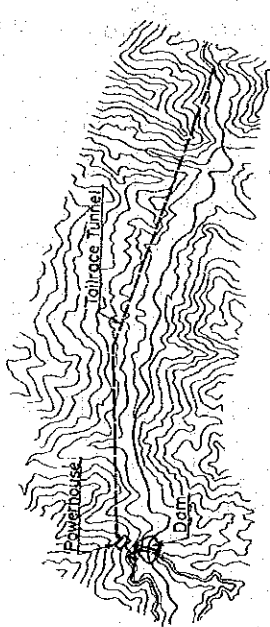
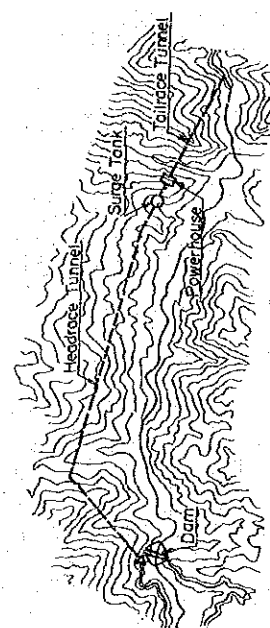
However, when attempting to utilize the head between the Ayvali dam and the Yusufeli reservoir to the maximum, there is no suitable site for a ground surface-type powerhouse which would be most economical. Therefore, a comparison study was made of two alternatives, that of providing an underground powerhouse immediately downstream of the dam and discharging downstream by the shortest route on the river side, and that of conducting water downstream by a headrace from the dam, providing an underground powerhouse in the vicinity of Anzav Gully, and discharging into the Yusufeli reservoir by a tailrace. The features of the respective alternatives are given in Table 11-12 below.

Table 11-12 General Factors for Waterway Route

Element/Type	Headrace Route	Tailrace Route
Topography & Geology	Adequately thick cover obtainable.	Cover thin compared with headrace route.
	Ayvali volcanic rocks	Ayvali volcanic rocks.
Construction conditions	Necessary for long period for construction of approach road.	Near river and work adit easy to provide.
	Necessary detour tributary.	Possible to shorten construction period.
Waterway shape	Circular pressure tunnel.	Horseshoe shape, non-pressure tunnel.

Upon consideration of the above elements, comparisons of the designs were made, and the economic natures were examined. The results of the examination are shown in Table 11-13.

Table 11-13 Comparison of Waterway, Ayvali

Item	APU	APL
General Plan	 <p>Topographic map of the APU project area. It shows a dam structure, a powerhouse, and two tunnels labeled 'Tolroce Tunnel' and 'Headroce Tunnel'. The terrain is hilly with contour lines.</p>	 <p>Topographic map of the APL project area. It shows a dam structure, a surge tank, a powerhouse, and two tunnels labeled 'Headroce Tunnel' and 'Tolroce Tunnel'. The terrain is hilly with contour lines.</p>
Total Length of Waterway	9,600 m	10,250 m
Construction Cost (Excluding Dam)	$159,925 \times 10^6$ TL	$197,988 \times 10^6$ TL
Ratio of Construction Cost	100	123.8

According to the above results, the tailrace alternative is advantageous from the aspect of the economics. Although the geology of the waterway route in the case of the headrace alternative is one where earth cover is thick, and it is thought there would be no problem geology-wise, cover of a minimum of 100 m can be obtained and there will be no problem in executing work. On the other hand, the geologies of the underground powerhouse sites are more or less equal, and there is not much difference between the two. Consequently, for the waterway route of the Ayvali Project, the tailrace alternative is to be adopted from the point of view of the economics.

(3) Design of Structures

1) Intake

The intake is to be provided at the left-bank side approximately 100 m upstream from the dam considering the topography, geology, and access road situation of the tunnel route and its surroundings.

The topography and geology at this site are as described in 7.3.2(2). Geology, there is no problem as a foundation for the intake structure. However, since the topography is rugged, if open excavation were to be done excessively, it would be economically disadvantageous. The type is an inclined type similar to the intake of the Olur Project

Additionally, the elevation of the basement of intake structure was decided at elevation 898 m to counter measure the maximum sedimentation volume and avoiding the vortex for the conduit system.