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Air defence systems for countries and continents ........................................ 11.23
Jockeying for supremacy in Europe’s own space race .................................. 11.36
How much longer will silicon be used? ......................................................... 11.40

Projects

Computerscope - 1 ......................................................................................... 11.18
Voltage comparison on a scope ........................................................................ 11.25
IDU for satellite TV reception - 1 ................................................................. 11.26
RF circuit design ............................................................................................ 11.38
Dark-room exposure meter ............................................................................ 11.44
High-power AF amplifier - 2 ........................................................................... 11.46

Information

News and views ............................................................................................... 11.16
New products .................................................................................................. 11.62
Corrections ...................................................................................................... 11.74

Guide lines

Classified ads .................................................................................................. 11.74
Index of advertisers ........................................................................................ 11.74

Selex - 17

Measuring Techniques (chapter-3) ................................................................. 11.55
Measuring power with a Multimeter .............................................................. 11.67
Power Calculations ......................................................................................... 11.60
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<tr>
<th>Parameter</th>
<th>Value</th>
</tr>
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<tr>
<td>Rated capacitance</td>
<td>1500 pF — 0.82 μF</td>
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<tr>
<td>Tolerance</td>
<td>±5% &amp; ±10%</td>
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<tr>
<td>Rated voltage (dc)</td>
<td>250-630, 1000, 1500, 2000 Volts</td>
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<td>Rated temp.</td>
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</tr>
<tr>
<td>Climatic category</td>
<td>IEC 68 40/085/56</td>
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COMPUTER-SCOPE-1

by R v Linden

To help those many people whose workshop includes a computer but not an oscilloscope, this article presents a drive unit that enables the computer to be used as an oscilloscope.

Block diagram

The first section of the unit, i.e., the AC-DC switch, attenuator, and amplifier, consists of analogue circuits, although the first two are controlled by binary signals. An off-set may be added to the amplifier via a digital-to-analogue converter.

The incoming signal is then digitized, after which a fast analogue-to-digital converter translates the samples into 7-bit data words that are stored in a random-access memory via a buffer stage.

The output of the A-D converter is also applied to a trigger and comparator. The latter compares the instantaneous value of the signal with the preset trigger level. If that level is exceeded, either in a positive or in a negative sense, depending on the preset trigger edge (leading or trailing), the data is stored in correct sequence in the RAM with the aid of an address counter. The eighth bit of every byte indicates when triggering
takes place.

The memory consists of two parts, each of 256 bytes. The first part holds 256 pre-trigger measuring points, and the second 256 post-trigger measuring points. A comprehensive address ensures that writing to, and reading from, the memory takes place in the correct sequence.

Sixteen different time bases, derived from a crystal oscillator, are provided.

Communication between the drive unit and the computer is effected via two 8-bit I/O buses; this enables most computers to be used.

The computerscope is controlled via software. A menu-type layout gives the user the possibility of setting all measuring parameters without few keys.

**Circuit diagram**

The input is at the centre of Fig. 2 and is connected direct to the AC-DC switch, which is formed by capacitor C1 shunted by D1 relays Relay R1. The switch is followed by the attenuator, which consists of two parts: the first can be arranged to attenuate by a factor of 1, 2, or 5, and the second by 1, 10, or 100. The total attenuation is set from the computer by means of multiplexers IC1 and IC2. Both IC1 and IC2 are controlled via lines V0 and V1, while the multiplexers in IC2 are switched via lines V3 and V6. In this manner the sensitivity can be set between 10 mV/div and 5 V/div.

At the input of the first divider relays instead of multiplexers are used because of the maximum allowable voltage here. Diodes D1 to D6 incl. protect the inputs against too high voltages.

A variety of fixed and variable capacitors in the attenuator sections provide compensation when squarewave signals are processed.

The signal at the output of the second attenuator has a maximum value of 89 mVpp. It is then raised to 2 Vpp by amplifiers A1, A2, and A3, to give a signal that can be processed by the A-D converter at the highest possible resolution.

An off-set is added to the signal in A3 to enable a vertical shift across the monitor screen. The off-set is provided by a D-A converter, which is housed in IC10 together with the fast A-D converter. The computer sends
Fig. 2. The circuit diagram of the computer-science drive unit.
the off-set data to the D-A converter via lines OF2 to OF10 incl. The standard value of the offset is 1 V when the sample lies exactly within the range of the A-D converter (bear in mind that the converter cannot process negative values). Only seven of the ten bits available at the D-A converter are used here which, because of the graphics resolution of the computers used, is more than enough.

The output signal of A5 is then applied to a fast A-D converter, which contains a separate comparator for each digital level, i.e. 256 in all. The conversion time is, therefore, only 26.3 ms (corresponding to a frequency of 38 MHz). The highest clock used is 8 MHz, resulting in 8 samples per period at an input signal of 1 MHz.

Because of the arrangement of the system, only 7 bits are used, which is sufficient for an accurate display. The eighth bit is used for storing the trigger data. The reference voltage for the A-D and D-A converters is generated by the IC itself. The only external component is capacitor C16.

The memory section also needs fast IC5, and in the present circuit Type IMS/1420 was chosen. This is a RAM with an access time of 45 ns and a capacity of 4 K x 4 bit. Two of these ICs, IC17 and IC18, are connected in parallel to give a width of 8 bits. The data lines of the RAMs are connected with the outputs of the A-D converter via buffer IC14. Of the total memory capacity only 512 bytes are used for data storage; the remainder is available for possible later extensions.

The digitizing and storing of the samples is taken care of by the clock provided by the time base. Operation of the A-D converter is commenced at the trailing edge of the clock pulse, and at the same time the address counter of the RAMs is increased by 1.

The random-access memory consists of two pages: page 1 contains address 000 to 0FF, and page 2, addresses 100 to 1FF. In the absence of a trigger signal, page 1 is written to. As soon as a trigger arrives, the eighth bit of the relevant byte goes high, when writing is transferred to page 2—see Fig. 3. Once this page is full, writing to the memory stops, and the computer is advised that writing is completed by the READY signal. In this way, there are always 256 pre-trigger and 256 post-trigger sampling points. After the computer has read the memory, the next writing cycle can be started. As long as there is old information on page 1, triggering is prevented for 256 clock pulses by signal INH. This is calculated by the computer from the state of the time base.

The time base (at the left of Fig. 2) consists of crystal oscillator N-N1, which is followed by a number of 2 and 5 dividers, IC2 to IC5 incl. The computer connects one of the outputs of the time base to the CLK line of the system via multiplexer IC4.

Latches IC7 to IC9 are provided for the exchange of signals between the drive unit and the computer. Circuits IC7 to IC9 serve to store data which are sent by the computer to the drive unit via PB1 to PB5 incl. PA5, PA1, and PA2 are used as write signals for the three ICs.

Circuits IC12 and IC15 form an 8-bit comparator for the trigger signal, which is provided (in binary form) by

the computer. As soon as the level of the input signal exceeds that of the trigger, the level at the >Q output of IC16 (pin 5) changes state. The edge of this pulse also indicates whether the input signal exceeded the trigger in a positive or in a negative sense.

The output signal of the comparator is applied to a dual four-channel multiplexer, IC17. The first part of this stage enables a choice to be made between the >Q output of the comparator and the external trigger input with the aid of the EXT signal. The second section, IC18, is used to choose between the output of the first multiplexer and the inverted output signal, that is, between triggering at the leading or at the trailing edge of the input signal.

The trigger signal is then fed to bistables FF1 and FF2, where it is combined with the CLK and INH signals. Bistable FF2 also provides the eighth data bit for the RAMs as soon as the circuit is triggered, its output goes high. The eighth data bit is also clocked in bistable FF1, after which it is used as ninth address bit for the memory via multiplexer IC1A. During reading, the circuit is switched via IC1A to the INH signal as ninth address line, whereas during writing, the output of FF1 is used for this purpose.

When the ninth address bit becomes logic 1, the address counter is reset via network R-C3.

Control signals

The connections between the drive

The connections between the drive unit and computer in a row.
The CPU is the clock provided by the computer for reading the RAMs (is applied to IC1ac).

The INH signal prevents triggering of the circuit via bistable FF3 until the RAM has been read completely by the computer.

The computer uses the C/I signal to determine whether the RAM is being written to by the drive unit, or is being read by the computer (logic 0 = reading).

A logic 1 on the EXT line actuates the external trigger input.

Lines O1 to O3 carry the off-set voltage that is added to the input signal via the D-A converter. The standard value should be 1000000 or 01111111 to give an off-set voltage of 1 V (O1 = MSB).

A logic 1 on the AC-DC line causes relay R4 to be energized, so that capacitor C1 is short-circuited and DC components in the input signal are passed on.

The data on the T0 to T3 lines determines the trigger level. To enable triggering at the zero crossing of a signal, the value should be set to exactly half the signal level, that is, 1000000, because of the standard offset of 1 V (T0 = MSB).

The +/− signal enables the computer to choose between triggering at a signal that exceeds the trigger level either in a positive or in a negative sense.

The READY signal indicates to the computer that writing into the RAM has been completed and that the reading process can begin. The internal clock signal is switched off at that instant.

Lines T0 to T3 serve to choose the required time base via IC2. A value of 000000 corresponds to 1 mV/div, and 1111 corresponds to 100 mV/div (T3 = MSB).

Lines V0 to V1 are used for setting the input sensitivity via the multiplexers in the attenuator stages. A value of 0111 gives a sensitivity of 10 mV/div, while 1100 gives a sensitivity of 5 V/div.

Data lines D1 to D7 incl. are used for transporting the binary values from the drive unit to the computer (D7 = MSB). Line D6 carries the trigger bit.

This information makes it possible to write a suitable program for the computer. In many cases that will not be necessary, however, because a complete program listing for the BBC, the Electron, the Commodore C64, and, very likely, MSX computers will be supplied with the printed-circuit board. More about this in Part 2 next month.
Air provides the breath of human life, and the space it occupies around the globe has become the dominant "play volume" in man's defence of his kingdom. The armed forces meet varying threats. The primary concern of navies is high and low flying attack weapons, armies have to contend with strike aircraft and missiles, while air forces are required to intercept bombers and fighters and eliminate all threats that exist in the air space. To maintain vigilance in times of peace and to meet the threats of insurgents in times of war it is vital that all the resources of the armed forces are co-ordinated. Co-ordination is important not only for the air battle but for the safe operation of civil aircraft, and requires close co-operation between the defence organizations and civil aviation authorities.

An air defence system provides this co-ordination. It is able to detect, recognize, and monitor information on objects in the air space. It can provide the command and control of interceptor aircraft, the dissemination of target information to field batteries of guns and missiles, and it can provide the command with a display of a recognized air picture and sometimes of the surface as well. This includes all information necessary for decisions to be made.

Modular advantages

Traditionally, only highly industrialized countries have been able to benefit from large scale air defence systems. Each system would be custom designed resulting in expensive and specialized solutions. Analysis has shown that common functions exist in all the elements of air defence systems around the world and it is this that, for instance, Plessey Radar has capitalized on with its air space management products.

The result is a base of hardware and software models, which include off- and on-line maintenance aids able to be configured at low cost into virtually any size or variety of air defence system by the addition of customer specific facilities.

The modular design approach has additional advantages. Rigorous specifications and methodologies can be applied and the modules subjected to thorough in-service use. It also makes the transfer of technology more amenable and simplifies local maintenance.

Plessey Radar already has a number of such systems installed and an operational development and demonstration air defence facility at its systems headquarters. It is now feasible for any nation to have an air defence system that precisely meets its needs and is affordable — both initially and throughout the life of the system. The next few years should see a large number being installed and integrated on an international scale.

The air picture

The first requirement of an air defence system is to detect the targets. This can be done passively with electronic sensor measurement (ESM) equipment to detect audio, radio frequency, and infrared emission. It can also be done with active sensors such as radar and laser equipment and, of course, that original sensor, the eye, to report visual sightings. All the information from these sensors, whether they be static or mobile, on the land, at sea, or in the air, is processed by reporting posts. The data provided by reporting posts range from crude directional bearings to accurate recognition of the targets along with positional coordinates.

From the reporting posts, the data are transmitted for processing at a control report post. Here, the data are combined with known information, such as civil aircraft flight plans and information from secondary surveillance radar, to provide a track database whose accuracy and completeness determines the quality of the air defence system.

Operators at display consoles in the control reporting post use this track information to select their targets, to control fighter aircraft on their interception missions using air-ground-air radios, or to provide targeting information to SAM (surface-to-
The information is also led to an air operations centre where again it is combined with inputs from other control reporting posts, processed with multi-sensor algorithms, and displayed as a recognized air picture for the whole play area covered by the sensors with each target having a unique track identity.

Defensive network
At the air operations centre, other information is also shown on consoles and large screen displays to enable the command to plan the tactical and strategic air battle. Details of the status and availability of aircraft, guns, and missiles are available along with mission designations and the logistic situation from national and allied forces. The remaining task of the air defence system is to pass the recognized air picture to a central joint operations centre. Here, a totally integrated view can be obtained by combining the information from the air operations centre or centres with similar inputs from navy and army systems to give a complete recognized air and surface picture of the country's defensive network.

The command and control aspects of an air defence system will vary, depending on factors such as the size of the country, the volume of air space to be defended, and the existing defence management organization. There are three fundamental methods of meeting the requirements.

First, command and control can be centralized. Activities undertaken by the control reporting post and the air operations centre will normally be centralized into one facility, often in a single building. For small countries, with, say, six sensors giving coverage of up to 930 x 930 kilometres, this is a suitable air defence system architecture.

Control options
Command and control structures can be kept relatively simple and the number of expert staff required can be limited. The track database — targets that can be effectively monitored — would typically contain up to 300 tracks and controllers could handle up to ten "close" and 20 "loose" interceptions.

Second, where air space management demands a significant number of sensors (more than six) and the defence assets such as guns and aircraft are distributed over large areas, a system with centralized command but decentralized control can be used. Groups of two or three reporting posts pass data to tactical command reporting posts which are normally termed sector operations centres when more than one sensor is connected. These in turn transmit their information to the air operations centre. This is the most common form of air defence system. The dispersed sensors and mobile command reporting posts concept is resilient to damage and the command organization is relatively easy to implement. Typically, each sector operations centre — the decentralized control — can handle 200 tracks and 30 control interceptors while the air operations centre — the centralized command — would have a track capacity of over 500 after combining all the inputs from the sectors. The third method is to distribute command and control responsibility to centres that exercise total authority in specified areas. Each then reports to a centralized strategic command function. This air defence architecture meets the needs when several states are involved or when a large continental area has to be defended. It is an extremely complex task to define the hierarchy of command, to identify and allow for overlaps of data, and to distribute this information around the regions. It requires close co-operation between participating states.

International integration
As well as accommodating the many variants of command and control, an air defence system design must allow for future expansion, not only within the country but also for integration with international air defence facilities.

The data processing system is the key to this versatility. A well-designed system will have modules of hardware and software that can be integrated in various forms depending on the requirements. It must at the same time be tolerant to failure and able to accommodate different computer types and software languages that are likely to be introduced as the system expands and equipment becomes obsolete.

The ideal architecture consists of distributed nodes of processing that are expandable in power. These are coupled via local area network (LAN) open system interconnection (OSI) standard data communications. Computers with applications software can be added to the local area network with no major impact on the logic system. Availability can also be ensured by building-in spare computers coupled with automatic fault detection and techniques that ensure graceful degradations.

The hardware must be capable of installation in mobile cabinets that can be transported by land, sea, or air, as well as in static facilities. Software, likewise, must be capable of being maintained and amended on site to handle local environmental data.

Compatible techniques
Most countries have many sensors, rarely from the same supplier. The other command and control systems with which the air defence network must interface are also likely to be different. It is, therefore, unlikely that the data to be exchanged are of similar format.

The air defence system must use techniques that ensure compatibility at all levels from the reporting post up to the air operations centre while minimizing any impact on the central air defence data handling systems and the other systems with which it is interfaced. The distributed logic system architecture can readily accommodate the special normalizing processors needed to overcome these problems and they can be interfaced to the local area network by the appropriate open system interconnection communications.

* John Nicholls is an Engineering Executive with Plessey Radar Systems • Oakcroft Road • Chessington • Surrey KT9 1GZ.
Voltage comparison on a 'scope

There is frequently a need, when experimenting with circuits, to measure or compare several DC voltages at test points etc. Since most readers are unlikely to possess more than one multimeter this can be rather tedious. Using this simple circuit, up to four voltages can be compared or measured on any oscilloscope that has a DC input and an external trigger socket. The circuit uses only three ICs, five resistors and a capacitor.

The complete circuit of the voltage comparator is given in figure 1. The four voltages to be measured are fed to the four inputs of a quad analogue switch IC, the outputs of which are linked and fed to the Y input of the 'scope. N1 to N3 and associated components form an astable multivibrator, which clocks counter IC3. This is a decade counter connected as a 0 to 3 counter by feedback from output 4 to the reset input. Outputs 0 to 3 of the counter go high in turn, thus 'closing' each of the analogue switches in turn and feeding the input voltages to the 'scope in sequence. Output 0 of the counter feeds a trigger pulse to the 'scope once every four clock pulses, so that for every cycle of the counter the 'scope trace makes one sweep of the screen. A positive-going trigger pulse is available via R4, or a negative-going pulse is available from the output of N4 via R5. The resulting display is shown in figure 2, four different input voltages being fed to the inputs in this case. The oscilloscope timebase speed should be adjusted so that the display of the four voltage levels just occupies the whole screen width.

The supply voltage +U_b may be from 3 to 15 V, but it must be noted that the input voltage should be positive with respect to the 0 V rail and not greater than +U_b. If voltages greater than this are to be measured then potential dividers must be used on the four inputs.

Setting up

To calibrate the circuit, simply feed a known voltage into one input and adjust the Y sensitivity of the 'scope to give a convenient deflection (for example one graticule division per volt input). The unknown voltages may then be fed in and compared against each other and against the calibration.

The circuit can easily be extended to eight inputs by adding an extra 4066 IC and connecting IC3 as a 0 to 7 counter (reset connected to output 8, pin 9).

Figure 1. The circuit diagram of the voltage comparator.

Figure 2. An example of 4 random voltage levels displayed simultaneously on the 'scope.
Following last month's general introduction to satellite TV reception, this article describes the construction and operation of the indoor unit (IDU). This is in essence an interface between the low-noise converter (LNB) at the dish aerial and a conventional television receiver. The first part of the article deals with the RF board contained in the IDU.

Indoor converter for satellite TV

- Single conversion, wideband FM tuner.
- Complies with standard LNB IF range (950...1750 MHz) and downlead feed systems for CS and future DB satellites [1].
- Includes VHF vision & sound remodulator, LNB theft alarm, polarization selector, audio and video outputs, and switchable AFC.*
- Remodulator test and satellite scan circuits simplify initial setting up and dish positioning.*
- Also usable as a 23 cm band (1240...1280 MHz) amateur television (ATV) receiver.

* To be detailed in a forthcoming article.

Before embarking on this project, make absolutely certain that a 1.2 m and 1.8 m dish aerial can be securely installed to give an unobstructed line-of-sight path to the relevant satellite(s). As stated last month, it appears that garden installations are all right, but roof installations require planning permission. Careful planning and expert counsel in this matter will prevent costly and frustrating disappointment at a later stage.

It should be noted that at present it is
virtually impossible for most home constructors to build either the dish, aerial or the low-noise converter, and these will, therefore, have to be bought or rented. In this context, see Satellite TV reception (p.40) and Harrison Electronics' advertisement (p.86) in the September 1986 issue of Elektor Electronics. Fortunately, prices of these units have already started to come down due to the rapidly growing interest in satellite TV reception.

Although the construction of the indoor unit is not recommended to absolute beginners in electronics, it should be noted that a number of prototypes were built by constructors with only limited experience. In the main, the results were fully satisfactory, although all agreed that their task had taxed them to the full, requiring not only great precision and care in soldering, but above all close attention to the constructional details. The present article, therefore, aims at giving the maximum clarity to all matters concerning very-high-frequency techniques. For an explanation of parameters and abbreviations used in this article see Satellite TV reception in the September issue of Elektor Electronics.

**Block diagram**

The block diagram in Fig. 1 shows that the indoor unit is a single-conversion superheterodyne tuner. A low-noise amplifier raises the level of the 950-1750 MHz input from the LNB, which is then mixed with the 1560-2360 MHz output of local oscillators T1 and T2. It should be noted that LNBS used for the reception of communication satellite TV programmes use a 10 GHz local oscillator to give an output of 10.96-11.75 GHz. Fortunately, the European Broadcasting Union (EBU) has recommended (Literature reference [1]) that LNBS for direct broadcasting satellite (DBS) services also have an output of 950-1750 MHz. IF amplifiers T3, T4, and T5, coupled by band-pass filters, provide a gain of about 24 dB at the half-power bandwidth (>36 MHz). A phase-locked loop (PLL) demodulates the 610 MHz IF signal as an IFF and passes to the baseband (about 0.55 MHz) to the video processing circuits (described in next month's issue) via buffer T6. The relatively high IF of 610 MHz ensures good rejection of the 2170-2370 MHz image frequencies. f = f0 + f0.

**Circuit description**

In the circuit diagram of Fig. 2, the SHF input stage, T1, a Type EBF65 transistor, has been designed for low-noise (F1 = 4.5 dB max) wideband operation. It presents a 50-ohm impedance to both the input from the LNB and to mixer MX1. Its gain ranges from about 12 dB at 950 MHz to around 8 dB at 1750 MHz.

MX1 is an HPFS11 monolithic, wideband, double-balanced mixer (DBM) consisting of four Schottky diodes, which have a low junction capacitance and provide linear operation over a wide range of LO and RF power levels. These diodes are fed via high-quality transformers to give a meticulously balanced set-up suitable for operation at high RF, LO, and IF frequencies. The internal organization of the device is shown in Fig. 3a.

The Type HPFS11 was chosen because of its robustness, excellent performance-to-price ratio, and stable impedance at all three ports, which are designed to handle a wide range of RF input signal levels. Its drawbacks are its cost as compared with a discretely built mixer, and its conversion loss. However, an active mixer, which would have some conversion gain, is difficult to keep stable over the RF input range of 950-1750 MHz. Moreover, the passive DBM typically causes the carrier-to-noise ratio to be less impaired in the mixing process.

The characteristic curves in Fig. 4 show some of the parameters of the HPFS11. In particular, Fig. 4c shows the excellent performance of the device at a local oscillator power level of +7 dBm (about 5 mW). Since the input impedance, Z1, at pin 8 is 50 ohms, the output voltage, Vout, of the local oscillator is given by

\[ V_{out} = |P_{in}| = 0.008 \times 50 = 0.4 \text{ Vrms}. \]

Readers interested in balanced RF mixers should find the RF/IF Signal Processing Handbook, Volume I (Literature reference [2]) well worth reading.

Local oscillators T3 and T4 cover the 1560-2360 MHz band at sufficient power for satisfactory operation of the mixer, and have the stability required for wideband FM TV reception. Since it proved virtually impossible to achieve this performance with a single transistor, two varactor-tuned Type BFW92 transistors are used. The two sections of the oscillator, LO1 and LO2, are tuned to the highest and lowest channels of satellite TV services respectively by C14 and C15. Section LO1 covers a range of about 1500-2000 MHz, and LO2 operates over roughly 1800-2400 MHz. The stability of the oscillators is so good that automatic frequency control (AFC) is not, strictly speaking, required.

The relevant oscillator section is selected with S. Resistors R1 and R2 in LO1 and R1 and R2 of LO2 are damping resistors, which provide enough inductance to ensure correct matching to the 50-ohm LO input of mixer MX1.

The common 3-32 V tuning voltage, Vsense, is applied to varactors Di-D4 (LO1) and Di/D4' (LO2) via resistors R3 and R4 respectively. The oscillator stages operate in the common collector mode: oscillation is achieved through positive feedback via the base-emitter capacitance of the transistors.

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Fig. 1. Block schematic diagram of the RF board in the IDU. Note that two local oscillators, LO1 and LO2, have been incorporated to supply the 1560-2360 MHz injection signal.
The 610 MHz IF signal is taken from pin 3 of MX and capacitively fed to a conventional amplifier, $T_2$, which provides about 10 dB gain at a relatively low noise figure; it also ensures correct termination of the IF output at MX.

The first IF band-pass filter consists of two critically to slightly over-coupled tuned line inductors, $L_0$ and $L_1$. Correctly aligned, these have a 3 dB pass band of about 40 MHz, a relatively low insertion loss, and cause minimal stray radiation. Both the collector of $T_2$ and the input of $IC_3$ are capacitively coupled to a low-impedance matching tap on the relevant inductor.

Second IF amplifier $IC_4$ is a wide-band hybrid IC Type OM361, which is primarily designed for VHF/UHF masterhead aerial amplifiers and MATV systems. This single-input (SIL) device contains a 3-stage RF amplifier as shown in Fig. 3b. The OM361 was chosen for its high gain (about 26 dB at 800 MHz) and ease of input/output matching. Power to the final two cascaded transistors is supplied via choke $L_5$ to prevent the RF signal from being short-circuited by the thoroughly decoupled positive supply rail.

Band-pass filter $L_0$ and amplifier $T_3$ have functions and characteristics similar to those of $L_0$ and $T_2$, respectively. The IF signal at the collector of $T_3$ is capacitively fed to phase-locked loop (PLL) decoder $IC_5$.

It must be stressed that the overall performance of the IDU depends to a large extent on the bandwidth, rather than the gain, of the IF chain. Since the deviation of the satellite TV signal is typically $\pm 13.5$ MHz, and the baseband occupies some 5 MHz, the IF bandwidth must be not less than 35 MHz for satisfactory performance.

It is, therefore, clear that the IF band-pass filters are crucial to the correct operation of the IDU. Since the combined gain of the IF amplifiers amounts to 43 dB and that of the IF chain is about 42 dB, it follows that the total insertion loss of the filters is around 5 dB.

Next month's article will contain measurement data relevant to the RF sections of the IDU.

PLL decoder $IC_4$ is a purpose-designed satellite TV FM demodulator Type SL1451 from Plessey and is part of an extensive range of passive and active components intended for satellite reception systems. The functional diagram of the device is shown in Fig. 3c; it is an on-chip voltage-controlled (Clapp) oscillator (VCO).
Fig. 4. Graphs correlating a number of important technical characteristics of double-balanced mixer MX₁. Remember that RF = 950 ... 1280 MHz, IF = 610 MHz and LO = 1500 ... 2400 MHz at +5 to +7 dBm, which is obviously the correct level for lowest conversion loss and highest port-to-port isolation.

**Construction**

Contrary to the normal order in which electronic projects are put together, it is necessary to finish all mechanical work as detailed below, before fitting any parts onto ready-made PCB Type 86082-1.

First, prepare a 160 x 100 x 28 mm (inside) brass or tin sheet enclosure as shown in Fig. 5. If you cannot obtain a preformed enclosure, you will need to cut four suitably sized pieces of 1 ... 2 mm thick brass, drill two of these as shown in Fig. 5a and 5b, and join them to form a neat box using Sellotape at the corners to maintain tight angles as you solder, using a heavy-duty (≥100 W) iron. Brazing is, naturally, even better.

Check whether the eight through capacitors and BNC flange socket K₁ fit snugly into the holes; if not, carefully ream the holes until they do. Do not solder anything as yet.

File a notch into the PCB to allow for the PTFE ring round the centre pin of K₂. Check whether the PCB needs any filing off the sides before it can be received into the box. Fit K₁ by its four small screws, but do not secure these as yet.

Pre-tin the holes provided for the feedthroughs, and insert these from the outer side of the enclosure. Point them downward as you apply heat and solder; if all goes well, the capacitors should slide snugly into place while hot solder runs smoothly round the conical metal bodies. While soldering, carefully manoeuvre the capacitor into its final position.

Since low capacitance (10 ... 27 pF) feedthrough capacitors are difficult
to obtain items, it may be necessary to make a DIY version from a number of parts intended for the isolating of power semiconductors on heatsinks. Fig 5d shows how a small washer, bush, two soldering tags and a bolt plus nut can be put together to act as a low-capacitance feedthrough. It is definitely less elegant than a real capacitor, but it works satisfactorily and has capacitance of about 90 pF. The PCB for this part of the project is a pre-tinned, double-sided type, equipped with 5 mm holes for T1 to T5 incl. and slots for C4 and C6. Through-plating is effected by soldering component leads at both PCB sides, where required. Start off by applying some solder onto all ground holes on the PCB, as well as onto its edges at both sides; this will facilitate soldering at a later stage, and prevents overheating of grounded components when these are fitted. Make sure, however, that holes remain open (use solder wick).

Resistors: with a few exceptions in the LO sections of the circuit, these should have their leads neatly bent equidistant from the body with snipe-nose pliers. Pre-tin any resistor lead that is to be inserted into a ground hole. All resistors should be \( \frac{1}{4} \) or \( \frac{1}{2} \) watt (except R4, which is \( \frac{1}{2} \) W) carbon film types; not metal film. Resistors should be fitted to rest securely on the PCB component side.

Capacitors: In the case of a supply decoupling capacitor (1 nF, 10 nF, 22 nF, 47 pF and 10 pF), pre-tin the ground lead close to the capacitor body. With some types of 2.5 mm type ceramic capacitors, it may be necessary to carefully remove some of the brittle material on the wire where it leaves the capacitor body; pre-tin as fast as possible, holding the far end of the lead in pliers. When soldering the ground terminal at the PCB component side, solder can be observed to creep right up to the ceramic body, and spread smoothly over the ground plane.

Coupling capacitors do not require this method of pre-tinning, although they should be mounted with the shortest possible lead length as well. Trimmers are to be pushed securely into the relevant holes and soldered rapidly to prevent deforming of the foil material.

Transistors: With the exception of the BFW92s and the BC547B, b and c leads should be cut off to about 2 mm, e leads to 3-4 mm. Before fitting, note the terminal assignment of the BFG65 to get its position correct. Transistors T1 and T2 should be mounted at the EPS side of the PCB, straight onto the relevant tracks and with the type lettering visible from the component side. The e leads should be soldered flush with the EPS side ground plane.

Inductors should present few problems, as their construction data and practical outlook are given in Table 1 and Fig. 7 respectively. Note that only two types of wire are required to make all inductors, except L5, which is a commercial choke. The silver-plated tuned lines should be accurately bent and, with the exception of the longer L3, pre-tinned at one end.

With reference to the component overlay and track pattern shown in Fig. 6, and observing the foregoing directions, the fitting of parts onto the PCB may now commence.

RF amplifier T1 and mixer (see also Fig. 9)

Fit all passive parts as set out above. Pay special attention to LNB block inductor L3, which is fitted slightly off the board surface, and has one end soldered directly onto the RF input plane. Fit T1 in its pre-drilled hole, soldering the b and c leads onto the relevant tracks, the e leads firmly to ground. Solder SMD capacitors C1 and C4 with a light-duty (15 W) iron to prevent damaging these devices. Alternatively, C1 and C4 may be 6p8 ceramic types mounted onto the relevant planes with the absolute minimum of lead length (\( \leq 0.5 \) mm). L2 and R1 must be

<table>
<thead>
<tr>
<th>Inductor</th>
<th>turns</th>
<th>SWG wire</th>
<th>internal diameter</th>
<th>remarks</th>
</tr>
</thead>
<tbody>
<tr>
<td>L1</td>
<td>12</td>
<td>24 enam.</td>
<td>3 mm</td>
<td>closewound.</td>
</tr>
<tr>
<td>L2</td>
<td>1</td>
<td>24 enam.</td>
<td>–</td>
<td>through 3 mm ferrite bead.</td>
</tr>
<tr>
<td>L3, L4</td>
<td>–</td>
<td>20 silv.</td>
<td>–</td>
<td>tuned line; length and location of tap governed by relevant PCB holes; fit 3 mm above ground.</td>
</tr>
<tr>
<td>L5, L7</td>
<td>–</td>
<td>20 silv.</td>
<td>–</td>
<td>as above but no tap.</td>
</tr>
<tr>
<td>L8</td>
<td>5</td>
<td>24 enam.</td>
<td>R1</td>
<td>closewound on R1.</td>
</tr>
<tr>
<td>L10, L10'</td>
<td>1/2</td>
<td>–</td>
<td>3 mm</td>
<td>spacing 1/2 mm initially; see Fig. 8c.</td>
</tr>
<tr>
<td>L11, L11'</td>
<td>–</td>
<td>–</td>
<td>–</td>
<td>tuned line; see Fig. 8c.</td>
</tr>
</tbody>
</table>
Fig. 6. Component overlay and track pattern of the IDU RF board. Note that the local oscillators incorporate components not fitted onto the board in the usual way. Not shown are feedthrough capacitors C49 to C44 incl., which are fitted into the lower side panel of the enclosure.

Fig. 7. Neatly made inductors are the key to satisfactory performance of the IDU RF board. Showing a number of parts mounted onto the PCB. Left to right, top row: PLL loop inductor Ls wound on R5, L6 on a small ferrite bead, a 1 nH feedthrough capacitor, and trapezoidal chip capacitors C49 and C41. Second row: + LNB block coil Ls, silver-plated strip lines Ls and soldered as close as possible to the transistor body (b and c terminals respectively); note that R5 may have to be mounted slightly asymmetrically to ensure minimal stray inductance at the transistor base. MX is located at the EPS side of the PCB, while its eight pins are soldered at the component side. Note that the RF input (pin 1) is marked in blue for location purposes.

IF amplifier.
Solder the MX+ connection of C7 at both PCB sides, but that to the base of T5 at the EPS side only. Tuned lines Ls, Ls, Ls and Ls are best fitted as follows (see also Fig. 8b). Insert a left over component wire into the PCB holes provided for the taps, and solder at the EPS side. Use the protruding pin at the far end of the vernier gauge handle to determine a wire length of 3 mm above the component side ground plane; cut the wire and level its top with a few strokes of a small file, while the wire is held securely in pliers. Pre-tin the top and position the wire at right angles to the PCB. Mount the silvered line, pushing it into place until it rests on the tap wire end. Make sure that the inductor is precisely angled and that its horizontal part is always 3 mm above ground. Solder the trimmer and double ground connections, and then the tap. Remember that any excess solder on its silver plated surface may degrade the inductor’s Q-factor. Make sure that the coupled lines run parallel and at identical height above ground.

After inserting the pins of the OM361 until all studs rest on the PCB surface, they must be soldered rapidly (five pins twice to ground), after which the SFL chip must be bent downward with its type indication facing the PCB component side ground plane. Do not use too much force, or one or more of the pins may then break off.

Fitting the remainder of the IF amplifier components should not cause difficulty, as the suggested methods for mounting have already been detailed above.

PLL and baseband output.
Mount IC2 without an IC socket, and remember to solder pins 2 and 8 at both PCB sides. The surrounding capacitors and resistors should be fitted as set out, while block inductor R5+Ls must be mounted at a small distance above the board (1 mm) to prevent any likelihood of a short-circuit. Tuned line Ls is fitted at precisely 3 mm above ground. Mount varactor D0 with the minimum of lead length at either side of its glass body. Make sure that it is really a BB405G.
it should have a green and a white ring, the latter indicating the cathode connection.

**Local oscillators** (see also Figs. 8c and 9). You are now well on the way towards completing the RF board, but the toughest part is yet to come: no PCB holes in many cases, and a few parts mounted three-dimensionally; and yet, it is not as difficult as it may seem at first.

Note that all part references in the following description also apply to the corresponding accented (*) parts, unless a specific description is thought necessary to make a distinction.

Fit decoupling and bias parts R1s, R1s, R1, C1s, C1s, and C1s as set out above. Pay due attention to the fitting of T3, as it has neither a hole nor any tracks to connect to other components. As illustrated in Fig. 8a, the transistor's collector lead is to be sharply bent where it leaves the enclosure. Push-fit the lead into the slot, along with chip capacitor C1s, until the latter's shoulders rest firmly on the PCB surface. Gently manoeuvre the decoupling capacitor and tap the transistor until this is felt to lie level onto the PCB surface.

*Note that emitter leads of T3 and T3' should face one another, requiring T3, unlike T3', to lie with its type indication facing the PCB ground plane.*

Carefully solder the track and double ground connections of the chip capacitor, and make sure that solder creeps up along the metalized area and the collector lead, whose excess length is then cut off. Shorten the transistor's e and b leads to 2 mm and pre-tin them. Fit stopper resistor R1 with the shortest possible lead length (≈1 mm) as close as possible to the transistor body (b); this may require the other lead (to junction R+R+T+T) to be rather longer than usual, but this is of no consequence. Sharply bend the anode lead of varactor D4, pre-tin, and solder to ground (3×) using the hole provided. Note that the ground connection of D4 (LOU) is closer to T3 than that of D4 to T3 (LOU). Shorten the D4 cathode lead to 2 mm, pre-tin, and do the same with D1. Carefully join these parts and run the appropriate length of the D4 anode wire to junction R+T+. Since junction R+D-D should exhibit the absolute minimum of stray capacitance.

**Fig. 8 Artist's impression of suggested component mounting methods.** Fig. 8a shows how the T3 and T3' collector leads, along with chip capacitors C1s and C1s, are push-fitted into 6×1 mm slots in the PCB. Fig. 8b shows that very little can go wrong if the tuned line inductors are neatly bent to suit the relevant PCB holes. Fig. 8c, finally, should be used as a guide in constructing the "three-dimensional" local oscillators. Stick to these guidelines and your IDU RF board will work!
and inductance, $R_s$ needs to be prepared as follows. At one side of the resistor body, the protective lacquer should be scratched off where the wire leaves the body. This is conveniently done by holding the relevant area in pliers and twisting the resistor until the brittle stuff comes off. Shorten the lead to 0.5 mm, prepare, and join it to junction D1-D4 with a minimum of solder. Note that the other lead of $R_s$ (and of $R_v$) must be left much longer, so that it can reach junction $R_s$-$R_v$-$C_w$. Since these resistors act as current limiters and chokes to the SHF signal on the varactors, this length is of little importance.

Inductors $L_w$ and $L_v$ are made from the terminal leads of $R_w$. One lead is wound as $1/2$ turns on a 3 mm former, which may be nail, screwdriver shaft, or even a ball-point refill, as long as it has a diameter of 3 mm. Leaving the turns to revolve around the former, the resistor is gently pulled back until the lead length between resistor body and start of winding matches that given in Figs. 7 and 8c. Space the turns as shown. The other resistor lead is to act as $L_v$. Observe its length, and edge the remainder of the wire two times as shown in the illustrations. Put the prepared resistor & inductors aside for the moment, and proceed with the most exotic, yet simplest, part of the board: $C_w$, which is simply some 10 mm of left-over component wire, 2 mm of which is slightly bent, soldered to the $T_e$ lead, and pointed towards $C_v$. The wire should not touch ground, of course. Solder $L_v$ to junction $C_w$-$T_e$; this requires some skill to prevent short-circuiting the inductor turns by either $C_w$ or the $T_e$ emitter lead. Check for any short-circuited components by excess solder, and carefully bend $C_w$ point to the body of $C_w$. The oscillators will not operate correctly if $C_w$ is left out.

Run $R_w$-$L_v$ exactly parallel to $D_1$-$D_4$, and solder $L_v$ to ground, straight onto the PCB surface. Note that no ground hole has been provided; use the relevant illustrations and the component overlay to find the correct location, level with $X_1$ pins 7 and 8. $R_w$ should now be positioned well above all other components. Solder $R_v$ very close to $R_s$ and run the other end direct to mixer pin 8. As $R_v$ should have exactly the same total length as its $L_v$ counterpart, the $L_v$-$R_v$-$L_v$ line needs to be mounted slightly slanting with respect to the D1-D4 line. Ground $L_v$ at the appropriate location, and check the cutout of the LO sections against Figs. 8c and 9.

PCB
Recheck all soldering joints at both sides of the PCB, and remove any stray bits of wire or solder. With a sharp appliance and a cotton bud dipped in 95% alcohol, remove all excess solder flux visible as brownish matter, from etched surfaces in the RF input and mixer stage; do the same at the PLL section. If you have so far followed the instructions, terminal holes $1-8$ incl. should still be open.

Enclosure
Fix $K$ securely by its four screws, whose heads should be at the inside of the enclosure. File off any protruding thread until it is flush with the socket flange. Insert the completed board into the enclosure, making sure that the centre pin of the BNC socket rests on the RF input plane ($L_v$-$C_v$) file or cut off any excess pin length. Refer to Fig. 5b for the positioning of the board and make sure that the bottom lid can be pressed or screwed on without touching MX1. Use a heavy-duty iron ($>50$ W) to solder the PCB into the enclosure; depending on the type of metal sheet, some pre-heating may be called for to be able to solder at all. Use an additional soldering iron or place the enclosure on the hot surface of a thermostatically-controlled smoothing iron; you will find that once the metal surfaces are reasonably warm to the touch soldering becomes much easier.

Mount eight soldering pins in the terminal holes if the wires of the feedthrough capacitors are not long enough.

Using the dotted lines on the PCB overlay as a guide, solder three 17 mm high metal screws onto the PCB component side (take care not to damage nearby parts). Note that the longest screw is to run right over IC2, so that a 20 x 4 mm recess hole should be made at the correct location.

If you have made your own metal enclosure, do not forget the top and bottom lids, which are to be screwed on after the box has been fitted with at least eight square brass nuts, soldered into the upper corners. A few additional nuts and screws along the enclosure side panels are, of course, good practice to make for an RF-tight unit. Finally, drill the top lid as shown in Fig. 5c.

Next time
Part two of the article in next month’s issue will describe details of the vision and sound processing circuits, the power supply, and the $S$-meter driver. Also, the alignment of the IDU will be gone into, and measurement data relating to its performance will be presented and discussed.

Literature references:

Important notice
Information on component availability for this project will be given in Part 2, to be published in next month’s issue. Meanwhile, many parts are available from Bonex Limited; 102 Churchfield Road; Acton; London W3 6DH; telephone 01-932 7248; Universal Semiconductor Devices; 17 Granville Court; Granville Road; Hornsey; London N1 4EF; telephone 01-438 9420; or Cirkit; Park Lane; Brxboune; Herts EN10 7QN; telephone: (0992) 444111.

11-34 elektor india November 1986
IDU for satellite TV reception
Within the next ten years, the European Space Agency (ESA) will be operating an autonomous space station known as Columbus. It will comprise manned modules and laboratories, free-flying platforms, and other equipment launched by the United States' Space Shuttle and the proposed European Ariane 5 rocket. It will be serviced by the European manned spaceplane, Hermes, and possibly by the British unmanned HOTOL shuttle.

As a member of ESA, the United Kingdom will play an important role in these projects, not only with HOTOL but also particularly the free-flying platforms for Earth observation, astronomy, and materials processing. Britain’s financial contribution to ESA is 12.9 per cent and its participation in mandatory ESA programmes represents 16 per cent of the effort and manpower.

There is a thriving space industry in the United Kingdom and this has been involved in the building and operation of over 50 satellites, 2000 sounding rockets, and over 100 ground stations. Unlike that of many European countries, British influence in space technology and commerce spreads beyond the continent itself, and especially to the United States of America.

British Aerospace (BAe) is the largest manufacturer of communications satellites outside the United States and is a major contractor to American companies on a number of key projects. Although BAe tends to be the most familiar, several other companies are also actively engaged in the space industry on a large scale.

More versatile

These include GEC-Marconi for communications payloads and ground equipment, while others involved are Ferranti, Logica, Klystrons, Centronic, Thorn-EMI, Racal, Software Systems, BAe Vickers, and IMI Summerfield. These manufacture and supply other ground equipment, microwave tubes, detectors, solid state devices, digital data recorders, software, gyro packages, and rocket motors. BAe pioneered use of three-axis controlled communications satellites, which are more versatile and can be built on a much larger scale for the multitude of developing applications, such as direct broadcast television, mobile and maritime communications, and business services. The company has developed a stable of communications satellite platforms: the European Communications Satellite (ECS), Eurostar, and Olympus. The ECS 2-Eutelsat series satellites being built by BAe have 12000 voice circuits, two television channels, and two repeaters to handle business traffic. They have a 1 kW capacity.

Three military communications satellites known as SkyNet are being built, too. These are also based on the ECS "bus" and include a communications payload from GEC-Marconi comprising four super high frequency channels, two ultra high frequency channels for voice data and telex, and one extra high frequency experimental uplink. Power generation is 1.25 kW.

Aircraft phone calls

Three Eurostar satellites are being built for the London-based international maritime organization, Inmarsat, on a contract worth $100 million. Producing 0.75 kW of power, the Inmarsat 2 satellites provide independent L band ship-to-shore and C band shore-to-ship communications, with 250 and 125 voice circuits respectively. Eventually, they will also provide aircraft communications enabling passengers to take and make telephone calls on civil aircraft.

The Eurostar satellites can generate as much as 2.3 kW of power but this is weak compared with the world’s largest communications satellite, Olympus, one model of which can generate 8 kW, enough to power 12 direct broadcast television channels.

The next British Aerospace satellite under development is called the Big Communicator. The concept envisages clusters of powerful communication satellites (comsats) sharing geostationary orbit, providing television broadcast and fixed and mobile communications services. Inter-satellite laser links will allow communications within and between clusters via gateway satellites. Three versions of the Big Communicator are planned, the largest being for direct broadcast television. This would generate 15 kW from a 50 m long solar array and carry 16 high power television channels. BAe also has a contract to build equipment for Intelsat 6, the next generation of satellites for the international telecommunications organization.

Spectacular data

The company produced significant Spacelab hardware which flew on Space Shuttle missions dedicated to Europe, the United States, and West Germany in particular in 1983 to 85. Twenty of the Spacelab pallets have been delivered to America’s National Aeronautics and Space Administration (NASA). These are used as
the essential mounting points for equipment in the Shuttle payload bay. The European spacecraft Giotto, which had its rendezvous with Halley's comet last March, returning spectacular data, was built in Britain with BAE as the main contractor. The $45 million contract is just one of a number of key science and applications satellites operated or planned for Europe and to be built using British expertise.

These spacecraft include the Ulysses international solar polar spacecraft, the European Remote Sensing spacecraft (ERS 1), Exosat, International Ultraviolet Explorer (IUE), Infrared Astronomical Satellite (IRAS), and the Geostationary Orbiting Satellite (GEOS 2).

The solar panels that will generate electrical power for the giant Hubble space telescope, hopefully to be launched by Shuttle later this year, were manufactured by BAE as a huge foldable array. The company expects to build a set of replacement panels under a $7 million contract. Britain is investing $58 million in joint science and industry programmes covering remote sensing, data acquisition, processing, dissemination, and forecast. The Royal Aircraft Establishment's National Remote Sensing Centre at Farnborough co-ordinates this activity.

In the meteorological field, GEC-Marconi is developing Europe's first advanced microwave sounding unit to fly on a United States' National Oceanographic Administration Agency (NOAA) satellite in 1990. Britain is a world leader in satellite instrumentation for remote sensing spacecraft.

Close relationships

The British National Space Centre (BNSC), formed in 1985, will in future co-ordinate the country's burgeoning space industry. Based in London with a small staff, it will formulate a national space policy to be presented to the Government in June 86. To be effective, it should cover the next 15 years. The BNSC will need to establish close relationships with industry, including the non-aerospace sector, both in contractual development and exploitation, and commercial space operations. It needs to provide a coherent voice on space matters, seeking comprehensive rather than a fractional approach, and to consider its role in education and public policy.

The centre's director-general is Dr Roy Gibson, who, as the European Space Agency's own first director-general, helped establish it and develop Europe's prestige in space. It is expected that Britain's space budget under his control will be doubled over the next two years to about $300 million to reflect the increased importance the country places on space. Although the United Kingdom only plays a small part in the Ariane launcher programme — Avra, Badg, Ferranti, and Midland Bank's share represents just 2.4 per cent — BAE has proposed a revolutionary launcher to beat them all, including the proposed Ariane 5. HOTOL, an initially unmanned spaceplane, will be the world's first single-stage-to-orbit (SSIO) satellite launcher and the first to take off and land like an airliner. It will cut by half the cost of deploying satellites into orbit. Indeed, the measure may be greater than that. BAE says it could place a five tonne payload into low Earth orbit at a fifth the cost of current vehicles.

Complementary or replacement

The revolutionary engine for HOTOL is designed by Rolls-Royce and will be dual-functioning. It breathes outside air like an ordinary airliner and mixes it with on-board supplies of liquid hydrogen during the initial climb through the atmosphere. HOTOL then switches to internal fuel supplies of liquid hydrogen and liquid oxygen, once the air at altitude gets too thin to be usable. It is expected to cost $520 000 million to develop, and whether it goes ahead will depend on whether it is accepted by Europe as a complementary vehicle to Ariane 5 and Hermes, or even as a replacement for Hermes which does not meet with universal approval within Europe. So far, only about $5 million has been forthcoming for a proof-of-concept study and Dr Gibson hopes to be able to present the HOTOL case to ESA before the end of this summer. Although the French Hermes manned spaceplane has more short-term support, HOTOL is conceived as compatible with the United States' Space Station and its eventual European counterpart. It will also be manned for some sorties. Ultimately, visionary engineers see HOTOL as the successor to Concorde, carrying a passenger pod in its payload bay on a journey between London and Sydney, Australia, in 67 minutes.

Enormous potential

Like France, the Netherlands, West Germany, and Italy, Britain has a squad of astronauts, or more correctly, Space Shuttle payload specialists. Two of these are due to fly Shuttle later this year and early in 1987, primarily to help in the deployment of SkyNet 4A and 4B military communications satellites. They will begin British experiments into microgravity processing. The potential of this business is enormous and the BNSC is anxious to educate British industry as to its possibilities. Kodak Ltd, a subsidiary of Eastman Kodak in the United States, has already flown experiments on the Shuttle and a fluid physicist from the company may be joining a crew in 1987 to 88 to operate his own experiments.

Clearly, commercial operations are some way off, perhaps 20 years away, but vital research and development work needs to be done in space now. This is an area where Britain has been slow to move but it has the capability to catch up with France and West Germany, which are already forging ahead in this field.
The fourth in this series on RF circuit design* describes a superregenerative short-wave receiver that can be coupled to a frequency counter for an accurate read-out of the frequency of the received signal.

**superregenerative short-wave receiver**

A superregenerative receiver is provided with ample positive feedback so as to be capable of oscillation at the desired radio frequency. It is also provided with a means by which oscillations can be stopped or started at will. During normal operation, the relevant circuit is just oscillating.

**Block diagram**

From the block diagram in Fig. 1 it is seen that the RF signal intercepted by the aerial is fed to an RF stage, which not only serves to amplify the signal but also to decouple the aerial from the remainder of the receiver. The amplified signal is fed to a buffer and a detector stage. The output of the buffer may be used to drive a frequency counter to give a read-out of the received frequency. The demodulated output from the detector is passed through a low-pass filter with a cut-off frequency of 5 kHz and then applied to an AF amplifier. The audio output is sufficient to drive a pair of headphones, but may also be used to drive a more powerful AF amplifier.

**Table 1. Winding data for L2**

<table>
<thead>
<tr>
<th>Band (m)</th>
<th>L2A (turns)</th>
<th>tap at (turns)</th>
<th>L2B (turns)</th>
</tr>
</thead>
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<td>120</td>
<td>132</td>
<td>12</td>
<td>7.5</td>
</tr>
<tr>
<td>90</td>
<td>99</td>
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<td>75</td>
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</tr>
<tr>
<td>11</td>
<td>12</td>
<td>1</td>
<td>1</td>
</tr>
</tbody>
</table>

The core is a Type T50-6 RF toroid available from Cirkit (telephone 0992 444111) or Bonex (telephone 01 902 7748), while the winding wire is 0.3 mm dia. enamelled copper.

**Circuit description**

With reference to the circuit diagram in Fig. 2, the aerial signal is applied across potentiometer P1, which enables the signal to be set to the correct level, as will be explained later. MOSFET T1 amplifies the input signal and decouples the aerial circuit. The amplified signal is applied to a detector, the G2-D junction of T1, via circuit L-C1-C2-C3-C4, which is tuned to the frequency of the incoming signal.

Part of the RF signal is applied to the G1-D junction of T1, from where it is fed back inductively to the tuned circuit. As this feedback is positive, oscillations tend to be set up in the tuned circuit at the frequency of the received signal. These oscillations are quenched by the resistance of P2, depending on its setting, so that this potentiometer affords a means of bringing the tuned circuit just into oscillation. The demodulated output at the source of T2 is applied to low-pass filter L3-C1-C4-C5, which has a cut-off frequency of about 5 kHz. Since many short-wave stations operate at 5 kHz channel separation, the filter provides effective adjacent-channel suppression.

The audio signal is then amplified in T3 and T4 whose gain is sufficient to enable a pair of high-impedance headphones to be driven from the AF output across C1-C2. If the audio output is used to drive an additional AF amplifier, the value of C1 should be reduced to 1 μF. The signal at the drain of T4 is also fed to buffer T5, whose output may be used to drive an external frequency counter. This is a very useful means of obtaining a read-out of the frequency of the received signal, which makes operation of the receiver immeasurably easier.

* The first three appeared in the March, April, and May issues of Elektor India.
Construction

The receiver is constructed on the Universal RF Board Type 85000, which is available through our Readers’ Services. As it is an un-pierced copper-clad board with fifty-seven isolated islands and three isolated tracks, it is also available from most electronics retailers. A suggested component layout is shown in Fig. 3.

Chokes L2a and L2b are commercially available components, but inductor L3 must be wound as shown in Fig. 2. The number of turns for the various short-wave bands are given in Table 1. It is imperative for correct operation of the receiver that the coils are wound in the direction shown and that correct polarity is observed (this is facilitated by the large black dots in the circuit and on the coil drawing).

Operation

For optimum performance, the G-D section of T2 should just oscillate. This is achieved by setting P2 to roughly its centre of travel and adjusting C1 till oscillations just occur: this is indicated by a whistle in the headphones or loudspeaker.

The input level is then set with P1, if this is too high, cross modulation occurs, i.e. apart from the wanted station, others are also audible. If the aerial signal is too weak, the detector does not operate correctly, and the signal is hardly audible.

It may be necessary to adjust P3 slightly before optimum performance is achieved: only when this is so, does the frequency counter indicate the frequency of the received signal.

Parts list

Resistors:
\[ R1 = 100 \text{ k} \]
\[ R2 = 27 \text{ k} \]
\[ R3 = 100 \text{ } \Omega \]
\[ R4 = 470 \text{ } \Omega \]
\[ R5 = 82 \text{ } \Omega \]
\[ R6 = 220 \text{ } \Omega \]
\[ R7 = 1 \text{ k} \]
\[ R8 = 56 \text{ } \Omega \]
\[ R9 = 68 \text{ } \Omega \]
\[ P1 = 1 \text{ k linear potentiometer} \]
\[ P2 = 5 \text{ k linear potentiometer} \]

Capacitors:
\[ C1 = 100 \text{ pF} \]
\[ C2 = 10 \text{ n ceramic} \]
\[ C3 = 10 \text{ n ceramic} \]
\[ C4 = 10 \text{ p trimmer} \]
\[ C5 = 1 \text{ p} \]
\[ C6 = 10 \text{ p NPO} \]
\[ C7 = 47 \text{ p} \]
\[ C8 = 22 \text{ p NPO} \]
\[ C9 = 100 \text{ p variable capacitor} \]
\[ C10 = 10 \text{ n} \]
\[ C11 = 22 \text{ n} \]
\[ C12 = 330 \text{ n} \]
\[ C13 = 10 \mu F/16 \text{ V} \]
\[ C14 = 47 \text{ n} \]
\[ C15 = 47 \mu F/10 \text{ V} \]

Semiconductors:
\[ T1 = 2N2907 \]
\[ T2 = 2N2907 \]
\[ T3 = 2N2907 \]
\[ T4 = 2N2907 \]

Miscellaneous:
\[ L1 = 1 \text{ mH} \]
\[ L2 = \text{ see text and Table 1} \]
\[ L3 = 100 \text{ mH} \]

Fig. 3. Suggested component layout of the short-wave receiver.
HOW MUCH LONGER WILL SILICON BE USED?

It sounds rather strange, against the background of the present development of microelectronics, to ask how much longer silicon will be used. The first quantities of one-megabit dynamic memories using existing silicon technology have been announced recently while four-megabit dynamic memories are expected in 1988. These are the most outstanding current examples of the state of the art of silicon microelectronics. These developments in large-scale integration (LSI) have been due to process technology or, to put it the other way around, it was mastery of process technology that made this progress in large-scale integration possible. A reduction in costs per bit on an integrated device went hand in hand with this large-scale integration. This is demonstrated by Fig. 1, which shows the evolution of costs per bit for the various generations of dynamic RAMs as "learning curves". The learning curves for one-megabit and four-megabit dynamic RAMs are estimated values. Before turning to the question of the limits of silicon technology and its replacement by gallium arsenide, we shall first briefly outline the development of silicon technology. By the standards of microelectronics, silicon technology is a "very old" technology. It was 25 years ago, in 1961, that the first IC was developed by Kilby in germanium and one year later in silicon. This process led in only 25 years from a small number of transistors on a chip to more than one million transistors in regular logic devices on the one hand and to more than a hundred thousand transistors on a chip in non-regular logic devices on the other hand. In other words, the complexity of the circuitry has increased by more than a hundred thousand times in this period of time. After these developments, is a competitor now appearing on the horizon in the form of gallium arsenide? The worldwide market potential of gallium arsenide is estimated at 3.2 billion dollars for 1992, a considerable amount when one considers that, e.g., the German microelectronics market was worth about one billion dollars in 1985. Against this background, one might after all be justified in asking how much longer silicon will be used. In order to answer this question we shall consider the following points:

- the mechanisms of substitution which result in the replacement of a technique or technology by another;
- the limits of silicon;
- the limits of integration techniques;
- the development of the market for silicon and gallium arsenide.

**Mechanisms of substitution**

A technique or technology is only replaced by another under the following conditions:

- Techno-economic limitations of a technique become apparent, i.e. substitution results in cost savings.
- A faster evolution of an alternative technique is expected and at the same time a tendency towards greater efficiency. In such a case a substitution is frequently made as a future investment.
- As well as the actual replacement of the existing technique, a new technique promises completely new applications. A substitution is made with a view to innovative potential.

**Limits of silicon**

In order to assess the limits of silicon and possibilities of the alternative material gallium arsenide, it is first necessary to consider the physical properties and also the technological status of the two materials. A comparison of the physical properties of the two basic materials reveals three salient factors:

- the much greater electron mobility of GaAs, which means that considerably faster circuits can be realized with GaAs;
- the much greater thermal stability of GaAs and greater resistance to radiation, which would be of particular advantage with very fast and highly integrated memories;
- a worse ratio of electron mobility to defective electron mobility in the case of GaAs, which also means that complementary electronics can be less easily used in GaAs than with silicon. The physical properties

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Fig. 1. The evolution of relative costs per bit for dynamic MOS-RAMs.
Table 1

<table>
<thead>
<tr>
<th>Technological status</th>
<th>Si</th>
<th>GaAs</th>
</tr>
</thead>
<tbody>
<tr>
<td>Chip diameter</td>
<td>6&quot;</td>
<td>3&quot;</td>
</tr>
<tr>
<td>Fault density</td>
<td>(&lt; 10 / \text{cm}^2)</td>
<td>(&gt; 10^2 / \text{cm}^2)</td>
</tr>
<tr>
<td>- chip uniformity</td>
<td></td>
<td></td>
</tr>
<tr>
<td>- chip purity</td>
<td></td>
<td></td>
</tr>
<tr>
<td>- surface smoothness</td>
<td></td>
<td></td>
</tr>
<tr>
<td>Chip surfaces</td>
<td>(&gt; 50 \text{mm}^2)</td>
<td>10 - 15 \text{mm}^2\</td>
</tr>
<tr>
<td>Components / IC</td>
<td>10^2</td>
<td>10^4</td>
</tr>
</tbody>
</table>

only represent one side, however. In order to make a final judgement we also have to take into account the state of the art in the two technologies. This has been done for silicon and GaAs in Table 1. If we look at this table, we see that the fault density for silicon chips is more than a thousand times less than for GaAs. This is due to a considerably greater uniformity, purity, and surface smoothness in the case of silicon chips; in other words, as a starting material silicon can be much better controlled than GaAs, which in turn results in far greater efficiency. We can also see that silicon chip surfaces are now more than 50 \text{mm}^2 in size, compared with 10-15 mm^2 for GaAs, in other words considerably larger and more complex ICs can at present be fabricated with silicon. On the basis of this table, it can be said that GaAs is at present technologically about a hundred times behind silicon in complexity, or more than two generations of components behind. The same conclusion is reached if one considers the evolution of the complexity of integrated circuits, as shown in Fig. 2. The thick line represents the evolution of the complexity of silicon circuits and the thin line the evolution of GaAs circuits. We can see how silicon has evolved to the four- and 16-megabit dynamic RAM, while GaAs has developed to the four-kilobit RAM. Fig. 2 does not show the production status of these circuits but the time at which the first design models were presented. If we look at the two curves for silicon and GaAs, we have to conclude that, even if we assume a more rapid development for GaAs than for silicon, it will not attain the degree of complexity of silicon until 1995. Such a rapid development of GaAs is not to be expected and we should assume that the broken line with shorter strokes is more probable, so that even in the year 1995 we can expect a difference in complexity of more than ten components per chip and per \text{mm}^2. This means, therefore, that not only the individual components on the chip have become faster, but that the total chip sizes and number of components have grown very rapidly. At present chips are produced which are 40-50 \text{mm}^2 in size, while chips up to 100 \text{mm}^2 are being developed and will be produced in 1987. This implies that from 1988 chips between 50 and 100 \text{mm}^2 will represent the state of the art. At the same time, the length of the circuit on such chips will also increase, so that a length of 10 mm on a chip of approximately 80 \text{mm}^2 will not be exceptional. If, however, we wish to determine the propagation delay on a circuit which is 10 mm in length and assume a value of \(10^{10}\) cm/s for the signal propagation, we obtain propagation delays of 0.1 ns. This means, therefore, that with chips whose geometry is smaller than 1 \mu m and with chip surfaces of 100 \text{mm}^2 the speed of the components and the propagation delay between the components are in the same order of magnitude. From the above observations, it can be deduced that with highly integrated circuits it is no longer the properties of the components on the ICs which determine the speed of signal processing, but that the arrangement of the circuitry has a major influence. This also applies to GaAs. If, therefore, we substituted GaAs for all the silicon

![Fig. 2. The evolution and complexity of integrated circuits.](image-url)
components on a highly integrated circuit of this kind, we would scarcely alter their speed, as this is largely determined by the propagation delays between the individual components.

**Limits of integration techniques**

These observations show that in highly integrated circuits the speed cannot be the decisive factor for a changeover from silicon to GaAs and hence for the replacement of silicon technology. Hence, an assessment of whether silicon is likely to be replaced can only be made if we first answer the question concerning the limits of silicon technology. We must first clarify, therefore, whether silicon is more likely to come up against technological limits than GaAs. The question concerning the limits of silicon technology raises, however, questions concerning the general limits of integration techniques, which basically exist independently of whether one uses silicon or GaAs as the starting material. They do very much depend, however, on the smallest geometries that can be used and on the process technology required for these.

If, for example, we consider the yield loss in 64 K dynamic memories, approximately 70 per cent is due to random defects, i.e., defects which are caused by the lithographical and deposition processes. About 10 per cent is correlated to the structure, in other words is due to geometrical tolerances, and only about 20 per cent is connected with device parameters, i.e., directly linked with the processing of the silicon. The physical limits for vertical and lateral structures are approximately 0.05 μm for bipolar transistors and approximately 0.1 μm for MOS transistors. The limits are set by the minimum dimensions of the space charge regions of the pn junctions, which at room temperatures and voltages of 1 V are approximately 0.03 μm.

For the limits that can be reached in practice, we can make the following estimates. For lithography and etching techniques it is not possible in terms of production to go below structures of 0.1 μm. Insulation requires geometrical distances of about 0.5 μm. In order to keep the contact resistance at a tolerable level for components of less than 100 ohm, the contact holes should likewise have a diameter of more than 0.3 μm. For the pitch, by which is meant the line width plus the spacing between it and the next line, it will not be possible to go below 1.0 μm, above all for reasons of reliability. The limits that one can expect in practice are thus higher than the physical limits. It is not the physical properties of silicon which set the limits for integration techniques, but the processing limits in lithography, insulation, contact diameter and pitch.

How then, is silicon technology expected to develop in the years 1990 to 1995?

Table 3 gives an estimate of what is expected in 1990 to 1995 on the basis of present technological knowledge. For the year 1990 structures of 0.3 μm are expected in the second generation, with chip surfaces between 100 and 200 mm². The difference between the first and the second generation can be seen in the lithographical process. While light-optical processes and steppers are still used for the first generation, the second generation will be based on processes that use X-ray lithography, as structures of 0.3 μm can no longer be realized with light optics. The limit for light-optical processes is put at about 0.5 to 0.6 μm. Below this limit it will be necessary to use new processes such as X-ray lithography. Developments in this direction are already taking place and it is to be expected that they will be available for production in 1990. These are projections for MOS devices. It is also interesting, however, to take a look at the right-hand side of Table 3, which shows the expected development of bipolar devices. We see here that geometries of 0.9 μm in the first generation and 0.5 μm in the second generation are expected. In the first generation this results in limit frequencies for bipolar silicon components of 12 GHz and of 40 GHz in the second generation.

This means that on the basis of bipolar silicon technology with devices of the first generation, it is possible to construct systems with transmission speeds of 2.4 Gbit/s and up to 10 Gbit/s with devices of the second generation. The degree of integration, however, will be considerably lower than with MOS circuits. Silicon elements will, therefore, also be suitable for the construction of very rapid signal processing systems.

If we now return to the question which was posed initially, namely how much longer silicon will be used, and the related question of substitution, we should consider the three points:

1. Substitution in order to save costs;
2. Substitution as a future investment and;
3. Substitution for the purpose of innovation.

Point 1: costs per bit for silicon were reduced by more than a thousand times between 1970 and 1985.

Point 2: Integration techniques, and not the properties of silicon devices, will set the limits.
Table 3
Expectations for 1990-1995

<table>
<thead>
<tr>
<th>Product level</th>
<th>Memory (1st gen.)</th>
<th>Memory (2nd gen.)</th>
<th>MOS logic (1st gen.)</th>
<th>MOS logic (2nd gen.)</th>
<th>Bipolar (1st gen.)</th>
<th>Bipolar (2nd gen.)</th>
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</thead>
<tbody>
<tr>
<td>Width of structure</td>
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<td>0.7 μm</td>
<td>0.3μm</td>
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<tr>
<td>Chip sizes</td>
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<td>&lt;200 mm²</td>
<td>100 mm²</td>
<td>200 mm²</td>
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<td>—</td>
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<tr>
<td>Transistor functions</td>
<td>$6 \times 10^6$</td>
<td>$10^8$</td>
<td>$10^8$</td>
<td>$10^8$</td>
<td>$10^8$</td>
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<tr>
<td>Wiring levels</td>
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<td>&lt;6</td>
<td>&lt;6</td>
<td>&lt;10</td>
<td>&lt;5</td>
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<td>Limit frequency</td>
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<td>12 GHz</td>
<td>40 GHz</td>
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<tr>
<td>Access time</td>
<td>40 ns</td>
<td>&lt;40 ns</td>
<td>—</td>
<td>—</td>
<td>2.4 Gbit/s</td>
<td>10 Gbit/s</td>
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</tbody>
</table>

Point 3: GHz devices have already been developed with silicon. Further possibilities will open up after 1990/95.

**Markets**

Where, then, are the applications and markets for GaAs devices? The applications of GaAs are in those areas where on the one hand increased thermal resistivity, high resistance to radiation, and optical-electrical applications are called for. These are physical properties in which GaAs is clearly superior to silicon or properties which silicon does not possess. This means that the main area of application for GaAs is the military sphere, aviation, and the aerospace industry. In 1984, these applications covered approximately 46 percent of the market share of GaAs, and it is expected that this market share will be extended to 56 percent in 1992. These circuits will not be highly integrated circuits. In other words, they will be MSI circuits rather than VLSI circuits. Alongside this area of application, GaAs is also expected to be used for small-scale integration circuits as interface circuits in the communication industry for large-scale parallel systems, or high-performance analog circuits. These circuits will have medium and small-scale integrated circuits, but not system integration with corresponding VLSI circuits. But then are the figures given at the beginning reliable, namely that the market potential for GaAs in 1992 will be worth about 3.2 billion dollars? These figures are right according to the information available at present. They show that GaAs will only account for a small share of IC consumption, while silicon will continue to dominate and provide the basic material for large-scale integration.

**Conclusion**

To sum up, we can say that GaAs will not become a substitute for silicon in large-scale integration or in system integration. It will be possible to realize some individual functions better in GaAs than in silicon. These functions, however, will have to be very critically examined, as realizing certain functions in a different technology always raises questions concerning interfaces. For example, the position of the interface between silicon and GaAs in systems for optical communications will vary much determine the success or failure of a new system. Not only physical properties will necessarily play a decisive role in this, but also the technological practicability of the whole system. The transition from a technology that is mastered to a new and relatively difficult technology always involves a large number of technical risks. It may thus be a better prospect to use a familiar and perfected technology, even perhaps at the cost of "technical elegance", and only use the other technology where it is absolutely necessary for making a system more efficient and more rapid. The choice of technology or the question of the interface between two technologies will thus be decisive for the success of complex systems. Realizing functions in GaAs which cannot be realized in silicon does not represent substitution. Questions of this kind are of course excluded from any consideration of the question of substitution.

Philips Report No. 10.051E
(from a speech by Dr Peter Draheim, Valvo, Philips GmbH, Federal Germany).

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**ELECTRONIC AND MAGNETIC QUANTITIES.**

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<th>Quantity</th>
<th>Symbol</th>
<th>SI Unit</th>
<th>ABBR</th>
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<td>C</td>
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<td>siemens/metre</td>
<td>S/m</td>
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<td>Ω</td>
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<td>S</td>
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<tr>
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</table>

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elektron india November 1986 11-43
DARK-ROOM EXPOSURE METER

No photographer can work properly in his dark-room without some sort of light meter. The instrument proposed here is not expensive, easy to build, and, apart from the exposure time, it indicates the contrast in relative light values.

In spite of its simplicity, the meter is accurate enough for virtually all requirements. Moreover, it is constructed from standard components throughout, with the possible exception of the Type BPW21 photodiode. Operation of the meter is simplicity itself: a push button for normal exposure measurement; another push button for contrast measurement; and a microammeter for the readout.

Circuit description

The first notable aspect of the circuit diagram in Fig. 1 is that three different levels of supply voltage are required: +2 V, +5 V, and +9 V. At first sight this may seem extravagant, but it is not really as will be seen later. Moreover, the three levels are obtained relatively easily. The +9 V is provided direct by the battery; since the total current consumption does not amount to more than 15 mA, a standard PP3 will do nicely. The +5 V supply is derived from the battery via a Type 7805 voltage regulator, while the +2 V supply is provided by a voltage divider (R19:R20) and an opamp (IC3). The exposure meter is based on a well-known principle: the photovoltaic effect. This effect causes certain semiconductor diodes to produce a forward voltage when they are illuminated. This voltage changes in direct proportion to the logarithm of the causative change in light flux, provided the diode is terminated in a high impedance. This proviso is met in the present circuit by terminating the photodiode, D1, into opamp IC1.

It should be noted that the spectral sensitivity of the BPW21 is very similar to that of the human eye. The maximum sensitivity of the diode and the human eye are about the same, but the BPW21 has a somewhat larger bandwidth. The diode voltage is amplified and inverted in three opamps: IC5; IC6; and IC7, and then applied to the sense combination of the meter, M1, resistor R18, and preset P1. In this application, the meter should have a logarithmic scale (see Fig. 2).

It should be noted that this exposure meter works in exactly opposite way from that in a camera, because the present meter should not indicate the amount of light, but the required period of illumination. Therefore, when the light flux is large, the diode voltage is high, and the voltage across the meter is low. Conversely, if there is but little light, the meter will deflect strongly. Diodes D2 and D3 serve to compensate for the variation of the diode voltage with temperature. In the prototype the variation resulted in a difference of only half a stop for every 7°C, a perfectly satisfactory value, more so when it is remembered that the temperature in a darkroom must be kept fairly constant. As long as the three diodes are not heated unnecessarily when the instrument is handled, all will be well. Potentiometer P1 affords compensation for different paper sensitivities.
because, in conjunction with R5, R6 and R7, it can add a small direct voltage to the measured voltage. Since the meter scale is logarithmic, this added voltage manifests itself as a multiplication of the indicated time. The effect of P1 is the same as that of P2, but this control is only set during the initial calibration of the instrument.

Contrast measurement is effected with the aid of electronic switches ES1, ES2, and ES3. When the contrast push button S1 is open, ES1 will also be open, while ES2 and ES3 will be closed (situation as shown in Fig. 1). The circuit operates as an exposure meter as described. In this state a light section of a negative should be measured.

When S1 is pressed, ES2 and ES3 open and ES1 will close for a short time (at the instant—after ES1 has opened—that C4 is charged via R10 and R11, junction C4-R11 will go high, which causes ES2 to close; once C4 is charged, junction C4-R11 will go low, and this causes ES3 to open again). During the time that ES1 is closed, C1 is charged to the potential then present at the output of IC1. The voltage across the microammeter then drops to zero, so that the pointer does not deflect at all. Even when ES2 opens again after a short while, the potential across C1 is maintained.

With S1 still depressed, hold the photodiode under a dark part of the negative: the meter will deflect again, but the voltage across C1 is now deducted from the measured value. In other words, the meter now indicates the contrast (in LV) between the first and second measurements, i.e. between the light and dark parts of the negative.

Since a difference of one LV (light value) corresponds to a doubling (or halving) of the light flux, the contrast scale of the meter is calibrated linearly as shown in Fig. 2.

**Construction**

The circuit is best constructed on a small piece of single-sided Veroboard. As far as the enclosure is concerned, any small one will do, as long as the board, microammeter, and operating controls can be fitted neatly. The controls should, of course, be easy to reach and operate. The photodiode should be mounted in a manner which ensures that the light from the enlarger reaches it freely. Diodes D1 and D2 should be placed as close as possible to the photodiode, so as to keep temperature differences between the three as small as possible.

**Setting up**

Set P1 to the centre of its travel. Using photographic paper of average sensitivity, make a test strip that is correctly exposed with an exposure time of 2 seconds. The lowest stop number should be used, and the correct illumination obtained by adjusting the height of the enlarger. Place the exposure meter on the base of the enlarger and disperse the light, for instance, by holding a piece of opaque paper in front of the lens. Adjust P2 until the microammeter indicates 2 LV.

Select the fourth lowest f-stop and adjust P3 until the microammeter reads 32 s (= contrast of 4 LV).

**Finally**

A calibrated scale needs to be made for P1, corresponding to the sensitivities of different types of photographic paper. This requires the making of a lot of test strips, but such a scale will be found very useful in practice for a long time to come.

For contrast measurements, the position of P1 is irrelevant (as long as it is not changed between the two measurements).
HIGH-POWER AF AMPLIFIER — 2

After last month's discussion of the power output boards and associated power supply, this concluding article details the design and construction of the bridge/stereo preamplifier, driver stage power supply, soft turn-on and protective circuitry, as well as an effective fan control section. In addition, a variety of illustrative material is offered as an aid to understanding the amplifier's mechanical construction.

Driver stage power supply

The circuit diagram of the combined +90 V and +12 V supply unit originally given in June issue of Elektor India unfortunately contains a small error and a correct version of it is, therefore, reproduced here. The +90 V driver stage supply is quite straightforward; it is capable of supplying up to 2 x 100 mA. The open-circuit voltage of this section is about ±100 V. The ±12 V supply is so conventional as to make any description of it unnecessary.

Preamplifier

In order that the amplifier can be switched from 2 x 250/500 W stereo to 1000 W mono operation in a bridge-connected configuration, the two power output boards must be driven with the normal left and right signals (stereo) or complementary phase signals (bridge set-up); both configurations are supported by the special preamplifier shown in Fig. 8. It can be seen that the antiphase signals come from opamps A5 and A6; the former functions as a non-inverting, the latter as an inverting amplifier stage. In order to ensure simultaneous clipping levels and
identical overall response for both power output boards, all resistors in the circuit must be 1% tolerance, high stability metal film types. The preamplifier inputs are balanced, so that the amplifier can be driven direct from low-noise, balanced cables, as is customary in large PA systems. However, where unbalanced inputs are preferred, the negative input terminals may simply be grounded.

Stereo potentiometer P1 should be a high-quality type, since any tracking error may readily lead to differences in output power from the amplifier's left and right channel. In some cases, a linear stereo potentiometer may, therefore, be preferable over a logarithmic type.

**Soft turn-on and protective circuitry**

The power-on delay circuit, shown in Fig. 9, has its own DC supply, which operates off the AC2 voltage from Tr1. When the mains switch is closed, C1 is charged rapidly, and C2 provides an approximately one second long, high logic level at the inputs of N1, whose inverted output level disables Tr9. Therefore, Re1 will not be energized until one second or so after power-on, and the initial turn-on current for Tr1 and Tr3 is forced to flow through "brake" resistor R50, whose function has already been explained in last month's article on the high-power AF amplifier — refer to Fig. 5 and the section on power supplies in that article. After the initial second has elapsed, Re1 is short circuited by the Re3 contact, and Tr9 and Tr3 receive the full mains voltage across the primary windings. The red LED goes out, and the yellow one lights to indicate the second interval of 1.5 seconds before C2 discharges. N3 supplies a low and N4 a high logic level, and Re3 is energized, connecting the loudspeakers to the amplifier outputs: at which moment the yellow LED goes out and the green one lights, indicating that the amplifier is fully operational. The outlined power-on timing sequence prevents blown fuses as well as loudspeaker destruction.

When the amplifier is switched off, the voltage across C2 is the first to break down, since this smoothing capacitor is rated at only 100 μF. The resulting low logic level at the second input of N3 causes Tr2 to deactivate Re2, disconnecting the loudspeakers from the amplifier outputs. At the same time, Re1 is deactivated, since the base current for Te1...
Fig. 9. Combined circuit diagram of the output DC monitor and the sequentially controlled relay drivers for loudspeakers and ±75 V supply.

has dropped. All LEDs go out, but the large capacitance of the ±75 V supply ensures the prolonged presence of a slowly falling supply voltage for the amplifier output stages, so that the loudspeakers are protected against clicks at power-off. The relays used in this design should be capable of switching and sustaining high contact currents; both $R_{es}$ and $R_{es}$ should be rated at least 10 A in this respect, and suitable types may be found in the parts list with this article.

The proposed circuit with $T_1$ and $T_2$ keeps tabs on the presence of any dangerously high direct voltage levels at the amplifier outputs. Should anything be amiss in this respect, the input of $N_3$ is pulled logic low and $R_{es}$ is consequently deactivated. The green LED goes out and the yellow one lights in order to signal the fault condition.

**Fan control**

A powerful fan capable of helping to keep the heat sink temperature within reasonable limits is an indispensable item in case the amplifier is to output continuous high power levels, such as may be required in applications involving multi- loudspeaker set-ups to cater for considerable sound pressure levels (SPL) at large sites.

Depending on the task assigned to the amplifier, the fan control circuit shown in Fig. 10 may be constructed single or two times over (one or two fans, as required). If a single fan is used, both $T_5$ and $T_6$ are required as temperature sensor devices, because it is preferable to monitor both heat sinks simultaneously. Do not forget to fit these sensors with insulating washers, bushes and heat conducting paste.

Two fans require two control circuits, and $T_2$ can be dispensed with in each of these; a single sensor $T_6$ suffices. Operation of the fan control circuit is straightforward; the Type BD139 transistors function as heat-sink-mounted temperature sensors. Given an ambient temperature of 20 °C and a collector current as defined with $R_{es}$, the base-emitter voltage ($U_{be}$) of the Type BD139 is typically about 625 mV. The fact that $U_{be}$ is temperature dependent to the extent of falling 2 mV per degree of temperature rise is put to good use in the present fan control circuit. The base of the sensor transistor is arranged to be at a voltage between 350 and 640 mV set by $P_z$. Assuming that $P_z$ has been set to a base voltage of 565 mV, then the sensor transistor will not conduct until its $U_{be}$ has
dropped to 565 mV, in other words, when the transistor junction temperature has risen to 50 °C. At this point, Tn is driven and the fan is switched on via the Res contact. This condition is signalled by LED Dn, which should have a suitable colour so as to be readily spotted from a distance. Resistor Rx effects positive feedback which ensures a hysteresis of about 5 °C, necessary to keep the relay from chattering as Tn approaches the preset value.

A common PCB

All of the discussed circuits are fitted on a single PCB as shown in Fig. 11, with the exception of transformers Tr, Tr, Tr, the relays, Ts and Ts (the latter as required in the specific fan arrangement). Where this is desirable, the ready-made PCB for this project may be cut into sections for mounting in suitable locations in the amplifier cabinet.

Fitting the parts should not present problems; the constructor need merely decide on the number of fans and associated control circuits; either one, two, or none.

Amplifier construction and wiring

The unit configuration shown in Fig. 12 may be studied and copied in case the amplifier is to operate without fans. It can be seen that the large heat sinks form the sides of the cabinet, the location ensures sufficient cooling for relatively low power use of the amplifier. However, heavy duty applications require the use of a special combination of heat sink and fan as illustrated in Fig. 13. Note the cooling fins at the inside of the tube-like construction, at one side of which the fan has been mounted to provide a continuous stream of fresh air passing along the inside surface of the tunnel.

All supply, relay and loudspeaker wiring in the amplifier should be made with heavy-duty stranded wire having a cross-sectional area of 2.5 mm² or more.

It would seem advisable to start the wiring job with the connection of the low power transformers Tr, Tr, Tr; refer to the relevant diagram and note the use of the separate switch section Sn for Trz. Now connect the driver supply board and verify the presence of about ±100 V (opencircuit voltage) and ±12 V. Discharge the smoothing capacitors via 10 k
Parts list
Preampifier (Fig. B.)*

Resistors:
R22...R25 = 47 k; 1%*
R3, R26 = 100 Ω
R1 = 10 k potentiometer;
logarithmic*

Capacitors:
C1, C11 = 1 μF MKT
C12, C13 = 470 n
C4, C8 = 22 n
C5, C6 = 100 n

Semiconductor:
IC1 = TL074

Miscellaneous:
S1 = single-pole toggle switch
Relevant section on
PCB Type 80567
Soldering pins as required
2 off 2-way
Cannon/XLR plugs
* see text

Parts list
DC monitor and power-on delay (Fig. 9)

Resistors:
R1, R2 = 15 k
R3...R6 = 56 k
R7, R8 = 10 k
R9, R10 = 34 k
R11, R12 = 1 k
R13 = 2M2
R14 = 3M3
R15 = 220 Ω; 0.5 W
R16 = 22 k
R17, R18 = 39 k
R19 = 56 k
R20...R22 = 390 Ω

Capacitors:
C1, C4 = 33 μF 63 V;
electrolytic
C2, C5 = 22 μF 16 V;
electrolytic
C7 = 100 μF 35 V;
electrolytic
C6, C8 = 560 n

Semiconductors:
D1...D4 = 1N4148
D5, D6 = 1N4001
D7 = zener diode
8V2; 1.3 W
D8 = LED; red
D9 = LED; yellow
D10 = LED; green
T1...T3 = BC547B
T4...T5 = BC557B
T6, T7 = BD679
IC1 = 4033

Miscellaneous:
R1 = high power relay;
coil voltage 12 V;
single contact rated at
10 A 240 V
(e.g. Schrack Type
RL200012; base Type
RN7576*).
Resistor:
R36 = 3k5
R37 = 120 Ω
R38, R39 = 3k3
R40 = 580 Ω
R41 = 68 k
P2 = 100 Ω; preset

Semiconductors:
D21 = LED
D34 = N4148
T6, T9 = BD139
T10 = BD80

Miscellaneous:
R53 = 12 V relay, contact rating 240 V AC; 1 A.
Axial transistor as required (e.g. 8E5 Type V125075-1A15-02; STC Electronic Services, telephone (0279) 26777)
Soldering pins as required.

Fig. 11: Component mounting plan for the combination of support circuits to the power output boards.
1 W resistors, before proceeding with the connection of the power-on delay and protective circuitry. Note the ground connection on the power-on circuit; it should be run direct to the centre tap of Tr1 to prevent the 10 ms charge pulse for C7 from causing hum on the supply lines to the preamplifier.

The power-on and protective circuits may now be tested: switch on the mains and verify the delayed action of Res and Res in that order. Applying a direct voltage, e.g. the + or -12 V supply rail, to either one of the protection DC sense inputs L or R should immediately deactivate loudspeaker relay Res; the ground terminals for delay and DC sense circuits should be temporarily connected for this test. When wiring the L and R inputs to the loudspeaker lines, remember to make the connections direct to the amplifier outputs, that is, not behind the Res contacts!

The construction is next proceeded with the wiring around the toroidal power transformers Tr1 and Tr6, taking due care not to confuse the X, Y and Z points. Brake resistor Res should be mounted on a set of soldering tags. Also remember to fit all mains wiring in an absolutely safe.
and sound manner, so as to prevent possible lethal contact with any of the live points or wires.

A common earth point should be created at the ground terminal of the \( \pm 75 \text{ V} \) power supply, and this point should serve as the common earth return terminal for the loudspeaker wires in the stereo set-up, as well as the point from which the earth return (or) wire to the amplifier boards is run. Since this wire does not carry high currents, it need not be as thick as those for the \( \pm 75 \text{ V} \) rails. The
ground terminals on the power amplifier boards, marked ⬤, are next wired to the corresponding terminal on the driver supply board. Since this point is also the ground terminal for the preamplifier board, the signal wires between drivers and preamplifier should have their screens connected at the preamplifier end only.

The preamplifier ground input connections should be isolated with respect to the amplifier enclosure, and it is best to purchase two Type XLR connectors for this purpose (see parts list).

The amplifier's metal enclosure is connected directly to the mains earth line, as well as to the central ground terminal on the +75 V supply, using a 100 Ω resistor.

Finally, the construction of the high-power AF amplifier is illustrated with a number of photographs in this article, offering suggestions regarding possible enclosure construction and wiring methods (note the purpose-welded framework to hold the PCBs and heavy parts). Keep in mind that a sound mechanical construction is paramount to reliability and the ability to resist the kind of rough treatment an amplifier of this type is likely to be forced to endure.

Testing

After the amplifier has been completed, it is time to check its correct operation. In case you have been patient enough not to test the power output boards as yet, start off with replacing the 6.3 A fuses with 22 Ω 1 W resistors, and turn the quiescent current presets (P1) fully anticlockwise. After switching on the amplifier, no voltage should be measured across these fuse substitutes. Should the loudspeaker relay remain off after the power-on delay, there is bound to be something amiss in the MOSFET output stage. Now check all supply voltages in the amplifier before proceeding with setting the quiescent current to 400 mA per 75 V supply rail (i.e. 0.4 V across each 'fuse'; 190 mA per transistor). The voltage levels given in circuit diagram Fig. 3 may now be checked, at the same time pay due attention to equal current distribution among the power MOSFETs, i.e. each group of parallel-connected source resistors should drop about 25 mV. Large differences in this respect may cause some of the transistors to provide all the power to the load, while others are idle.

Leave the amplifier switched on for some time to verify its thermal stability under quiescent current conditions. The remainder of the test procedure includes checking the output power capability and distortion-free amplification within the frequency bands and signal level range given in the feature list of article (See Elektor India June 86).

There is one final point to make concerning the loudspeaker polarity in the stereo set-up (Fig. 14), note the reversed polarity of the loudspeaker at the R output; this is the result of the 180° phase turn occurring in the inverting opamp in the preamplifier. However, if the wiring to the loudspeaker output sockets is made as shown in Fig. 14, this oddity need not concern the user once the amplifier is fully operational.

Finally, the bridge configuration requires the amplifier to be driven monaurally at the left-hand channel input.
When a multimeter is used for measurement, there can be two types of errors: genuine measuring errors, and errors which are not really measuring errors. Even the specified technical data for electronic components allows for deviations as high as 10%. The carbon film resistors show tolerance values of ±5% or ±10%. Capacitors with ±10% tolerance are considered to be very good. Transistors are tested after manufacture and classified according to their current amplification factor. This classification is designated by a letter or a number appearing after the type number of the transistor. (i.e., BC147B). In spite of this classification, the values of the current amplification factor within a group deviate from each other by more than ±30%. In case of components like electrolytic capacitors, the specified values may also change with time.

Nevertheless, most of the circuits function correctly in spite of considerable component tolerances. Where accuracy is required, suitable compensations must be provided. Whenever we find that a measured value deviates from a specified value, we must first see if this is due to component tolerance. We must also consider whether the deviation has an effect on the functioning of the particular circuit. It is not possible to make any generalised statement about the effect of such deviations on the circuit performance, because the deviations vary from component to component and the effect is different for different circuits.

Let us take an example to see how a genuine measuring error can occur irrespective of the actual measuring accuracy of the measuring instrument.

Figure 1: shows a voltage divider made of two 100KΩ resistors and a 4.5V battery connected across the

**Figure 1:**
A potential divider made of two equal resistors of 100KΩ each. The expected voltage across R2 is 2.25

**Figure 2:**
When a multimeter is connected to measure the voltage across R2, its internal resistance Ri is effectively connected in parallel with R2 and thus affects the equivalent resistance across which the voltage gets measured.
combination. Theoretically, these two resistors must divide the battery voltage of 4.5V into 2.25 + 2.25. Now let us connect a multimeter across R2 and measure the voltage across R2. Surprisingly it is only 1.8V. The cause of this measuring error is the current drawn by the multimeter itself. This current is the result of the internal resistance of the multimeter. This value can be calculated from the Ohms per Volt specification of the multimeter. The internal resistance is obtained by multiplying the ohms per Volt value by the measuring range in volts. Thus a multimeter with 20KΩ per Volt being used on 10 V range will give the internal resistance Ri as follows:

Ri = 10V × 20KΩ/V = 200 KΩ

As this is effectively connected in parallel with the 100KΩ resistance in our above example, the equivalent resistance becomes 67KΩ. The potential divider thus becomes a combination of 100KΩ + 67KΩ. The voltage across R2 thus becomes 1.8 instead of 2.25. The voltage being measured is really 1.8V as shown by the multimeter. The voltage across R2 changes due to the presence of multimeter and the measured value is falsified.

To get over this difficulty, the meter can be connected as shown in figure 3. The voltage across R2 is not measured directly, but compared with another voltage across the potentiometer P1. The difference in voltages will be shown by the multimeter, and will become zero when the two voltages are equal. The meter does not draw any current in this condition as voltages on both the terminals are equal. The potentiometer can have a directly calibrated dial to read the voltage, or we can now measure the voltage across the sliding contact of P1 independently. If the total value of P1 is kept low enough, the internal resistance of the multimeter will not affect the reading.

Another way to get rid of this problem is to use a high impedance input circuit with the multimeter. The schematic diagram of such a circuit is shown in figure 4. The impedance converter contains an amplifier which requires a very low input current and gives a high output current which is proportional to the input current. An emitter follower circuit can be used in case of ordinary transistors, and a source follower circuit can be used in case of FETs. The input voltage is not amplified by the amplifier and thus the measured voltage is indicated accurately by the multimeter without drawing input current from the voltage under test.

Effectively, the Ri of the multimeter can be said to have become very high. In case of an emitter follower, the theoretical value can be estimated as follows. The effective Ri is the product of the current amplification factor of the transistor and the parallel combination of Re and Ri of the multimeter. As RE is much smaller than Ri, we can have Ri (effective) = RE × Current Gain. Assuming that RE = 4.7 KΩ and current gain = 250, the effective input impedance Ri = 1.18 MΩ. Thus the multimeter can now be said to have an input resistance of about 1 MΩ which is sufficiently high for most measuring applications.

Figure 3: No current can flow through the meter when voltages on both terminals become equal to each other. The meter indicates zero at this point. As the meter does not load the test circuit, the voltage values are not affected by internal resistance of the meter.

Figure 4: The high impedance attachment for a multimeter using an emitter follower amplifier. The amplifier draws negligible current and does not load test circuits.
Measuring Power With A Multimeter

The ability to convert a specific quantity of energy in a specific period into another form of energy is called power. Speaking in electrical terms we can say:

\[ p(t) = u(t) \cdot i(t) \]

where \( p(t) \) is electrical power, \( u(t) \) is voltage and \( i(t) \) is current. In case of DC currents and voltages the relation becomes:

\[ P = U \cdot I \]

Capital letters are used for DC quantities which are not a function of time, that is, they do not change with time. Lower case letters are used for Alternating quantities, and to show their dependence on time they are written as \( p(t) \), \( u(t) \), \( i(t) \) etc.

Let us first look at the DC quantities. Figure 1 shows \( U \), \( I \) and \( P \) as steady levels (DC quantities). The current \( I \) flows through a resistance \( R \) and the voltage across that resistance is \( U \). All these values do not change with time and hence at any given time the following relation holds good:

\[ P = U \cdot I \]

Now for example let \( U = 24V \), \( I = 2A \) then we have the result as:

\[ P = 24V \times 2A = 48W \]

Here the \( W \) stands for Watts, which is the unit of the so called DC power. You must have noticed by now that measuring DC power with a multimeter is very simple! Just measure the voltage, then measure the current and then multiply them to get the DC power.

Figure 1:
The values of DC voltage \( a) \) and DC current \( b) \) are constant at every instant. Hence the product \( c) \) is also constant at every instant. The shaded area in \( c) \) represents the energy which is converted into heat in the resistance \( R \)
Now let us consider the other possibility. The voltage and current can be alternating values, as shown in figure 2. We can come across such type of quantities in case of an output stage of an amplifier. The output stage produces an alternating voltage and current. Can we use the multimeter to measure the power even in this case? Though the answer is Yes, it is not as simple as in the case of the DC quantities.

Let us have another look at figure 2. Here an alternating current i flows through a resistance R and the voltage across the resistance is U, which is also an alternating voltage. Naturally so, because the resistance has a fixed value of R ohms. The nature of waveforms for U and i are both sine waves. These are known as sinusoidal waveforms. They are variable with respect to time and said to be functions of time. If we take the value of u and i at any given instant of time and multiply them together, we can get the power at that instant of time. This process can be carried out at every point on the time axis and the resulting power, which will also be a function of time can be plotted as in figure 2 C. The power can also be expressed as an average over a period T. For this purpose the average value converter can be used as shown in figure 3. Using such converter, it is possible to obtain average voltage and current and then multiply to get average power. However, this does not give the active power or the effective power. Also, please note that what we mean by average current and voltage is not the true average, because average current and voltage of a sinusoidal alternating waveform would be zero! The average we are talking about is after the full wave rectification by the bridge rectifier.

Why the active power or the effective power alone is a true measure for the power can be explained with help of the set up shown in figure 4. Here we see the output stage of a Hi-Fi amplifier connected to a loudspeaker and a headphone, along with a multimeter. Now consider an alternating current flowing through the speech coil of the loudspeaker. Not only the membrane of the loudspeaker moves to and fro, the alternating current produces heat in the speech coil of the loudspeaker, because it also has an

Figure 2:
If a sinusoidal voltage (a) is applied to a resistance, then a sinusoidal current (b) is generated through the resistor. The power calculated at consecutive points can be plotted as in (c).

Figure 3:
The average value converter consists of a full wave rectifier bridge connected to a moving coil meter. The moving coil meter effectively averages out the input voltage.

Figure 4:
The measuring set up for measuring power with a multimeter. A practical set up is shown in the photograph in the beginning of this chapter.
Omnic resistance. The same quantity of heat could also be produced by a direct current. This particular value of direct current is called the effective current. The effective current is greater than the average value by 11%. For example, the mains supply voltage is 230V (effective value), where as the average value is only about 207V. If we connect the multimeter across the mains outlet, it reads 230V, because it is designed to read the effective value in the AC ranges. The scale of the multimeter is calibrated in such a manner that it directly reads the effective value for a sinusoidal alternating input. This is well suited for our requirement of power measurement.

The effective values of current and voltages are also called RMS values. Without going into the details, we can just note that RMS stands for Root-Mean-Square. This notation comes from the fact that for a sinusoidal alternating waveform the effective value is the square root of the mean of the squares.

Using the effective (or RMS) values of voltages and currents, the same formula that is used for DC quantities becomes valid once again.

\[ P = U.I \]

This can be further simplified by using Ohm's law

\[ I = \frac{U}{R} \]

Thus the power equation becomes

\[ P = \frac{U^2}{R} \]

From this relation, the power measurement becomes still more easier, because we need only one measurement - that of the voltage across the loudspeaker. Resistance (impedance) of the loudspeaker is specified on the loudspeaker as either 8Ω, 4Ω or 2Ω.

All we have to do is measure the voltage, square it and divide it by the loudspeaker impedance. For example, if we read 4.5V across the loudspeaker, and if the loudspeaker has 8Ω impedance then the effective power is 2.53W. These measurements are carried out at a standard input frequency of 1KHz as can be seen from figure 4. The measurement however will depend on the amplifier setting.

What is of real interest is the non-distorted power output. To decide this, it is better to believe in one's own ears. A headphone can be connected as shown in figure 4, to check for exact setting of the amplifier volume control where distortion just sets in. The voltage can be measured at this setting and then from the loudspeaker impedance, the non-distorted power output can be calculated.

The 1KHz sinewave generator can be constructed as shown in figure 5 and 6. Component list is also provided for the circuit. For the same output voltage the effective power output depends on the loudspeaker impedance.

This can be confirmed by setting the sinewave generator and the amplifier for an output voltage of 4.5V and then changing the loudspeakers from 8Ω to 4Ω and then to 2Ω. The 8Ω loudspeaker has about 2.6W, the 4Ω loudspeaker has about 5W and 2Ω loudspeaker has about 10W. If the amplifier is not rated for 10W output, distortion will set in with a 2Ω loudspeaker.

Component List:
- R1 to R3 = 39KΩ
- R4 = 1KΩ
- R5 = 2.7KΩ
- P1 - 2.5KΩ Trim Pot.
- C1 to C3 = 10 nF
- C4 = 100 μF / 10V
- T1, T2 - BC 547 B
- 1 Standard SELEX PCB
- 1 S-V Battery Pack.
The most important formula used in power calculations is:

\[ P = U I \]

which means that electrical power is the product of current and voltage. Another thing that becomes clear from this relation is that voltage or current alone is not sufficient to deliver any power output. A very common example of this is the crackling noise we sometimes hear while taking off a synthetic pullover. This noise is generated by the minute sparks generated by the static electricity. The voltages involved can be as high as 10KV. However, these sparks do not harm us as the currents produced are negligible.

The power is measured in Watts, and a Watt is defined as follows:

\[ 1 \text{W} = 1 \text{V} \times 1 \text{A} \]

If any two of the three quantities in the power equation are known, the third can be calculated.

Let us take an example. A bicycle dynamo produces up to 3W at 6V. So the current produced by the dynamo is

\[ I = \frac{P}{U} = \frac{3}{6} = 0.5 \text{A} \]

Half an Ampere is not a very high current but it serves the purpose of lighting 3W bulbs at 6V! Here the voltage and current are both small and produce a small power output. However, if we take the same current (0.5A) from our mains supply of 230V, the power output produced will be

\[ (230\text{V}) \times (0.5\text{A}) = 115\text{W} \]

which is a substantial value. The difference is due to the higher voltage. From this we can clearly see that voltage and current both play an equally important role in producing power.

Let us consider a practical situation. A 100W bulb connected to mains supply of 230V. Its current can be calculated as:

\[ I = \frac{P}{U} = \frac{100}{230} = 0.43\text{A} \]

And using the Ohm's law for calculating the resistance, we have

\[ R = \frac{V}{I} = \frac{230\text{V}}{0.45 \text{A}} = 535 \Omega \]

This must be the resistance of the bulb. Surprisingly, a measurement of the bulb resistance with a multimeter gives a very low reading; about 30 to 400. What went wrong? Our calculations, or the multimeter? Both of them are correct, and the

difference in two values can be explained by the fact that what we measured with a multimeter was the resistance of the cold element and what we calculated was the resistance of the hot element when bulb is glowing. When the bulb glows, there is a strong movement of electrons inside the glowing wire and the effective resistance increases. Unfortunately the mains voltage remains same even when the cold bulb is switched on across the mains supply. This gives rise to a very high initial current given by

\[ I = \frac{U}{R'} = \frac{230\text{V}}{30\Omega} = 7.6\text{A} \]

This initial current flowing into the bulb lights the bulb and the element is instantly heated up. The resistance then increases to about 535\Omega as seen before. The initial power drawn by the bulb is enormous:

\[ P = (230\text{V}) \times (7.6\text{A}) = 1748\text{W} = 1.75\text{KW} \]

**Figure 1:**
The bicycle dynamo produces 3W electrical power at 6V output. This gives a current output of 0.5A. **Figure 2:**
Although the 100W bulb requires less current than that produced by a bicycle dynamo, its power is much higher because the mains supply voltage is much higher than the dynamo voltage. **Figure 3:**
When a 100W bulb is switched on, it takes an instantaneous power of about 1.75 KW. For an old bulb this may prove to be fatal.
Fortunately, this power is just the initial instantaneous power and drops down immediately to 100 W as the bulb glows.

Let us take another example to see how the supply voltage affects the power output to the same device. Consider a 1000W Hair Dryer connected to 230V mains supply. The current drawn is 4.3A.

The consumption of the fan motor is negligible compared to that of the heater. Now if we connect the same device to a supply voltage of 110V, will the power also reduce to about half the value? No, the power drops to one fourth the original value. As the voltage becomes half the current also becomes half and thus their product becomes one fourth. The Hair Dryer now operates only at 250W.

Let us now turn to some very low power devices. The antenna required by a radio is such a device. The antenna intercepts the radio waves and produces a tiny voltage of about 10 uV. This voltage drives an equally tiny current through the antenna cable and the receiver. If we assume the resistance to be of about 50Ω, then the current is:

\[
\frac{10 \mu V}{50 \Omega} = 0.2 \mu A
\]

and the power delivered to the receiver is:

\[
(10 \mu V) \cdot (0.2 \mu A) = 2 \text{ pW}
\]

That is two picowatt or two billionth of a Watt!

Figure 4: A 1000 W Hair Dryer works only at 250 W when the input voltage becomes half.

Figure 5: Although radio transmitters operate at very high power values, what reaches the reception antenna is just a few billionths of a Watt.

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CORRECTIONS

In various car theft alarms published in Elektor Electronics over the past year, the alarm had the facility of incapacitating the ignition system when the alarm is set. It has been found that this can lead to damage to electronic ignition systems, and in cars fitted with such a system, it is therefore better to use the relay that incapacitates the ignition system to break the supply to the starter motor (relays).

VHF Preamplifier
(May 1986 p 32)
The value of capacitor C8 should be 1n.

Car burglar alarm
(Aug 1 Sept p 67)
As drawn, the voltage across relay R8 cannot drop to zero: it is, therefore, better to connect the emitter of T1 to the +8 V line via an 1kΩ resistor instead of to the collector of T2.

Indoor unit for satellite TV reception — 1
(In This Issue)
1. Owing to a processing error at the printer's, the lines between C8 and M1 pin 1, and that between R8 and the collector of T2, have short, yet incorrect gaps. As the T2 base resistor is badly blurred, this is R8, 10 kΩ.
2. In the component overlay, Fig. 6, the resistor identified R8 in the IQ section should be R5.
3. Please add to Fig. 5d: C = isolating bush.

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